

26 May 2025

Genesis eyes further growth in production and cashflow with acquisition of Laverton Gold Project

Delivers Genesis ~4Moz Resource near its Laverton mill

HIGHLIGHTS

- Genesis has entered into a binding Share Purchase Agreement to acquire the Laverton Gold Project for upfront cash consideration of A\$250 million from Focus Minerals Limited (ASX: FML)
- ► The Laverton Gold Project has a global Mineral Resource of ~4Moz at 1.7g/t¹; Consideration equates to ~A\$63 per Resource ounce; Reserves stand at 546koz at 1.3g/t²
- Substantial scope for Resource growth, with a large tenement package comprising highly prospective gold tenure
- Clear potential for the Laverton Gold Project to supply open pit and underground ore to Genesis' operating 3Mtpa Laverton mill approximately 30km away
- Acquisition will be **funded from existing cash and an upsized corporate revolver finance facility** (currently undrawn)
- ▶ Genesis retains significant balance sheet flexibility post-completion with ~A\$350 million in available liquidity³
- The acquisition is consistent with Genesis' "ASPIRE 400" accelerated growth strategy; It also provides the opportunity to unlock significant synergies, including the optimum pairing of deposits with processing infrastructure at Genesis' Leonora and Laverton operations
- ▶ Completion expected to occur in early June 2025
 - Genesis' **immediate priorities** at the acquired Laverton assets include:
 - In-fill and extensional drilling to de-risk and rebuild the Resource
 - Studies Optimisation of multiple oxide / transitional pits to feed into Laverton mine plan
 - Approvals 99% of Resources and Reserves on granted Mining Leases
 - ▶ Staged mill expansion studies at Laverton (in addition to studies at Leonora)
 - Exploration over a large, highly prospective tenement package
- In light of Genesis' strong ongoing growth, **Genesis has bolstered its Board** with the **appointment of highly experienced mining executive and mining engineer Duncan Coutts as an Executive Director**, further strengthening the Board's capability in the core areas of project development and growth
- Mr Coutts' extensive experience includes due diligence, feasibility studies, design, approvals and ultimately project development and integration in respect to multiple "bolt-on" acquisitions at Ramelius Resources⁴. This specialist skill set will be invaluable to Genesis as part of the Laverton Gold Project acquisition and Genesis' broader "ASPIRE 400" growth strategy
- Mr Coutts' appointment will also enable the Company's Chief Operating Officer Matt Nixon to continue focusing on his pivotal role in delivering the 5-year strategic plan and driving Genesis' operations, which has seen his team deliver exceptional results; This also reflects the Company's strategic philosophy that its success stems from delivery and outperformance at the operational level

The Global Mineral Resource is inclusive of a historical JORC 2004 estimate of 4.8Mt at 1.6g/t equating to 240koz contained gold reported by Focus. The Competent Person has not completed sufficient work to classify the historic estimate as mineral resources in accordance with JORC 2012. It is uncertain, following evaluation and/or further exploration work that the historical estimate can be reported as mineral resources in accordance with JORC 2012.

Refer to Appendices C and D for JORC information in relation to Mineral Resources and Ore Reserves of Laverton Gold Project

²Refer to Appendices C and D for JORC information in relation to Mineral Resources and Ore Reserves of Laverton Gold Project

³Genesis has A\$372 million in cash and equivalents as at 30th April, and A\$225 million in undrawn corporate revolver facilities (totalling ~A\$597 million in available funding)

⁴Refer to Appendix A for additional background information

- Mick Wilkes will retire as a Non-Executive Director but will be retained as a Technical Advisor to support "ASPIRE 400"
- Further senior appointments of leading industry specialists to support growth include (refer to Appendix A for profiles):
 - ▶ Eugenio Gatto (Group Manager Processing) Project Manager staged mill expansion studies
 - Neil Sutcliffe (Project Manager Rail and Logistics) Initially focused on Tower Hill project logistics
 - Dan Schwann (Metallurgist) Staged mill expansion studies
 - Neuplan Multi-disciplinary mining project management consultancy delivering end-to-end capital project solutions

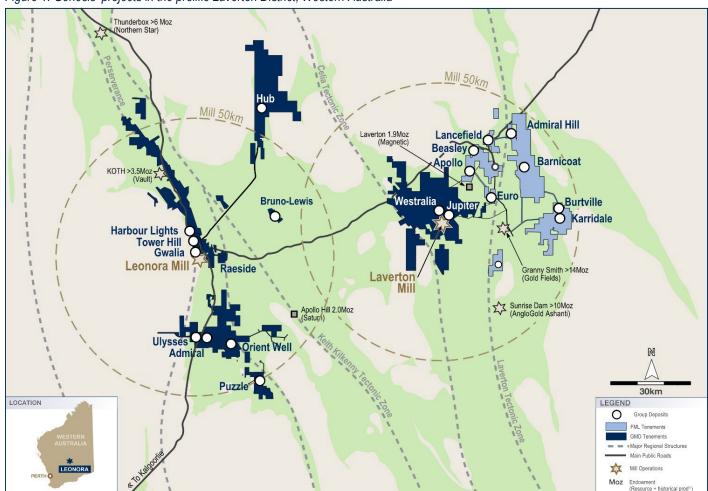
Strategic acquisition

Genesis Minerals Limited (ASX: GMD) (**Genesis** or the **Company**) advises that it has entered into a binding Share Purchase Agreement to acquire the Laverton Gold Project in Western Australia's Laverton District from Focus Minerals Limited (ASX: FML) (**Focus**).

Genesis considers this to be a **highly strategic acquisition**, **consolidating the Laverton assets of Focus and Genesis**, which include the recently restarted 3Mtpa Laverton mill.

Situated approximately 30km from Genesis' Laverton mill, the acquisition of the Laverton Gold Project is expected to unlock significant synergies, including the optimum pairing of deposits and regional processing infrastructure.

Figure 1. Genesis' projects in the prolific Laverton District, Western Australia



The Laverton Gold Project comprises a global Mineral Resources of 73Mt @ 1.7g/t for 3,900koz (Refer Appendices C and D), contained across a series of open pit deposits and one underground deposit.

The Global Mineral Resource is inclusive of historical JORC 2004 estimate of 4.8Mt at 1.6g/t equating to 240koz contained gold reported by Focus. The Competent Person has not done sufficient work to classify the historic estimate as mineral resources in accordance with JORC 2012. It is uncertain that following evaluation and/or further exploration work that the historical estimate will be able to be reported as mineral resources in accordance with JORC 2012. Nothing has come to the attention of Genesis that causes it to question the accuracy or reliability of the historical estimate. However, Genesis has not independently validated the historical estimate and therefore it is not to be regarded as reporting, adopting or endorsing that estimate

In addition to the large Mineral Resource, the Laverton Gold Project includes 455km² of prospective gold tenure which offers substantial exploration upside both in-mine and regionally⁵

As part of the acquisition, Genesis will also take ownership of the Laverton Gold Project's site infrastructure (workshops, haul roads, bore field etc). The Laverton Gold Project has historically produced approximately 3.6Moz⁶.

Under the agreement, Genesis has agreed to pay Focus consideration of A\$250 million in cash on completion.

Various Third-party royalties range from 1.0-5.0% across the acquired package.

There are no conditions precedent to completion which is expected to occur in early June 2025.

The purchase price will be funded via Genesis' existing cash and undrawn corporate revolver facility which has been upsized from \$120 million to \$225 million, with Genesis having total available funding of ~A\$597 million⁷. Following completion, Genesis will retain significant balance sheet flexibility with ~A\$350 million in available funding.

Genesis Managing Director Raleigh Finlayson said the Laverton Gold Project was a highly strategic and opportunistic acquisition for the Company:

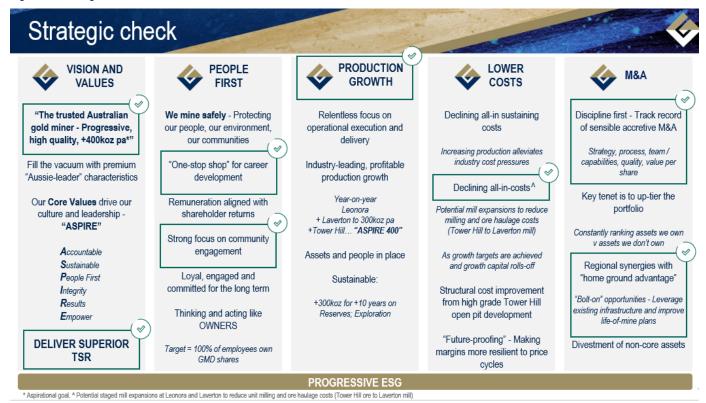
"This is the perfect bolt-on acquisition," Mr Finlayson said. "It delivers a substantial 4Moz Resource with immense exploration upside right next to our Laverton mill.

"It offers supplementary open pit and underground ore to our Laverton mill and in the process gives us flexibility regarding the most efficient pairing of deposits and processing infrastructure between Laverton and Leonora.

"With more ore available at Laverton, our flagship Tower Hill deposit can potentially be processed at Leonora resulting in significantly lower operating costs. With both the Laverton and Leonora mills now 'long ore', studies into staged plant expansions continue apace.

"These benefits make the transaction entirely consistent with our 'ASPIRE 400' accelerated growth strategy".

Figure 2. Strategic check



Genesis intends to provide further details on the transaction in an ASX investor presentation (post-completion).

Canaccord Genuity (Australia) Limited and Sternship Advisers acted as corporate advisors and Gilbert + Tobin acted as legal adviser to Genesis.

Mineral Resources and Ore Reserves for the Genesis Group are extracted from the GMD ASX announcement 8th April 2025 "Reserves rise to 3.7Moz, underpinning ASPIRE 400 strategy. Genesis confirms that it is not aware of any new information or data that materially affects the information included in that announcement and, in relation to the estimates of Mineral Resources and Ore Reserves in that announcement, confirms that all material assumptions and technical parameters underpinning the estimates in that announcement continue to apply and have not materially changed. See also Appendix C for details of the Mineral Resource estimate for the Laverton Gold Project.

⁶ Refer to Focus Minerals Limited ASX Announcement 27 May ²019 "25% Increase in Karridale Gold Deposit's Mineral Resource" page 2, 28th October 2019 "Resource Upgrade for Telegraph Open Pit Deposit" page 3, 30th January 2020 "Outstanding Results at Beasley Creek South" page 13, 5 May 2022 "Upgrade for Euro deposits build Laverton Gold Project Resource Base" page 5,8th March 2024 "Laverton Gold Project Mineral Resource Updates" page 4 and Focus Minerals Limited ASX Announcement 18th January 2022 "Lancefield Far North Maiden Mineral Resource" page 3

⁷Genesis has A\$372 million in cash and equivalents as at 30th April, and A\$225 million in undrawn corporate revolver facilities (totalling ~A\$597 million in available funding)

Board appointment

Genesis also advises that Mick Wilkes will retire as a Non-Executive Director of the Company and will be retained as a Technical Advisor. Further to this, Genesis is pleased to announce that it has appointed highly experienced resources executive and mining engineer Duncan Coutts as an Executive Director. The material terms of Mr Coutts' Employment Agreement with the Company are included in Appendix B.

Genesis Chairman Tony Kiernan thanked Mr Wilkes for his immense contribution to Genesis' successful project acquisition and development strategy.

"Mick has played a key role in the development and execution of Genesis' growth strategy. His experience has helped establish the Company's exceptional asset base and strong outlook we now have.

On behalf of the Board, I would like to thank him for his counsel and guidance, and we look forward to his ongoing contribution as a technical advisor".

Mr Kiernan said Mr Coutts' vast experience would further bolster the Company's development and operational capability.

"We are committed to building a world-class team of highly experienced specialists to help ensure we maximise the opportunities presented by our growing asset base.

"This approach is particularly important as we expand the assets and implement our strategy to increase production and cashflow.

"Duncan's immense experience in project due diligence, project development and integration will be invaluable as part of our commitment to growth and development.

"Importantly, his appointment will also enable Chief Operating Officer Matt Nixon to remain heavily focused on our core operations and five-year plan. The huge success of Matt and his team has been pivotal to Genesis' strong results and rapid growth, and this structure is aimed at ensuring we continue to deliver on this front".

Corporate structure

Ordinary shares on issue 1,130m Unquoted securities 39m

Market Capitalisation (23rd May 2025) A\$4.9b (share price of \$4.34)

Cash and equivalents (30th April 2025) A\$372m Undrawn Corporate Revolver (pre-completion) A\$225m

Substantial shareholders Australian Super 17.5%

State Street Corporation 6.9%

Van Eck Associates Corporation 6.8% Paradice Investment Management 5.9%

Vanguard Group 5.0%

This announcement is approved for release by Raleigh Finlayson, Managing Director, Genesis.

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Some statements in this announcement regarding estimates or future events are forward-looking statements. They include indications of, and guidance on, future matters. Forward-looking statements include, but are not limited to, statements preceded by words such as "planned", "expected", "projected", "estimated", "may", "scheduled", "intends", "anticipates", "believes", "potential", "could", "nominal", "conceptual" and similar expressions. Forward-looking statements, opinions and estimates included in this announcement are based on assumptions and contingencies which are subject to change without notice, as are statements about market and industry trends, which are based on interpretations of current market conditions.

Forward-looking statements are provided as a general guide only and should not be relied on as a guarantee of future performance. Forward-looking statements may be affected by a range of variables that could cause actual results to differ from estimated results and may cause or Genesis' actual performance and financial results in future periods to materially differ from any projections of future performance or results expressed or implied by such forward-looking statements. These risks and uncertainties include but are not limited to liabilities inherent in mine development and production, geological, mining and processing technical problems, the inability to obtain any additional mine licenses, permits and other regulatory approvals required in connection with mining and third party processing operations, competition for among other things, capital, acquisition of reserves, undeveloped lands and skilled personnel, incorrect assessments of the value of acquisitions, changes in commodity prices and exchange rate, currency and interest fluctuations, various events which could disrupt operations and/or the transportation of mineral products, including labour stoppages and severe weather conditions, the demand for and availability of transportation services, the ability to secure adequate financing and management's ability to anticipate and manage the foregoing factors and risks. These and other factors should be considered carefully, and readers should not place undue reliance on such forward-looking information. There can be no assurance that forward-looking statements will prove to be correct.

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APPENDIX A - PROFILES

Duncan Coutts

Duncan Coutts is a qualified mining engineer with more than 30 years resource industry experience.

Mr Coutts was previously Chief Operating Officer at Ramelius Resources (ASX: RMS) (Ramelius), where he oversaw management of Ramelius' operating mines. During his time at Ramelius, Mr Coutts was heavily involved in due diligence and the acquisition of a number of projects, which he then managed through to integration and ultimately production.

Prior to joining Ramelius, he held a combination of consulting roles and senior management and executive level positions for both large scale and junior mining companies, including Kimberley Metals Group, Galaxy Resources, Metals X and Harmony Gold Australia.

He holds a Bachelor of Engineering (Hons) in Mining Engineering from the Western Australian School of Mines in Kalgoorlie.

Eugenio Gatto

Eugenio Gatto is a senior Minerals Processing professional with over 30 years' experience across gold, copper, and uranium operations in Australia and internationally. He has successfully led multiple plant improvement initiatives, operational expansions, and major project studies throughout his career.

Mr Gatto is currently the Group Manager of Processing and has held leadership roles with Northern Star Resources, Saracen Minerals, and Kalgoorlie Consolidated Gold Mines, and was a Lead Process Engineer at Ausenco, where he worked on a range of project development expansion studies and operational improvement projects. Notable achievements include the delivery of the KCGM Emissions Reduction Project, Carosue Dam Plant Expansion and was a technical lead on the Thunderbox and KCGM plant expansions scoping and prefeasibility studies.

Mr Gatto holds a Bachelor of Engineering in Minerals Processing from the University of Queensland.

Neil Sutcliffe

Neil Sutcliffe is a highly accomplished logistics and operations executive with over 20 years of leadership experience managing and delivering large scale logistics projects, most recently as the General Manager Bulk West at Aurizon.

Mr Sutcliffe brings a unique blend of aviation, rail, trucking and mining sector logistics expertise delivering complex, multi-modal, efficient, safe, and scalable solutions for some of Australia's largest resource companies including BHP, Glencore and Lynas.

Mr Sutcliffe is a results-driven project leader whose career is marked by his ability to drive strategic growth, execute large-scale integration projects who is known for his strategic leadership and stakeholder engagement capabilities. His leadership has resulted in significant safety improvements, contract expansions, and innovative logistics solutions that have enhanced customer value and operational resilience.

Mr Sutcliffe holds an MBA from the University of Western Australia, a Bachelor of Science, and multiple certifications in safety, auditing, and corporate governance.

Dan Schwann

Dan Schwann is a metallurgical and processing consultant specialising in mining and metallurgical projects.

Mr Schwann is currently Managing Director of Daniel Schwann Consulting Pty Ltd, established in 2008. During 2015 to 2022 he was Group Manager - Processing for Evolution Mining. Prior to establishing the consulting business, Mr Schwann worked in various operational processing and metallurgy management roles for 12 years.

He brings a strong background in processing, metallurgy and fixed plant maintenance across operations in Australia, Canada, Asia and Africa. This includes studies, design, construction and commissioning of significant expansions in processing capacity at Evolution Mining's Cowal and Mungari gold mines and Pan Aust's Phu Kham copper gold mine.

Mr Schwann is a Fellow of the AusIMM with +25 years of experience and has tertiary qualifications in Mineral Science - Extractive Metallurgy (Murdoch University), Mineral Economics (WA School of Mines).

Neuplan

Neuplan Pty Ltd is a multi-disciplinary capital project management consultancy, established in 2013. With a team of over 40 professionals including engineers, lawyers, quantity surveyors, project controls specialists, and construction managers, Neuplan delivers tailored project management solutions across the full lifecycle of capital projects.

Their expertise spans project engineering and delivery, commercial and legal management, procurement, contract administration, estimating, and project controls. Neuplan provides either complete project solutions or targeted specialist support to meet the unique demands of each project.

With a proven track record across a diverse range of commodities, Neuplan's recent engagements include extensive involvement in the A\$1.5 billion KCGM Growth project, as part of the Integrated Owner's Team managing the process plant expansion, and as Project Management Consultant for the non-process infrastructure, and new tailings storage facilities.

APPENDIX B

Summary of Material Terms of Mr Coutts' Employment Agreement

In accordance with ASX Listing Rule 3.16.4, the material terms of Mr Coutts Employment Agreement with the Company are as follows:

Effective Date of Appointment:	Appointed as Executive Director from 26 May 2025.
Duration of Employment:	Employment continues until terminated in accordance with the Employment Agreement.
Fixed Remuneration:	Base Salary is \$595,000 exclusive of superannuation. Mr Coutts base salary is inclusive of directors' fees.
Performance-based remuneration (Short Term Incentive)	Subject to and in accordance with the GMD STI Plan rules, STI targets are: • FY2025: 50% of TFR at target • FY2026: 75% of TFR at target Incentive arrangements may be varied or withdrawn at the absolute discretion of the Company in accordance with the rules of any incentive scheme adopted by the Company from time to time.
Equity-based remuneration (Long Term Incentive)	Subject to and in accordance with the GMD Equity Incentive Plan rules, LTI target is: • 100% of TFR For FY2025, Mr Coutts has been granted 361,000 Performance Rights to be issued within 7 days of commencement of employment. FY2025 LTI performance period is 1/7/2024 to 30/6/2027. Vesting conditions are subject to the FY2025 LTI KPIs as disclosed in the Company's FY24 Annual Report, and in addition, remaining: 1. employed as an Executive Director of the Company at 30/6/2026, and/or 2. engaged as a Director of the Company (in an executive or non-executive capacity) and/or consulting to the Company, at 30/6/2027. Incentive arrangements may be varied or withdrawn at the absolute discretion of the Company in accordance with the rules of any incentive scheme adopted by the Company from time to time.
Notice period, termination and termination payments	Termination by Notice: Employee notice period – 3 months Company notice period – 3 months Termination Without Notice: Company may terminate without notice in circumstances including serious misconduct or breach of material terms. Right to Terminate for Material Downgrade: Mr Coutts may terminate if the Company seeks to materially downgrade employment conditions. Severance Payment: On the occurrence of certain events, Mr Coutts is entitled to a severance payment for past services rendered equal to 6 months base salary. If required, the severance payment will be reduced in accordance with the formula specified in section 200G of the Corporations Act (which formula takes into account any amount payable to the Employee in lieu of notice, where notice periods are not worked) and subject to ASX Listing Rule 10.19.

APPENDIX C - RESOURCES AND ORE RESERVES

Mineral Resources

Table 1. Laverton Gold Project Detailed Mineral Resources*

			Measured			Indicated			Inferred			Total	
Deposit	JORC Category	Tonnes (000s)	Grade (g/t Au)	Ounces (000s)	Tonnes (000s)	Grade (g/t Au)	Ounces (000s)	Tonnes (000s)	Grade (g/t Au)	Ounces (000s)	Tonnes (000s)	Grade (g/t Au)	Ounces (000s)
Barnicoat Project		· · · · ·	, <u> </u>			, <u> </u>			· · · · · · · · · · · · · · · · · · ·				
Admiral Hill	JORC 2004	-	-	-	660	1.4	30	1,300	1.1	46	2,000	1.2	76
Barnicoat	JORC 2004	-	-	-	340	1.3	14	250	1.0	8	590	1.2	22
Bells	JORC 2004	-	-	-	590	2.0	38	36	1.4	2	630	2.0	40
Castaway	JORC 2004	-	-	_	250	1.6	13	28	1.8	2	280	1.6	15
Grouse	JORC 2004	_	_	_	450	1.7	24	27	1.3	1	470	1.7	25
Sickle	JORC 2004	390	1.7	21	200	2.6	16	150	3.1	15	740	2.2	52
Total Barnicoat	JORC 2004	390	1.7	21	2,500	1.7	140	1.800	1.3	74	4,700	1.5	230
Karridale - Burtville Project	JONE 2004	390	1.7	21	2,300	1.7	140	1,000	1.3	/4	4,700	1.0	230
Burtville	JORC 2012	-			5,100	1.0	160	1.600	0.9	47	6,600	1.0	210
Karridale	JORC 2012			_	22,000	1.4	970	5,600	1.2	220	28,000	1.3	1,200
Total Karridale - Burtville	JORC 2012	-	_	_	27,000	1.3	1,100	7,100	1.2	270	34,000	1.3	1,400
Central Laverton Project	JORC 2012	-		-	21,000	1.3	1,100	7,100	1.2	210	34,000	1.3	1,400
Euro South	JORC 2012				520	1.4	24	50	1.2	2	570	1.4	26
Euro North	JORC 2012 JORC 2012	-	-	-	560	2.1	38	270	2.1	18	830	2.1	56
		-	-	-									
Total Central Laverton Project	JORC 2012	•	•		1,100	1.8	62	320	1.9	20	1,400	1.8	82
Craigiemore - Mary Mac Trend	1000 0040							020	4.4	40	000	4.4	40
Golden Pinnacles	JORC 2012	-	-	-	-	-	-	230	1.4	10	230	1.4	10
Mary Mac Hill and North	JORC 2012	-	-	-	410	1.3	17	140	1.1	5	550	1.2	22
Mary Mac South	JORC 2012	-	-	-	990	1.3	42	380	1.6	19	1,400	1.4	61
Craigiemore	JORC 2012	-	-	-	1,100	1.5	54	210	1.1	7	1,300	1.4	61
Total Craigiemore - Mary Mac	JORC 2012	-	•	-	2,500	1.4	110	960	1.3	41	3,500	1.4	150
West Laverton - Bulldog Trend													
West Laverton and Rega	JORC 2012	-	-	-	1,100	1.8	65	1,800	1.5	90	2,900	1.6	150
Bulldog	JORC 2012	-	-	-	-	-	-	670	1.4	30	670	1.4	30
Total West Laverton - Bulldog	JORC 2012	-		-	1,100	1.8	65	2,500	1.5	120	3,600	1.6	190
Chatterbox Trend													
Apollo (Whisper)	JORC 2012	-	-	-	3,700	1.6	190	140	1.1	5	3,900	1.6	200
Eclipse (Garden Well)	JORC 2012	-	-	-	200	1.7	11	99	1.0	3	290	1.4	14
Innuendo	JORC 2012	-	-	-	300	1.4	14	740	1.0	23	1,000	1.1	37
Rumor	JORC 2012	-	-	-	-	-	-	2,600	1.4	120	2,600	1.4	120
Total Chatterbox	JORC 2012	-		-	4,200	1.6	220	3,500	1.3	150	7,800	1.5	370
Gladiator Trend													
Gladiator West	JORC 2012	-	-	-	470	0.8	12	670	0.8	18	1,100	0.8	30
Gladiator and Murrays	JORC 2012	-	-	-	140	1.1	5	740	1.2	28	880	1.1	33
Total Gladiator	JORC 2012	-			610	0.9	17	1,400	1.0	45	2,000	1.0	63
Chatterbox Project								1,100					
Beasley Creek	JORC 2012	-	-	-	3,700	2.0	240	390	1.6	21	4,100	2.0	260
Beasley Creek South	JORC 2012	-	-	_	1.600	2.1	110	430	0.8	11	2,100	1.8	120
Total Chatterbox	JORC 2012				5,300	2.1	350	820	1.2	32	6,200	1.9	380
Lancefield - Wedge Project	CONC LUIL				0,000	2.1	000	020		ŲŽ.	0,200	1.0	000
Telegraph	JORC 2012	-			640	2.1	44	530	1.4	25	1,200	1.8	68
Wedge - Lancefield North	JORC 2012		_		2,700	1.7	140	750	1.1	27	3,400	1.5	170
Lancefield Far North	JORC 2012 JORC 2012		-		2,700	1.7	1+0	790	1.1	34	790	1.3	34
South Lancefield	JORC 2012 JORC 2004				72	4.0	9	3	5.0	1	790 75	4.1	10
			-	-									
Total Lancefield - Wedge	JORC 2012	-	-	-	3,400	1.8	190	2,100	1.3	87	5,400	1.6	280
Laverton Underground	IODC 2012						_	3.900	6.3	790	3.900	6.3	790
Lancefield UG	JORC 2012	-			-			-,			-,		
Total Laverton Underground	JORC 2012	-	4.7	-	40.000	4.5	-	3,900	6.3	790	3,900	6.3	790
Total Laverton Surface		390	1.7	21	48,000	1.5	2,300	21,000	1.3	840	69,000	1.4	3,100
Grand Total Underground Grand Total		390						3,900	6.3	790	3,900	6.3	790
			1.7	21	48.000	1.5	2.300	25.000	2.1	1.600	73.000	1.7	3.900

^{*}The Global Mineral Resource is inclusive of historical JORC 2004 estimate of 4.8Mt at 1.6g/t equating to 240koz contained gold reported by Focus. The competent person has not done sufficient work to classify the historic estimate as mineral resources in accordance with JORC 2012. It is uncertain that following evaluation and/or further exploration work that the historical estimate will be able to be reported as mineral resources in accordance with JORC 2012. Nothing has come to the attention of Genesis that causes it to question the accuracy or reliability of the historical estimate. However, Genesis has not independently validated the historical estimate and therefore it is not to be regarded as reporting, adopting or endorsing that estimate

Geology and Geological Interpretation

The Laverton Gold Project lies within the Laverton Greenstone Belt of the Eastern Yilgarn Craton, specifically within the Kurnalpi Terrane. Gold mineralisation is structurally controlled and hosted in a range of lithologies, including andesitic volcanics, mafic to ultramafic sequences, banded iron formations (BIF), and felsic intrusives.

Key structural controls include the Chatterbox Shear Zone, the Laverton Shear Zone, and various subordinate NE- and NNW-trending faults. Mineralisation is associated with brittle–ductile deformation, folding, quartz veining, silica-sericite alteration, and sulphide (mainly pyrite) mineralisation.

The majority of wireframe interpretations for lithology, mineralisation, and structure were created using Leapfrog Geo. Some older models used Surpac for wireframing and estimation. Interpretations relied on logged lithology, alteration, veining intensity, and assay data, and were adjusted based on known structural trends and geophysical inputs. A nominal 0.5 g/t Au cut-off was used to guide domain interpretations, with a minimum downhole width of 1m (RC) or 0.2m (DDH).

Drilling and Sampling

Sampling and sub-sampling techniques across the project were consistent with industry standards. Reverse Circulation (RC) drill samples were collected at one-metre intervals using riffle or cone splitters, producing sub-samples typically weighing between 2 and 4 kilograms. Early RC programs by Crescent Gold used riffle splitters, while later campaigns utilised onboard cone splitters. Composite drill samples, usually four metres in length, were spear-sampled and submitted for initial analysis; in instances where gold grades exceeded 0.1 g/t Au, one-metre re-splits were then submitted for re-assay. Wet RC drill samples were logged separately and flagged for exclusion from estimation if they were considered to compromise data quality. Diamond drill core was cut in half using a core saw, with sampling guided by geological boundaries and mineralisation zones. Sample lengths varied from 0.1 to 1.3 metres, depending on lithology and structural context. Core recovery, RQD, and sample integrity were routinely recorded, with most programs achieving recoveries in excess of 90 percent.

Drilling techniques across the various programs included RC drilling with face-sampling hammers and diamond drilling, predominantly in NQ and HQ diameters. Downhole surveys were undertaken using north-seeking gyroscopic tools or single-shot magnetic instruments. Drill collars were surveyed with differential GPS (DGPS), and collar positions were cross-verified against surveyed topographic surfaces and historical mine plans to ensure positional accuracy. Multiple companies conducted the drilling over several decades, but all programs adopted practices aligned with accepted industry norms.

Sample analysis was conducted at multiple certified laboratories, including ALS, SGS, Amdel, Genalysis, Ultratrace, and Jinning. Gold assays were primarily undertaken using fire assay with either a 40-gram or 50-gram charge, and results were reported using either Atomic Absorption Spectroscopy (AAS) or Inductively Coupled Plasma Optical Emission Spectrometry (ICP-OES) finish. Composite samples were typically assayed initially, and when gold results exceeded 0.1 g/t Au, one-metre re-splits were submitted for confirmation. In selected diamond core intervals, screen fire assays and multi-element analyses were conducted to assess the presence of coarse gold or associated geochemical pathfinders. Extensive QAQC procedures were implemented, including the insertion of certified reference materials (CRMs), field duplicates, and blanks, as well as laboratory repeats and umpire checks. QAQC results were reviewed routinely and were found to be within acceptable tolerance limits.

Estimation Methodology

Estimation methodology involved geological domaining based on lithology, structure, and mineralisation controls, with interpretation conducted in Leapfrog Geo. A small number of older Mineral Resource estimates were completed using Surpac software. One-metre composite samples were generated within mineralised domains, and top cuts were applied where necessary based on statistical outlier analysis. Grade estimation was generally completed using Ordinary Kriging (OK) in Datamine, with variograms constructed in Snowden Supervisor for each domain. The block model design incorporated appropriate parent and sub-block dimensions to reflect the drill spacing and geometry of the mineralisation. Search ellipses were oriented parallel to the dominant structural and mineralisation trends within each deposit and were varied according to data density and classification criteria.

The cut-off grade used for reporting mineral resources, typically ranging between 0.5 and 1.0g/t Au, was selected based on preliminary assumptions around mining method, metallurgical recovery, and site-specific cost structures.

Mineral resource classification for Karridale - Burtville Project, Central Laverton Project, Craigiemore - Mary Mac Trend, West Laverton - Bulldog Trend, Chatterbox Trend, Gladiator Trend and Chatterbox Project was conducted in accordance with the 2012 Edition of the Australasian Code for Reporting of Exploration Results, Mineral Resources and Ore Reserves (JORC 2012) guidelines and was based on drill spacing, geological continuity, assay confidence, and QAQC outcomes. This represents 93% and 94% of tonnes and ounces respectively reported in the Mineral Resource estimate.

Table 2. Laverton Gold Project Detailed JORC 2012 Mineral Resources

Deposit	JORC Category	Measured Tonnes (000s)	Grade (g/t Au)	Ounces (000s)	Indicated Tonnes (000s)	Grade (g/t Au)	Ounces (000s)	Inferred Tonnes (000s)	Grade (g/t Au)	Ounces (000s)	Total Tonnes (000s)	Grade (g/t Au)	Ounces (000s)
Karridale - Burtville Project	J	(3333)	(3,)	(3333)	(3333)	(3, 1 1 11)	(3333)	(3333)	(3,)	(3333)	(3333)	(3, 5 : 12)	(3333)
Burtville	JORC 2012	-	-	-	5,100	1.0	160	1,600	0.9	47	6,600	1.0	210
Karridale	JORC 2012	-	-	-	22,000	1.4	970	5,600	1.2	220	28,000	1.3	1,200
Total Karridale - Burtville		-	-	-	27,000	1.3	1,100	7,100	1.2	270	34,000	1.3	1,400
Surface Deposits													
Euro South	JORC 2012	-	-	-	520	1.4	24	50	1.2	2	570	1.4	26
Euro North	JORC 2012	-	-	-	560	2.1	38	270	2.1	18	830	2.1	56
Central Laverton Project Craigiemore - Mary Mac Trend		-		•	1,100	1.8	62	320	1.9	20	1,400	1.8	82
Golden Pinnacles	JORC 2012	-	-	-	-	-	-	230	1.4	10	230	1.4	10
Mary Mac Hill and North	JORC 2012	-	-	-	410	1.3	17	140	1.1	5	550	1.2	22
Mary Mac South	JORC 2012	-	-	-	990	1.3	42	380	1.6	19	1,400	1.4	61
Craigiemore	JORC 2012	-	-	-	1,100	1.5	54	210	1.1	7	1,300	1.4	61
Total Craigiemore - Mary Mac		-	-		2,500	1.4	110	960	1.3	41	3,500	1.4	150
West Laverton - Bulldog Trend													
West Laverton and Rega	JORC 2012	-	-	-	1,100	1.8	65	1,800	1.5	90	2,900	1.6	150
Bulldog	JORC 2012	-	-	-	-	-	-	670	1.4	30	670	1.4	30
Total West Laverton - Bulldog		-	-	-	1,100	1.8	65	2,500	1.5	120	3,600	1.6	190
Chatterbox Trend													
Apollo (Whisper)	JORC 2012	-	-	-	3,700	1.6	190	140	1.1	5	3,900	1.6	200
Eclipse (Garden Well)	JORC 2012	-	-	-	200	1.7	11	99	1.0	3	290	1.4	14
Innuendo	JORC 2012	-	-	-	300	1.4	14	740	1.0	23	1,000	1.1	37
Rumor	JORC 2012	-	-	-	-	-	-	2,600	1.4	120	2,600	1.4	120
Total Chatterbox		-	-	-	4,200	1.6	220	3,500	1.3	150	7,800	1.5	370
Gladiator Trend													
Gladiator West	JORC 2012	-	-	-	470	8.0	12	670	0.8	18	1,100	0.8	30
Gladiator and Murrays	JORC 2012	-	-	-	140	1.1	5	740	1.2	28	880	1.1	33
Total Gladiator		-	-	-	610	0.9	17	1,400	1.0	45	2,000	1.0	63
Chatterbox Project													
Beasley Creek	JORC 2012	-	-	-	3,700	2.0	240	390	1.6	21	4,100	2.0	260
Beasley Creek South	JORC 2012	-	-	-	1,600	2.1	110	430	8.0	11	2,100	1.8	120
Total Chatterbox Lancefield - Wedge Project					5,300	2.1	350	820	1.2	32	6,200	1.9	380
Telegraph	JORC 2012	-	-	-	640	2.1	44	530	1.4	25	1,200	1.8	68
Wedge - Lancefield North	JORC 2012	-	-	-	2,700	1.7	140	750	1.1	27	3,400	1.5	170
Lancefield Far North	JORC 2012	-	-	-	-	-	-	790	1.3	34	790	1.3	34
Total Lancefield - Wedge		-	-	-	3,300	1.7	180	2,100	1.3	86	5,400	1.6	270
Laverton Underground													
Lancefield UG	JORC 2012	-	-	-	-	-	-	3,900	6.3	790	3,900	6.3	790
Total Laverton Underground			-	-		-	-	3,900	6.3	790	3,900	6.3	790
Total Laverton Surface					45,000	1.5	2,100	19,000	1.3	760	64,000	1.4	2,900
Grand Total Underground						-		3,900	6.3	790	3,900	6.3	790
Grand Total				-	45,000	1.5	2,100	23,000	2.1	1,600	68,000	1.7	3,700

Inferred resources were defined in areas where drill spacing typically ranged between 40 and 80 metres, with sufficient geological and grade continuity established. Indicated resources were classified where drill spacing was between 20 and 40 metres, and where geological interpretation and sampling demonstrated higher confidence. Measured resources were defined where drill spacing was 20 metres or less and were supported by high-quality data from both RC and diamond drilling, as well as consistent assay and survey data. All classification decisions were underpinned by reviews of sample recovery, geological logging, assay QAQC results, and structural control confidence.

Mining and metallurgical considerations were factored into the classification and estimation processes. The deposits are considered suitable for conventional open pit or selective underground mining methods, depending on depth, geometry, and continuity. Historical mining has taken place at several of the deposits, including Chatterbox Trend deposits, providing practical insights into potential mining scenarios. Metallurgical data, though limited in recent years, suggest that the mineralisation is generally free-milling and amenable to gravity concentration and cyanide leaching, with recoveries historically reported as favourable.

Mineral resource estimates for the Barnicoat Project area as well as South Lancefield have been reported under JORC Code (2004) and are hence considered historic estimates. The estimates are as of 30 June 2011; and were completed by Crescent Gold. These resources account for 7% and 6% of tonnes and ounces as reported in the overall Mineral Resource Estimate and do not underpin development and mining plans⁸. Reliability of the estimate is inferred through extensive work including RC and diamond drilling, Leapfrog/Surpac/Datamine modelling, reinterpretations, SG, variography. These estimates have not been updated to comply with the JORC Code (2012) and are therefore considered historical. A Competent Person has not completed sufficient work to classify these estimates as current Mineral Resources in accordance with the JORC Code (2012), and it is uncertain whether further evaluation will result in the estimates being reported in accordance with the JORC Code (2012). The company is not treating these estimates as current, and further work, including data validation, QAQC review, and re-estimation, will be required to report updated resources.

Refer to Focus Minerals Limited ASX Announcement 8th March 2024 "Laverton Gold Project Mineral Resource Updates" page 4, Focus Minerals Limited ASX Announcement 18th January 2022 "Lancefield Far North Maiden Mineral Resource" page 3

Table 3. Laverton Gold Project Detailed Historic JORC (2004) Mineral Resources

		Measured			Indicated			Inferred			Total		
Deposit	JORC Category	Tonnes (000s)	Grade (g/t Au)	Ounces (000s)									
Barnicoat Project													
Admiral Hill	JORC2004	-	-	-	660	1.4	30	1,300	1.1	46	2,000	1.2	76
Barnicoat	JORC2004	-	-	-	340	1.3	14	250	1.0	8	590	1.2	22
Bells	JORC2004	-	-	-	590	2.0	38	36	1.4	2	630	2.0	40
Castaway	JORC2004	-	-	-	250	1.6	13	28	1.8	2	280	1.6	15
Grouse	JORC2004	-	-	-	450	1.7	24	27	1.3	1	470	1.7	25
Sickle	JORC2004	390	1.7	21	200	2.6	16	150	3.1	15	740	2.2	52
Total Barnicoat		390	1.7	21	2,500	1.7	140	1,800	1.3	74	4,700	1.5	230
Lancefield - Wedge Project													
South Lancefield	JORC 2004	-	-	-	72	4.0	9	3	5.0	1	75	4.1	10
Total Lancefield - Wedge		-	-	-	72	4.0	9	3	5.0	1	75	4.1	10
Total Laverton Surface		390	1.7	21	2,600	1.8	140	1,800	1.3	75	4,800	1.6	240

The tenure is 100% owned by Focus Minerals (Laverton) Pty Ltd, a wholly owned subsidiary of Focus Minerals Limited (Focus), with all relevant mining leases in good standing. Environmental, permitting, and infrastructure considerations have been preliminarily assessed and do not present any known impediments to project development. Royalties applicable to the various tenements are detailed in the Focus 2024 Annual Report released to the ASX on 1 April 2025.

Ore Reserves

The Ore Reserve for the Laverton Gold Project is based exclusively on Indicated Mineral Resources and classified as Probable, with no Measured Resources included.

Table 4. Laverton Gold Project Detailed Ore Reserves

		Proved			Probable			Total	
Deposit	Tonnes	Grade	Ounces	Tonnes	Grade	Ounces	Tonnes	Grade	Ounces
Deposit	Mt	(g/t Au)	(000s)	Mt	(g/t Au)	(000s)	Mt	(g/t Au)	(000s)
Karridale - Burtville Project									
Burtville	-		-	4	0.9	103	4	0.9	103
Karridale	-		-	6	1.1	205	6	1.1	205
Total Karridale - Burtville	-		-	9	1.0	308	9	1.0	308
Chatterbox Project									
Burtville	-		-	2	2.3	133	2	2.3	133
Karridale	-		-	1	2.7	65	1	2.7	65
Total Chatterbox	-		-	3	2.5	198	3	2.5	198
Wedge/Lancefield									
Wedge-Lancefield North	-		-	1	1.6	41	1	1.6	41
Total Wedge/Lancefield	-	-	-	1	1.6	41	1	1.6	41
Total Ore Reserve				13	1.3	546	13	1.3	546

The Ore Reserve is underpinned by a Pre-Feasibility Study (**PFS**) completed to a minimum of ±25% accuracy. A gold price of AUD \$2,207/oz was used in deriving the Ore Reserve. Sensitivity analysis was conducted, with the project remaining economically viable down to AUD \$1,900/oz. All cost inputs (mining, processing, G&A, capital) were sourced from first principles or contractor quotes and benchmarked against comparable operations.

The PFS confirmed positive project economics with a forecast IRR above 25% and a payback period of less than 3 years. While current cost and revenue assumptions support the economic viability of the reported Ore Reserves, additional technical and economic studies are planned to further refine and update these inputs. This will ensure that any future changes in processing arrangements, or operating strategies are appropriately reflected in revised cut-off grade determinations.

Resource Classification and Ore Reserve Confidence

Only Indicated Mineral Resources have been converted to Ore Reserves. Classification reflects geological confidence, data spacing, QAQC performance, and kriging efficiency metrics. All modifying factors applied are derived from PFS-level studies or higher and are considered sufficiently reliable to support Probable classification. No Inferred Resources were used in the estimation or design of the Ore Reserve.

⁹Refer to Focus Minerals Limited's ASX announcement 16th April 2021 "Updated Laverton Stage 1 Open Pit PFS Progressive Results"

Mining Method and Assumptions

The selected mining method is conventional open pit mining using hydraulic excavators and rigid dump trucks. Minimum mining width of 20m and geotechnical berm and wall design parameters have been applied based on site-specific pit slope studies.

Designs were based on optimisations completed using Whittle 4X pit optimisation software, incorporating geotechnical parameters, ramp access, dilution buffers, and minimum mining width constraints. Ore loss and dilution assumptions were derived from regularised model-to-mine shape comparisons and vary by deposit. Average dilution applied ranges from 6% to 20%, and ore loss ranges from 13% to 16%, dependent on orebody geometry and continuity. Mining recovery includes planned grade control drilling and visual ore identification practices. Geotechnical criteria for the design of the open pits were developed for the purpose of the PFS. The resultant overall slope angles, following pit design, are summarised below. Pits were sequenced to maximise early cash flow while optimising equipment usage and plant feed continuity.

Deposit	Hanging Wall (degrees)	Footwall (degrees)		
Karridale	32 to 46	36 to 42		
Burtville	46 to 43	40 to 45		
Beasley Creek	35 to 46	36 to 38		
Beasley Creek South	43	37 to 43		
Wedge	43 to 47	44 to 57		

Cut-off grades were calculated based on ore haulage distance, processing cost, recoveries, and sustaining capital allowances. The applied cut-off grades vary slightly by deposit due to differences in haulage distances and operational assumptions but generally fall within the range of 0.45–0.60g/t Au. It is noted that these cut-off grades are specific to the current development and operating strategy assessed in this Pre-Feasibility Study. Should project processing arrangements change in the future, variations in operating cost structures, haulage strategies, or processing routes may lead to a revision of applicable cut-off grades in line with the updated project assumptions.

Processing Method and Recovery Assumptions

The Ore Reserve is based on the assumption material will be processing via the existing gravity and carbon-in-leach (CIL) circuit at the Barnicoat Mill, which will be refurbished. The metallurgical recovery applied is 91% for all deposits, based on recent testwork and historical reconciliation performance from similar ore types treated at the Barnicoat mill. No material deleterious elements are present in the ore. The mill has a nameplate capacity of 1.5 Mtpa and sufficient tailings and water infrastructure to support the Ore Reserve throughput.

Modifying Factors and Approvals

All material modifying factors have been considered. Mining will occur on granted Mining Leases held 100% by Focus. The land is subject to a registered Native Title claim (Nyalpa Pirniku WC2019/002), and environmental and heritage approvals are in place for the majority of areas, with remaining permits expected in the ordinary course of development.

Key infrastructure including haul roads, workshops, bore fields, and processing facilities are already in place, significantly de-risking the project. Power will be provided via on-site diesel generation, with allowance made for connection to a long-term renewable or grid-supplied solution.

The proximity of Laverton town provides strong access to transport routes and a mining-experienced workforce.

Competent Persons Statement

The information in this announcement that relates to Exploration Results is based on information compiled by Mr Alex Aaltonen, a Competent Person who is a Member of the Australasian Institute of Mining and Metallurgy (AusIMM). Mr Aaltonen is an employee of Focus Minerals Limited. Mr Aaltonen has sufficient experience that is relevant to the style of mineralisation and type of deposit under consideration and to the activity which he is undertaking to qualify as a Competent Person as defined in the 2012 Edition of the Australasian Code for Reporting of Exploration Results, Mineral Resources and Ore Reserves.

The Mineral Resource estimates were compiled by Mr Alex Aaltonen, an employee of Focus Minerals. Mr Alex Aaltonen, who is a Member of the Australasian Institute of Mining and Metallurgy (AuslMM). Mr Aaltonen is an employee of Focus Minerals Limited. Mr Aaltonen has sufficient experience that is relevant to the style of mineralisation and type of deposit under consideration and to the

activity which he is undertaking to qualify as a Competent Person as defined in the 2012 Edition of the Australasian Code for Reporting of Exploration Results, Mineral Resources and Ore Reserves.

The information in this announcement that relates to Ore Reserves is based on information compiled by Mr Alex Aaltonen, who is a Member of the Australasian Institute of Mining and Metallurgy (AusIMM). Mr Aaltonen is an employee of Focus Minerals Limited. Mr Aaltonen has sufficient experience that is relevant to the style of mineralisation and type of deposit under consideration and to the activity which he is undertaking to qualify as a Competent Person as defined in the 2012 Edition of the Australasian Code for Reporting of Exploration Results, Mineral Resources and Ore Reserves.

Mr Aaltonen consents to the inclusion in the report of the matters above based on the information in the form and context in which it appears.

Table 5. Laverton Gold Project Admiral - Drill results >200 gram metres

						End of Hole			Downhole		Gram metres
Hole ID	Easting	Northing	RL	Dip (°)	Azimuth (°)	(m)	From (m)	To (m)	Length (m)	Au (g/t)	(g*m)
BEC380	465344	6817963	454	-60	270	42	24	30	6	1514.2	9085
BEC419	465364	6817843	466	-60	270	42	5	20	15	502.1	7030
SL119	440332	6829848	419	-60	259	78	60	77	17	223.71	3803
BD23	439023	6831091	439	-90	0	70	16	28	12	145.61	1747
CM005	440297	6829588	450	-60	256.7	50	9	37	28	18.55	519
BSR101	454279	6827858	473	-60	270	86	4	28	24	17.18	412
BTRC058	465060	6817696	434	-60	90	75	42	55	13	29.34	381
LFU025-02	439776	6840629	-49	-40	44	134.2	97.35	127.7	30.35	11.99	364
WG039	450567	6831867	478	-59	272	42	16	32	16	21.88	350
SL260	440314	6829682	462	-90	0	20	16	19	3	110.36	331
SYRC079	450365	6834149	469	-55	270	100	18	19	1	330	330
BTRC034	464997	6817696	438	-60	90	60	29	58	29	11.3	328
BDD1	439024	6831091	444	-90	0	34.5	10.85	23.5	12.65	25.05	317
EZ026	451167	6827567	494	-45	254.6	21.2	5.6	20.4	14.8	21.2	314
CRC132	450765	6832225	462	-70	253	60	24	60	36	8.55	308
BTRC099	464951	6817753	432	-60	90	69	42	69	27	11.27	304
GP85	437610	6832921	431	-60	270	57	31	51	20	14.36	287
LFU056-02	439731	6840756	-91	-89	163	155	144.59	149.56	4.97	55.57	276
BCP0224	434092	6838698	356	-60	270	110	75	110	35	7.83	274
LFP0192	439030	6840878	419	-90	0	34	28	34	6	43.03	258
BCP0540	434065	6838698	365	-60	270	54	18	54	36	7.16	258
BTRC041	464870	6817692	415	-60	90	75	67	75	8	31.36	251
CMRC319	440322	6829811	380	-75	104	60	29	59	30	8.35	250
BTRC160	464886	6817714	438	-60	90	50	38	50	12	20.28	243
HPC016	450458	6833333	438	-58.5	274	72	61	72	11	21.49	236
HPC109	450458	6833340	446	-50	270	30	0	24	24	9.8	235
WG038	450564	6831867	480	-56	275	42	12	32	20	11.24	225
GWRC082	433417	6829757	387	-60	270	100	44	78	34	6.58	224
HPD002	450421	6833529	427	-59.3	90.8	100	69	85	16	13.8	221
LFP0363	439431	6842040	412	-90	0	40	36	40	4	54.09	216
299_277	440520	6843494	453	-90	0	5	0	5	5	42.9	214
SL121	440321	6829674	413	-90	0	85	47	85	38	5.55	211
BER068	466588	6817811	452	-60	270	51	25	47	22	9.37	206
SL127	440333	6829880	434	-60	259	95	47	54	7	28.53	200
LFP0361	439432	6841999	413	-90	0	40	34	40	6	33.28	200

APPENDIX D - JORC TABLE 1s

JORC Code, 2012 Edition – Table 1 sections 1 – 3 and Section 4 2021 PFS Reserves for: Burtville, Karridale, Beasley Creek, Beasley Creek South and Wedge – Lancefield North, Follows

For the purpose of assessing and reporting compliance with the JORC (2012) code, Table 1 of the of the JORC code has been compiled and provided below. Further detail regarding the basis of the Ore Reserve estimates can be found in the 2020 PFS Update and the original 2017 PFS study and relevant Mineral Resource reports.

Section 1 Sampling Techniques and Data

(Criteria in this section apply to all succeeding sections.)

Section 1 Details for the Karridale Deposit from ASX Announcement "Karridale Mineral Resource increases by 60%"
 Dated 24/09/2020

Criteria	Commentary
Sampling	RC Sampling
techniques	 RC percussion drill chips were collected through a cone splitter from the drill rig. The bulk sample from drilling was placed in neatly rows on the ground with the nominal 2-3kg calico split sub- sample placed on top of the corresponding sample. RC chips were passed through a cone splitter to achieve a nominal sample weight of approximately 3kg. The splitter was levelled at the beginning of each hole. In the 2018 and 2019 drilling geological logging defined whether a sample was to be submitted as a 1m cone split sample or a 4m spear composite sample. Split samples (1m) were transferred to sample numbered calico bags for submission to the laboratory. Composite samples were spear sampled using a spear to obtain a small representative sample and deposited into numbered sample bags. Previous drill programs from 2017 and earlier have submitted 1m samples for assay taken from the drill rig for the entire hole length with no compositing of samples.
	 Diamond Core Sampling Diamond core was collected into standard plastic core trays. Down hole depths were marked onto wooden core blocks and stored in the trays. The diamond core was marked up for sampling by the supervising geologist during the core logging process, with sample intervals determined by the presence of mineralisation and/or alteration. Whenever possible the cut-line was drawn parallel to and close to the down hole core orientation line to ensure the cut-line was consistent over the hole. The core was cut in half using an automatic core saw, with half-core samples submitted for analysis.
Drilling	RC drilling was conducted using a 5 3/8inch face sampling hammer for RC drilling.
techniques	At hole completion, downhole surveys for RC holes were completed at a 10m interval by using True North Seeking Gyro tool. Otherwise, a single shot Eastman camera downhole survey was used either "in-rod" or "open hole".
	Diamond core was drilled at NQ2/HQ size. All drill core was oriented where competent by the drilling contractor using an Ezy-mark or similar system.
	• At hole completion diamond holes were survey using a single shot tool at a range of intervals between 20m and 50m, averaging 30m.
Drill	RC sample recovery was recorded by a visual estimate during the logging process.
sample recovery	DD sample recovery was measured and calculated (core loss) during the logging process. DD core had generally good to excellent recovery.
Logging	All RC samples were geologically logged to record weathering, regolith, rock type, alteration, mineralisation, structure, texture and any other notable features that are present. All data is entered directly into validating digital software.
	All core samples were oriented where possible, marked at metre intervals and compared to the depth measurements on the core blocks. Any loss of core was noted and recorded in the drilling database.
	All diamond core was logged for structure, geology and geotechnical data using the same system as that for RC.
	Logging was qualitative, however the geologists often recorded quantitative mineral percentage ranges for the sulphide minerals present.
	The logging information was transferred into the company's drilling database once the log was complete.
	Diamond core was photographed one core tray at a time using a standardised photography jig. RC chip trays are routinely photographed.
	The entire length of all holes is geologically logged, except for rock roller diamond pre-collars which produce no sample.
Sub-sampling	All samples were collected in a pre-numbered calico bag bearing a unique sample ID.
techniques and sample	Core samples were taken from half core, cut using an Almonte automatic core saw. The remainder of the core was retained in core trays tagged with a hole number and metre mark.
preparation	 At the assay laboratory, all samples were oven dried, crushed to a nominal 10mm using a jaw crusher (core samples only) and weighed. Samples in excess of 3kg in weight were riffle split to achieve a maximum 3kg sample weight. All samples were pulverized to 90% passing 75µm.
	Gold analysis was by a 30 to 50g Fire Assay with an ICP-OES or AAS Finish.
	Different laboratories have been used over the years. Most recently Jinning Testing & Inspection completed the assay testing, with

Criteria	Commentary
	sample preparation completed in Kalgoorlie or Perth and analysis completed in Perth for the 2018/2019 drilling. Previously drill samples were submitted to Kalgoorlie Assay Laboratories for sample preparation and analysis.
	The assay laboratories' sample preparation procedures follow industry best practice, with techniques and practices that are appropriate for this style of mineralisation. Pulp duplicates were taken at the pulverising stage and selective repeats conducted at the laboratories' discretion.
	QAQC checks involved inserting standards 1:20 samples (with minimum 3 standards every submission). Duplicate samples for RC were achieved by producing 2 samples for each metre one hole every 20th hole drilled and submitting all produced samples. The remaining bulk sample was also bagged to plastic bags for retention and further checks. Diamond core field duplicates were not taken.
	Regular reviews of the sampling were carried out by the supervising geologist and senior field staff, to ensure all procedures were followed and best industry practice carried out.
	The sample sizes were appropriate for the type, style and consistency of mineralisation encountered during this phase of exploration.
Quality of assay data and	The assay method and laboratory procedures were appropriate for this style of mineralisation. The fire assay technique was designed to measure total gold in the sample.
laboratory tests	No geophysical tools, spectrometers or handheld XRF instruments were used for assay determination.
	The QA/QC process described above was sufficient to establish acceptable levels of accuracy and precision. All results from assay standards and duplicates were scrutinised to ensure they fell within acceptable tolerances and where they did not further analysis was conducted as appropriate.
	Umpire samples are collected on a routine basis will be submitted to independent ISO certified labs in 2019.
	Additional bulk mineralised RC samples have also been collected and retained for follow up QAQC, metallurgical and sample characterisation purposes.
Verification of sampling	Significant intervals were visually inspected by company geologists to correlate assay results to logged mineralisation. Consultants were not used for this process.
and assaying	Primary logging data is sent in digital format to the company's Database Administrator (DBA) as often as was practicable. The DBA imports the data into an acQuire database, with assay results merged into the database upon receipt from the laboratory. Once loaded, data was extracted for verification by the geologist in charge of the project.
Location of	Drill collars are surveyed after completion using a DGPS instrument.
data points	A True North Seeking Gyro for RC end of holes surveys or a Reflex single shot camera for diamond drilling was used for "single shot" surveys whilst advancing drilling.
	All coordinates and bearings use the MGA94 Zone 51 grid system.
	FML utilises Landgate sourced regional topographic maps and contours as well as internally produced survey pick-ups produced by the mining survey teams utilising DGPS base station instruments.
	After completion, the drill hole locations were picked up by DGPS with accuracy of +/-20cm.
Data spacing and distribution	Drill spacing at Karridale varies from 40m x 40m to 80m x 80m on the wider fringes of the known deposit.
Orientation of data in relation	Drilling was designed based on known/developing geological models, field mapping, verified historical data, cross-sectional and long-sectional interpretation.
to geological structure	Where achievable, drill holes were oriented at right angles to strike of deposit, with dip optimised for drill capabilities and the dip of the ore body.
	True widths have not been calculated for reported intersections. However, drill orientation was wherever possible consistently optimised to approximate true width of mineralisation.
Sample	All samples were reconciled against the sample submission with any omissions or variations reported to FML.
security	All samples were bagged in a tied numbered calico bag. The bags were placed into green plastic bags with a sample submission sheet secured by cable ties and delivered directly from site to the Kalgoorlie laboratories by FML personnel at completion of each hole.

• Section 1 Details for the Burtville deposit from ASX Announcement "115% Increase to Burtville Mineral Resource" Dated 21/10/2020

Criteria	Commentary
Sampling techniques	 Earliest RC drilling at Burtville used in the estimate was by Thames Mining NL (Thames), only 8 RC holes were used, limited information on the Thames drilling is reported by Aberfoyle Resources Ltd (Aberfoyle). Aberfoyle conducted RC drilling collecting 1m samples that were composited to 4m for analysis. Later programs riffle split the 1m sample into 2 samples, submitting 1 sample for analysis and retaining the duplicate sample onsite for future QAQC analysis. Gwalia Consolidated NL (Gwalia) RC drill cuttings were collected at 1m intervals and riffle split into 3kg samples for analysis.
	 Sons of Gwalia Ltd (SOG) mined the Burtville deposit during the 1990's with RC drilling carried out by the site mining department and not reported to the Department of Mines. In the Crescent Gold Ltd (Crescent) Bankable Feasibility Study of January 2005 (WAMEX reference A070179 appendix), extensive geological and mining data acquired from SOG were validated against original records by an independent geologist. Early Crescent Drilling submitted 1m 3-4kg samples for analysis. Later drilling by FML collected 1m samples by cone splitter off the drill rig and submitted for analysis.

Aberfoyle diamond core was sampled at 1m intervals. In areas of poor sample recovery core was sampled using a knift and chisel. Competent core was sawn, and one half submitted for analysis. Focus Diamond core was sampled at 1m intervals or to geological contacts, half core was submitted for assay. Aberfoyle states RC drilling was by a VK600 rig with a 5 ½ inch hole diameter. Aberfoyle diamond core was drilled from an RC pre-collar for all but 2 holes. Diamond core was drilled at NQ size. Gwalia Consolidated NL RC drilling was conducted using a 5 3/8 inch face sampling hammer for RC drilling. At hole completion, Focus and Crescent surveyed RC holes using True North Seeking Gyro tool. Otherwise, a single si camera downhole survey was used either "in-rod" or "open hole". Diamond core was drilled at NQ/HQ size. All drill core was oriented where competent by the drilling contractor using an or similar system. Prill sample ecovery Historic sample recovery is not well recorded. Aberfoyle details poor diamond core sample recovery (74% in some cases) above the clay/granodiorite contact. SOG recorded recovery as a visual qualitative estimate. RC sample recovery was recorded by a visual estimate during the logging process. DD sample recovery was measured and calculated (core loss) during the logging process. DD sample recovery was measured and calculated (core loss) during the logging process. DD core had generally good recovery. Aberfoyle logged 1m RC and Diamond intervals for colour, weathering, lithology and visual percentage estimate of sulf quartz. Gwalia logged 1m RC intervals for colour, lithology and quartz. Gwalia logged 1m RC intervals for colour, lithology and quartz. All data is entered directly into validating digital software. All Focus core samples were oriented where possible, marked at metre intervals and compared to the depth measurer. All Focus core samples were oriented where possible, marked at metre intervals and compared to the depth measurer.	e or hammer
Aberfoyle states RC drilling was by a VK600 rig with a 5 ½ inch hole diameter. Aberfoyle diamond core was drilled from an RC pre-collar for all but 2 holes. Diamond core was drilled at NQ size. Gwalia Consolidated NL RC drilling used a Gemco H22A rig and 4 ¼ diameter face sampling hammer drill. Crescent and Focus RC drilling was conducted using a 5 3/8inch face sampling hammer for RC drilling. At hole completion, Focus and Crescent surveyed RC holes using True North Seeking Gyro tool. Otherwise, a single sl camera downhole survey was used either "in-rod" or "open hole". Diamond core was drilled at NQ/HQ size. All drill core was oriented where competent by the drilling contractor using an or similar system. Drill sample ecovery Historic sample recovery is not well recorded. Aberfoyle details poor diamond core sample recovery (74% in some cases) above the clay/granodiorite contact. SGG recorded recovery was recorded by a visual estimate during the logging process. DD sample recovery was measured and calculated (core loss) during the logging process. DD core had generally good recovery. Aberfoyle logged 1m RC and Diamond intervals for colour, weathering, lithology and visual percentage estimate of sulf quartz. Gwalia logged 1m RC intervals for colour, lithology and quartz. SGG logging included colour, lithology, weathering, texture, grain size, veining Crescent and Focus RC samples were geologically logged to record weathering, rock type, alteration, mineralisation, s texture and any other notable features that are present. All data is entered directly into validating digital software. All data is entered directly into validating digital software.	
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All Focus core samples were oriented where possible, marked at metre intervals and compared to the depth measurer	tructure,
 All Focus core samples were oriented where possible, marked at metre intervals and compared to the depth measures 	
core blocks. Any loss of core was noted and recorded in the drilling database.	ments on the
All diamond core was logged for structure, geology and geotechnical data using the same system as that for RC.	
 Logging was qualitative, however the geologists often recorded quantitative mineral percentage ranges for the sulph present. 	nide minerals
The logging information was transferred into the company's drilling database once the log was complete.	
Diamond core was photographed one core tray at a time using a standardised photography jig. RC chip trays photographed.	are routinely
The entire length of all holes is geologically logged.	
Sub-sampling echniques and sample preparation • Early Aberfoyle programs split 1m samples on site before compositing to 4m for analysis. Where the composited assay > 0.5g/t Au, the individual 1m samples for that interval were submitted. Later programs submitted 1m samples. All sample assayed for Au by Genalysis Kalgoorlie for a single stage mix and grind sample preparation followed by 50g fire assay Au.	oles were
 Aberfoyle diamond core was also submitted to Genalysis Kalgoorlie for the same sample preparation and analysis as the samples outlined above. 	he RC
Gwalia submitted 3kg samples for analysis by Leonora Laverton Assay Laboratories.	
SOG Mining submitted 3m composites or 1m samples for analysis	
 Later SOG programs from year 2000 sent 3m composite samples to Ultra Trace Laboratories in Perth for Au analysis u aqua regia digest followed by ICP-MS determination. 	ising an
All Crescent and Focus samples were collected in a pre-numbered calico bag bearing a unique sample ID.	
 Core samples were taken from half core, cut using an Almonte automatic core saw. The remainder of the core was reta trays tagged with a hole number and metre mark. 	ained in core
 At the assay laboratory, samples were oven dried, crushed to a nominal 10mm using a jaw crusher (core samples only weighed. Samples in excess of 3kg in weight were riffle split to achieve a maximum 3kg sample weight. All samples we pulverized to 90% passing 75µm. 	
Gold analysis was by a 30 to 50g Fire Assay with an ICP-OES or AAS Finish.	
Different laboratories have been used over the years. Early Crescent Drilling submitted samples to SGS Leonora, drill swere also submitted to Kalgoorlie Assay Laboratories and Amdel for sample preparation and analysis.	samples
The assay laboratories' sample preparation procedures follow industry best practice, with techniques and practices tha appropriate for this style of mineralisation. Pulp duplicates were taken at the pulverising stage and selective repeats co the laboratories' discretion.	
QAQC checks involved inserting standards and field duplicate samples for RC. Diamond core field duplicates were not	taken.
Regular reviews of the sampling were carried out by the supervising geologist and senior field staff, to ensure all proce followed and best industry practice carried out.	
The sample sizes were appropriate for the type, style and consistency of mineralisation encountered during this phase	dures were

Criteria	Con	nmentary
Oritoria	OOII	exploration.
	•	Early Aberfoyle programs split 1m samples on site before compositing to 4m for analysis. Where the composited assay returned >0.5g/t Au, the individual 1m samples for that interval were submitted. Later programs submitted 1m samples. All samples were assayed for Au by Genalysis Kalgoorlie for a single stage mix and grind sample preparation followed by 50g fire assay analysis for Au.
	•	Aberfoyle diamond core was also submitted to Genalysis Kalgoorlie for the same sample preparation and analysis as the RC samples outlined above.
	•	Gwalia submitted 3kg samples for analysis by Leonora Laverton Assay Laboratories.
	•	SOG Mining submitted 3m composites or 1m samples for analysis
	•	Later SOG programs from year 2000 sent 3m composite samples to Ultra Trace Laboratories in Perth for Au analysis using an aqua regia digest followed by ICP-MS determination.
	•	All Crescent and Focus samples were collected in a pre-numbered calico bag bearing a unique sample ID.
	•	Core samples were taken from half core, cut using an Almonte automatic core saw. The remainder of the core was retained in core trays tagged with a hole number and metre mark.
	•	At the assay laboratory, samples were oven dried, crushed to a nominal 10mm using a jaw crusher (core samples only) and weighed. Samples in excess of 3kg in weight were riffle split to achieve a maximum 3kg sample weight. All samples were pulverized to 90% passing 75µm.
	•	Gold analysis was by a 30 to 50g Fire Assay with an ICP-OES or AAS Finish.
	•	Different laboratories have been used over the years. Early Crescent Drilling submitted samples to SGS Leonora, drill samples were also submitted to Kalgoorlie Assay Laboratories and Amdel for sample preparation and analysis.
	•	The assay laboratories' sample preparation procedures follow industry best practice, with techniques and practices that are appropriate for this style of mineralisation. Pulp duplicates were taken at the pulverising stage and selective repeats conducted at the laboratories' discretion.
	•	QAQC checks involved inserting standards and field duplicate samples for RC. Diamond core field duplicates were not taken.
	•	Regular reviews of the sampling were carried out by the supervising geologist and senior field staff, to ensure all procedures were followed and best industry practice carried out.
	•	The sample sizes were appropriate for the type, style and consistency of mineralisation encountered during this phase of exploration.
Quality of assay data and	•	The assay method and laboratory procedures were appropriate for this style of mineralisation. The fire assay technique was designed to measure total gold in the sample.
aboratory tests	•	No geophysical tools, spectrometers or handheld XRF instruments were used for assay determination.
	•	Aberfoyle details check sampling between labs for repeatability. They also submitted re-splits of the Thames RC drillholes and concluded results could be reproduced. Two samples were submitted for screen fire assay. In later programs they also submitted lab duplicates at approximately 1 in 20, standards at one per batch, resubmitted pulps with different sample ids as a check and submitted field duplicates.
	•	The QA/QC process described above was sufficient to establish acceptable levels of accuracy and precision. All results from assay standards and duplicates were scrutinised to ensure they fell within acceptable tolerances and where they did not further analysis was conducted as appropriate.
Verification	•	Historic logging data is verified against available WAMEX reports.
of sampling and	•	Crescent Gold Ltd engaged the services of an Independent Geologist to validate the electronic databases acquired from SOG
assaying		using original records. Primary logging data is sent in digital format to the company's Database Administrator (DBA) as often as was practicable. The DBA imports the data into an acQuire database, with assay results merged into the database upon receipt from the laboratory. Once
		loaded, data was extracted for verification by the geologist in charge of the project.
Location of data points	•	Aberfoyle used a local grid with unknown survey methods.
iata points	•	Gwalia used survey consultants to survey their holes, the Aberfoyle drilling and previous drill programs. Gwalia also established permanent survey stations.
	•	During mining operations by SOG site surveyors surveyed the drill collars.
	•	Crescent and Focus drilled holes were also surveyed by site based mine survey team.
	•	Crescent/Focus used True North Seeking Gyro for RC downhole surveys. A Reflex single shot camera was used for "single shot" surveys whilst advancing diamond drill holes.
	•	All coordinates and bearings use the MGA94 Zone 51 grid system.
	•	FML utilises Landgate sourced regional topographic maps and contours as well as internally produced survey pick-ups produced by the mining survey teams utilising DGPS base station instruments.
Data spacing and	•	Drill spacing at Burtville is variable with 10m x 10m spacing in areas RC grade control drilled, with a nominal 20m x 20m spacing
distribution		across most of the east and west existing pit areas. Drilling spacing is irregular across the saddle and increases out to 40m x 60m along the southern extents of the deposit. The average depth of the SOG drilling was 50m, more recent Crescent and Focus drilling was an average of 81 and 89m, respectively.
Orientation of	•	Drilling was designed based on known/developing geological models, field mapping, verified historical data, cross-sectional and
data in relation		long-sectional interpretation.

Criteria	Commentary
o geological structure	Where achievable, drill holes were oriented at right angles to strike of deposit, with dip optimised for drill capabilities and the dip of the ore body.
Sample security	Historic sample security is unknown.
	Crescent and Focus samples were reconciled against the sample submission with any omissions or variations reported.

Section 1 Details for the Beasley Creek deposit from ASX Announcement "Beasley Creek Mineral Resource Grows by 29%" Dated 20/08/2020

	" Dated 20/08/2020
Criteria	Commentary
Sampling	Focus Minerals RC Sampling
techniques	 RC percussion drill chips were collected through a cone splitter from the drill rig. The bulk sample from drilling was placed in neat rows directly on the ground (not bagged) with the nominal 2-3kg calico split sub-sample placed on top of the corresponding pile.
	 RC chips were passed through a cone splitter to achieve a nominal sample weight of approximately 3kg. The splitter was levelled at the beginning of each hole. Geological logging defined whether a sample was to be submitted as a 1m cone split sample or a 4m spear composite sample. Split samples (1m) were transferred to sample numbered calico bags for submission to the laboratory. Composite samples were spear sampled using a scoop to obtain a small representative sample and deposited into numbered sample bags.
	Focus Minerals Diamond Sampling
	• Diamond core was sampled across geologically identified zones of mineralisation, the sample widths varied between a minimum of 0.2m and a maximum of 1.2m with material on either side sampled to capture the entire mineralised zone.
	 The diamond core was marked up for sampling by the supervising geologist during the core logging process, with sample intervals determined by the presence of lithology, alteration and where applicable core loss. The core was cut in half using a core saw and the same half of the core (RHS looking downhole) was routinely sent to the laboratory for analysis. Some soft core was sampled half by using a bolster, and some fractured quartz core were cut in half by using manual diamond core saw to ensure half core was sampled.
	A small number of whole core samples where routinely collected for bulk density analysis. These samples were submitted to the same lab for gold analysis after bulk density measurement. **These samples were submitted to the same lab for gold analysis after bulk density measurement.** **These samples were submitted to the same lab for gold analysis after bulk density measurement.** **These samples were submitted to the same lab for gold analysis after bulk density measurement.** **These samples were submitted to the same lab for gold analysis after bulk density measurement.** **These samples were submitted to the same lab for gold analysis after bulk density measurement.** **These samples were submitted to the same lab for gold analysis after bulk density measurement.** **These same lab for gold analysis after bulk density measurement.** **These same lab for gold analysis after bulk density measurement.** **The same lab for gold analysis after bulk density measurement.** **The same lab for gold analysis after bulk density measurement.** **The same lab for gold analysis after bulk density measurement.** **The same lab for gold analysis after bulk density measurement.** **The same lab for gold analysis after bulk density measurement.** **The same lab for gold analysis after bulk density measurement.** **The same lab for gold analysis after bulk density measurement.** **The same lab for gold analysis after bulk density measurement.** **The same lab for gold analysis after bulk density measurement.** **The same lab for gold analysis after bulk density measurement.** **The same lab for gold analysis after bulk density measurement.** **The same lab for gold analysis after bulk density measurement.** **The same lab for gold analysis after bulk density measurement.** **The same lab for gold analysis after bulk density measurement.** **The same lab for gold analysis after bulk density measurement.** **The same lab for gold analysis after bulk density measurement.** **The same lab for gold ana
	 WMC Sampling RC samples were collected in plastic bags in 1m intervals.
	 Diamond core was sampled to at 1m intervals or on geological contacts. Metex Sampling Diamond core was halved by core saw or hand split when too friable. Individual 1m samples of 1/2 core were submitted for assay.
Drilling	Focus Minerals Drilling
techniques	RC drilling was conducted using a 5 3/8inch face sampling hammer for RC drilling.
,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,	At hole completion, downhole surveys for RC holes were completed at a 10m interval by using True North Seeking Gyro tool.
	At hole completion diamond holes were survey using a single shot tool at a range of intervals between 20m and 50m, averaging 30m
	Diamond drill holes with dips less than 50 degrees were collared from surface to a predetermined depth using a rock roller bit.
	Where possible on holes with dips more than 50 degrees an RC pre-collar was completed to improve drilling efficiency.
	All pre-collars were cased off and the diamond component of the drill hole completed using HQ3 (producing 63mm core diameter)
	equipment.
	 Wherever core conditions and hole orientation would allow, drill core was oriented by the drilling contractor using the electronic ACT III Tool.
	WMC Drilling
	It has been reported by Metex that RC holes were drilled with conventional crossover subs.
	Some of the later diamond holes had pre-collars, otherwise it was diamond core from surface and HQ and NQ coring.
	Metex
	Diamond holes had an RC pre-collar and then cored to end of hole.
Drill	Focus Minerals Drilling
sample recovery	RC sample recovery was recorded by a visual estimate during the logging process.
recovery	 DD sample recovery was measured and calculated (core loss) during the logging process. DD core had generally reasonable recovery <10% core loss in and around mineralisation. Some holes had more than 30% core loss. Where this core loss was experienced around HG and VHG it likely had a material impact on reported calculated intersection grade as all core loss in reported intersections was fully diluted and assigned a grade of 0.0g/t Au. WMC Drilling
	Sample recovery was not recorded Metex Drilling
	Recorded <10% core loss in diamond core and mostly excellent sample recovery in RC drilling.
Logging	Focus Minerals Drilling
•	 All RC samples were geologically logged to record weathering, regolith, rock type, colour, alteration, mineralisation, structure, texture and any other notable features that are present. All data is entered directly into validating digital software directly.
	 All core samples were oriented where possible, marked into metre intervals and compared to the depth measurements on the core blocks. Any loss of core was noted and recorded in the drilling database.

Criteria	Commentary
- Troria	All diamond core was logged for structure, geology and geotechnical data using the same system as that for RC.
	Logging was qualitative, however the geologists often recorded quantitative mineral percentage ranges for the sulphide minerals present.
	The logging information was transferred into the company's drilling database once the log was complete.
	Diamond core was photographed one core tray at a time using a standardised photography jig. RC chip trays are routinely photographed.
	The entire length of all holes is geologically logged, except for rock roller diamond pre-collars, which produce no sample. WMC Drilling
	RC samples were logged to record colour, grain size, occasional weathering, structural fabric and rock type.
	Diamond core was logged to lithological boundaries, recording rock type, structure, texture, alteration and veining. The pre-collar drill cuttings do not appear to have been logged. Metex Drilling
	RC and DD were logged for: Colour, Weathering, structural Fabric, Alteration Veining, Mineralisation and lithology
Sub-sampling	Focus Minerals Drilling
techniques	All samples were collected in a pre-numbered calico bag bearing a unique sample ID.
and sample preparation	 At the assay laboratory, all samples were oven dried, crushed to a nominal 10mm using a jaw crusher (core samples only) and weighed. Samples in excess of 3kg in weight were riffle split to achieve a maximum 3kg sample weight before being pulverized to 90% passing 75µm.
	Gold analysis was by 40g Fire Assay with an AAS Finish.
	Jinning Testing & Inspection completed the assay testing, with sample preparation completed in Kalgoorlie or Perth and analysis completed in Perth and Kalgoorlie.
	 The assay laboratories' sample preparation procedures follow industry best practice, with techniques and practices that are appropriate for this style of mineralisation. Pulp duplicates were taken at the pulverising stage and selective repeats conducted at the laboratories' discretion.
	 QAQC checks involved inserting standards 1:20 samples (with minimum 3 standards every submission). Duplicate samples for RC were achieved by producing 2 samples for each metre one hole every 20th hole drilled and submitting all produced samples. The remaining bulk sample was also bagged to plastic bags for retention and further checks. Diamond core field duplicates were not taken.
	Regular reviews of the sampling were carried out by the supervising geologist and senior field staff, to ensure all procedures were followed and best industry practice carried out.
	The sample sizes were appropriate for the type, style and consistency of mineralisation encountered during this phase of exploration. WAY OR JULY
	WMC Drilling RC samples were collected as 1m samples and submitted to the WMC Windarra laboratory for Au analysis by fire assay.
	Diamond core was submitted as 1m samples or to geological contact to the Windarra laboratory for fire assay.
	Metex
	RC was collected into plastic bags in 1m intervals. All dry sample were riffle split to return a representative split sample for analysis. Any wet/Moist samples where 50mm PVC spear sampled.
	Diamond drilling was ½ core sampled to geological intervals and generally 1m intervals.
	All Au Analysis was completed at were submitted to Amdel Kalgoorlie for 50g Fire Assay for Au
Quality of assay data and laboratory tests	Focus Minerals Drilling The assay method and laboratory procedures were appropriate for this style of mineralisation. The fire assay technique was
laboratory tests	designed to measure total gold in the sample. No geophysical tools, spectrometers or handheld XRF instruments were used for assay determination.
	The QA/QC process described above was sufficient to establish acceptable levels of accuracy and precision. All results from assay standards and duplicates were scrutinised to ensure they fell within acceptable tolerances and where they didn't further analysis was conducted as appropriate.
	Umpire samples are collected on a routine basis will be submitted to independent ISO certified labs in 2020
	Additional bulk mineralised RC samples have also been collected and retained for follow up QAQC, metallurgical and sample characterisation purposes.
	WMC Drilling Notwithstanding the lack of information on WMC laboratory techniques, the assay method and laboratory procedures were appropriate for this style of mineralisation. The fire assay technique was designed to measure total gold in the sample.
	Metex Drilling An appropriate assay method and laboratory procedures were used for the style of mineralisation. Metex reported frequent inspections of the drill rig cyclone and splitter whilst
	drilling. Duplicates were taken at a frequency of approx. one in thirty. Laboratory replicates were also reported, and results monitored.
Verification	Focus Minerals Drilling
of sampling and	Significant intervals were visually inspected by company geologists to correlate assay results to logged mineralisation. Consultants were not used for this process.

Criteria	Commentary								
assaying	 Primary logging data is sent in digital format to the company's Database Administrator (DBA) as often as was practicable. The DBA imports the data into an acQuire database, with assay results merged into the database upon receipt from the laboratory. Once loaded, data was extracted for verification by the geologist in charge of the project. 								
Location of data	Focus Minerals Drilling								
points	 Drill collars are surveyed after completion using a DGPS instrument. Where possible, all drill core was oriented by the drilling contractor using an ACT III electronic system. 								
	 A True North Seeking Gyro for RC end of holes surveys or a Reflex single shot camera for diamond drilling was used for "single shot" surveys whilst advancing drilling. 								
	All coordinates and bearings use the MGA94 Zone 51 grid system.								
	 Focus Minerals utilises Landgate sourced regional topographic maps and contours as well as internally produced survey pick-ups produced by the mining survey teams utilising DGPS base station instruments. 								
	After completion, the drill hole locations were picked up by DGPS with accuracy of +/-20cm. WMC Drilling								
	Holes were surveyed by WMC survey staff in local mine grid Metex Drilling								
	 Holes were surveyed by a consultant survey company. Diamond core holes were downhole surveyed by an Eastman single shot camera. 								
Data spacing	Beasley Creek drill spacing approximates 40m x 20m								
and distribution	Spacing is deemed to be appropriate for the type of mineralisation								
Orientation of data in relation	 Drilling was designed based on known/developing geological models, field mapping, verified historical data, cross-sectional and long-sectional interpretation. 								
to geological structure	 Where achievable, drill holes were oriented at right angles to strike of deposit, with dip optimised for drill capabilities and the dip of the ore body. Please note this was not always possible in the NW part of the pit where relatively complex mineralisation has been intersected in the footwall of the Beasley Creek Shear. 								
	True widths have not been calculated for reported intersections. However, drill orientation was wherever possible consistently optimised to approximate true width of mineralisation.								
Sample security	Focus Minerals Drilling								
	All samples were reconciled against the sample submission with any omissions or variations reported to Focus Minerals.								
	All samples were bagged in a tied numbered calico bag. The bags were placed into green plastic bags and cable tied before								
	depositing into sample cages. Sample cages were routinely delivered directly from site to the Kalgoorlie laboratories by Focus								
	Minerals personnel and or freight contractors.								
	WMC and Metex sample security is not recorded.								

 Section 1 Details for the Beasley Creek South deposit from ASX Announcement "Beasley Creek South Delivers High Grade Mineral Resource" Dated 15/07/2020

mmentary EMI PC Sampling									
EMI_DC Compling									
ML RC Sampling									
RC percussion drill chips were collected through a cone splitter from the drill rig. The bulk sample from drilling was placed in neat rows directly on the ground (not bagged) with the nominal 2-3kg calico split sub-sample placed on top of the corresponding pile.									
RC chips were passed through a cone splitter to achieve a nominal sample weight of approximately 3kg. The splitter was levelled at the beginning of each hole. Geological logging defined whether a sample was to be submitted as a 1m cone split sample or a 4m spear composite sample. Split samples (1m) were transferred to sample numbered calico bags for submission to the laboratory. Composite samples were spear sampled using a scoop to obtain a small representative sample and deposited into numbered sample bags. FML Diamond Sampling									
Diamond core was sampled across geologically identified zones of mineralisation, the sample widths varied between a minimum of 0.2m and a maximum of 1.2m with material on either side sampled to capture the entire mineralised zone.									
The diamond core was marked up for sampling by the supervising geologist during the core logging process, with sample intervals determined by the presence of lithology, alteration, and where applicable core loss. The core was cut in half using a core saw and the same half of the core (RHS looking downhole) was routinely sent to the laboratory for analysis. Some soft core was sampled half by using a bolster, and some fractured quartz core were cut in half by using manual diamond core saw to ensure half core was sampled.									
A small number of whole core samples where routinely collected for bulk density analysis. These samples were submitted to the same lab for gold analysis after bulk density measurement.									
RC drilling was conducted using a 5 3/8inch face sampling hammer for RC drilling.									
At hole completion, downhole surveys for RC holes were completed at a 10m interval by using True North Seeking Gyro tool.									
At hole completion diamond holes were surveyed using a single shot tool at a range of intervals between 20m and 50m, averaging 30m.									
Diamond drill holes with dips less than 50 degrees were collared from surface to a predetermined depth using a rock roller bit.									
Where possible on holes with dips more than 50 degrees an RC pre-collar was completed to improve drilling efficiency.									
All pre-collars were cased off and the diamond component of the drill hole completed using HQ3 (producing 63mm core diameter) equipment.									
Wherever core conditions and hole orientation would allow, drill core was oriented by the drilling contractor using the electronic ACT III Tool.									
RC sample recovery was recorded by a visual estimate during the logging process.									
DD sample recovery was measured and calculated (core loss) during the logging process. DD core had generally reasonable recovery									

Criteria	Commentary
recovery	<10% core loss in and around mineralisation. Some holes had more than 30% core loss. Where this core loss was experienced around HG and VHG it likely had a material impact on reported calculated intersection grade as all core loss was fully diluted and assigned
Logging	 a grade of 0.0g/t Au. All RC samples were geologically logged to record weathering, regolith, rock type, colour, alteration, mineralisation, structure, texture
	 and any other notable features that are present. All data is entered directly into validating digital software directly. All core samples were oriented where possible, marked into metre intervals and compared to the depth measurements on the core
	blocks. Any loss of core was noted and recorded in the drilling database.
	 All diamond core was logged for structure, geology and geotechnical data using the same system as that for RC. Logging was qualitative, however the geologists often recorded quantitative mineral percentage ranges for the sulphide minerals present.
	 The logging information was transferred into the company's drilling database once the log was complete. Diamond core was photographed one core tray at a time using a standardised photography jig. RC chip trays are routinely photographed.
Out a surelline	The entire length of all holes is geologically logged, except for rock roller diamond pre-collars, which produce no sample. All the entire length of all holes is geologically logged, except for rock roller diamond pre-collars, which produce no sample.
Sub-sampling techniques	 All samples were collected in a pre-numbered calico bag bearing a unique sample ID. At the assay laboratory, all samples were oven dried, crushed to a nominal 10mm using a jaw crusher (core samples only) and
and sample preparation	weighed. Samples in excess of 3kg in weight were riffle split to achieve a maximum 3kg sample weight before being pulverized to 90% passing 75µm.
	Gold analysis was by 40g Fire Assay with an AAS Finish.
	 Jinning Testing & Inspection completed the assay testing, with sample preparation completed in Kalgoorlie or Perth and analysis completed in Perth.
	 The assay laboratories' sample preparation procedures follow industry best practice, with techniques and practices that are appropriate for this style of mineralisation. Pulp duplicates were taken at the pulverising stage and selective repeats conducted at the laboratories' discretion.
	 QAQC checks involved inserting standards 1:20 samples (with minimum 3 standards every submission). Duplicate samples for RC were achieved by producing 2 samples for each metre one hole every 20th hole drilled and submitting all produced samples. The
	remaining bulk sample was also bagged to plastic bags for retention and further checks. Diamond core field duplicates were not taken.
	Regular reviews of the sampling were carried out by the supervising geologist and senior field staff, to ensure all procedures were followed and best industry practice carried out.
Quality of assay	 The sample sizes were appropriate for the type, style and consistency of mineralisation encountered during this phase of exploration. The assay method and laboratory procedures were appropriate for this style of mineralisation. The fire assay technique was designed
data and	to measure total gold in the sample.
laboratory tests	 No geophysical tools, spectrometers or handheld XRF instruments were used for assay determination. The QA/QC process described above was sufficient to establish acceptable levels of accuracy and precision. All results from assay
	standards and duplicates were scrutinised to ensure they fell within acceptable tolerances and where they didn't further analysis was conducted as appropriate.
	Umpire samples are collected on a routine basis will be submitted to independent ISO certified labs in 2020.
	 Additional bulk mineralised RC samples have also been collected and retained for follow up QAQC, metallurgical and sample characterisation purposes.
Verification	Significant intervals were visually inspected by company geologists to correlate assay results to logged mineralisation. Consultants were not used for this process.
of sampling and	 were not used for this process. Primary logging data is sent in digital format to the company's Database Administrator (DBA) as often as was practicable.
assaying	The DBA imports the data into an acQuire database, with assay results merged into the database upon receipt from the laboratory.
	Once loaded, data was extracted for verification by the geologist in charge of the project.
Location of data points	 Drill collars are surveyed after completion using a DGPS instrument. Where possible, all drill core was oriented by the drilling contractor using an ACT III electronic system.
	 A True North Seeking Gyro for RC end of holes surveys or a Reflex single shot camera for diamond drilling was used for "single shot" surveys whilst advancing drilling.
	All coordinates and bearings use the MGA94 Zone 51 grid system.
	 FML utilises Landgate sourced regional topographic maps and contours as well as internally produced survey pick-ups produced by the mining survey teams utilising DGPS base station instruments.
	After completion, the drill hole locations were picked up by DGPS with accuracy of +/-20cm.
Data spacing	Beasley Creek South drill spacing on indicated resource parts of the main lode between surface and 130m depth approximates 20m x 25m. There are limited holes targeting the main lode beneath 130m depth and these parts of the model are classified as inferred.
and distribution	 Drill spacing on the hanging wall lodes approximates 20m x 40m. however there are sample gaps and these lodes have been classified as inferred at this stage.
	Spacing is deemed to be appropriate for the type of mineralisation.
Orientation of	Drilling was designed based on previous geological models, historical data, cross-sectional and long-sectional interpretation.
data in relation to	 Where achievable, drill holes were oriented at right angles to strike of deposit, with dip optimised for drill capabilities and the dip of the ore body.
geological	True widths have not been calculated for reported intersections. However, drill orientation was wherever possible consistently
structure	optimised to approximate true width of mineralisation.
Sample security	All samples were reconciled against the sample submission with any omissions or variations reported to FML.
	 All samples were bagged in a tied numbered calico bag. The bags were placed into cable tied numbered green bags and loaded into bulka cages. On an approximately biweekly basis bulka cages were delivered with a sample submission sheet directly to the Kalgoorlie laboratories by FML personnel or freight contractor.
i	parameters.

 Section 1 Details for the Wedge deposit from ASX Announcement "Wedge Open Pit Resource Update" Dated 24/01/2020

Criteria		Commentary
Sampling	٠	This report relates to results from Reverse Circulation (RC) and diamond core (DDH) drilling.
techniques	•	Wedge has been drilled by various companies over the years and this report contains information on holes drilled by Focus Minerals Ltd (FML); Teck Explorations Ltd (Teck) and Hillmin Gold Mines Pty Ltd (Hillmin), which was renamed Ashton Gold Mines Pty Ltd (Ashton) in October 1989. This was dissolved in December 1990 with all rights and obligations assumed by Ashton Gold (WA) Ltd. Metex Resources NL (Metex) subsequently acquired the tenement and conducted 2 drill campaigns.
	•	Teck collected 1m samples in plastic bags from the drill rig cyclone and were split for assay. The 1m splits were combined to form 2m samples which were assayed for gold by AAS methods. Where anomalous AAS results were returned, 1m samples were submitted for fire assay.
	•	Hillmin/Ashton collected 1m RC samples via a riffle splitter. A spear sample was taken of the intervals in the form of 2m and 4m composites for subsequent drill programs. Where composite assays exceeded 0.25 ppm Au, the corresponding 1m sample was submitted.
	•	Ashton recorded duplicate samples in the assay files. Hillmin reported a comparison check between assay laboratories in a 1988 WAMEX report.
	•	Hillmin diamond core was sampled as either 4m filleted composites or a sawn core sampled to lithological contacts.
	•	Metex collected 1m samples split from the rig using a cyclone riffle splitter. A 4m composite sample was taken by spear sampling the 1m interval spoils. Resampling of the composite intervals where assay results were 0.1 ppm Au or greater was carried out on an individual 1m basis.
	•	The information of sampling techniques below applies to the drill holes drilled by Focus Minerals (FML) only.
	•	RC percussion drill chips were collected through a cyclone and in-line cone splitter under driller control.
	•	RC chips were passed through a cone splitter to achieve a sample weight of approximately 3kg. The splitter was levelled at the beginning of each hole using a bullseye level. The spoils were collected in green bags or heaped neatly on the ground at 1m intervals. Samples for assay were collected in pre-numbered calico bags.
	•	Standards of appropriate grade were inserted into the RC and DDH sample runs at a rate of 1 per 20. No blanks were used as many of the primary samples on the project recorded assays below or close to the detection limit making the role of the blank superfluous. Instead, gold geochemical standards with low expected values were utilised regularly.
	•	RC samples were collected as either a 4m composite taken from the bulk 1m sample or the 1m cyclone cone split sample. Where 4m composites returned a grade over 0.2ppm the corresponding cyclone split sample was collected.
	•	Diamond core was sampled across identified zones of mineralisation by site geologists, the sample widths varied between a nominal minimum of 0.3 m and a nominal maximum of 1m.
	•	The diamond core was marked up for sampling by the supervising geologist during the core logging process, with sample intervals determined by the presence of mineralisation and/or alteration. Sample intervals did not overlap zones of core loss. The core was cut in half using an automatic core saw. Samples for assay were put into pre-numbered calico bags.
	•	At the assay laboratory all calico bagged assay samples were oven dried, core samples (only) crushed to a nominal 10mm using a jaw crusher and weighed. Samples in excess of 3kg in weight were riffle split to achieve a maximum 3kg sample weight before being pulverized to 90% passing 75µm.
Drilling	•	Only RC and Diamond drilling methods have been included in the resource estimate.
techniques	•	Ashton reports state drilling was by a face sampling hammer RC rig.
	•	Hillmin used rotary mud pre-collars or existing RC holes for its diamond drilling using a PQ diameter drill bit.
	•	Metex used a face sampling hammer RC drill rig with 5 3/8" drill bits.
Drill	•	All FML drilling was completed using RC gear with face sampling hammer or HQ-PQ triple tube diamond drilling
sample		Teck made no attempt to estimate cutting recovery due to wide range of sample weights and wet samples.
recovery	.	Hillmin early RC drill logs do not document drill recovery, however later drill logs have a percentage estimate recorded. Hillmin Diamond core recovery is recorded in the drill logs.
		Metex recorded sample recovery in the drill logs.
	•	FML RC sample recovery was recorded by a visual estimate during the logging process. Diamond core recovery was measured and recorded as a percentage of the core "run". That is, the measured length of core recovered against the increase in hole depth.
Logging	•	Teck logged the entire drill hole for colour, rock type, texture, weathering, structure, alteration and veining.
	•	Hillmin logged the entire drill hole for colour, weathering, rock type, texture, structure, alteration, veining and mineralisation.
		Ashton logged the entire hole for weathering, rock type, structure, texture, alteration, veining, mineralisation and colour.
	•	Hillmin diamond core was photographed, geotechnically logged and inspected by Golder Associates prior to diamond sawing and sampling. Holes were also geologically logged for colour, weathering, rock type, texture, structure, alteration, veining and mineralisation.
	•	Metex holes were logged for colour, weathering, rock type, texture, structure, alteration, veining and mineralisation.
	•	The information of logging techniques below applies to the drill holes drilled by FML only.
	•	Core hole samples were oriented where possible and marked into metre intervals with relation to hole depth. Any loss of core was noted and recorded in the drilling database. Recovery and RQD measurements were recorded. SG readings were taken using the water displacement method on competent representative lengths of core. SG samples were collected nominally at 10m intervals
		through zones of waste rock and at 1-5m intervals through zones of mineralisation. All RC and DDH samples were geologically logged to record weathering, grain size, lithology, texture, alteration, veining,
		mineralisation and structure. In addition to parameters logged over RC chips, all diamond core was also logged for structure. If an orientation line was available,
		structure orientation measurements were taken and recorded.
	•	The logging information was transferred into the company's drilling database once the log was complete. Logging was qualitative, however the geologists often recorded quantitative mineral percentage ranges for the sulphide minerals

Criteria	(Commentary									
		present.									
1	•	Diamond core was photographed one core tray at a time using a standardised photography jig.									
	•	Samples from RC holes were archived in standard 20m plastic chip trays.									
	•	The entire length of all holes was logged.									
Sub-sampling techniques	•	Teck submitted 2m composites to Analabs in Kalgoorlie. The composite samples were analysed by aqua regia digest, with subsequent anomalous values and/or chert intersections were assayed at 1m intervals by fire assay with an AAS finish.									
and sample preparation	•	Hillmin submitted 4m composite samples in numbered bags that corresponded to the 1m intervals they had composited. Samples were sent to AAS Laboratories in Leonora, RDL or SGS for Fire Assay. Where the composite sample exceeded 0.25 ppm Au, the pre-numbered individual 1m samples were submitted for Fire Assay to a lower detection limit of 0.01ppm Au.									
	•	thon submitted 4m composite samples to SGS Kalgoorlie, samples were dried, jaw crushed, hammer milled, split and pulverised. Imples were analysed for gold by fire assay on a 50g charge to a lower limit of detection of 0.01 ppm Au. Where the composite assay ceeded 0.25 ppm, the relevant 1m interval was submitted to SGS for analysis.									
	•	Hillmin diamond core was sampled as either 4m filleted composites or a sawn core sampled to lithological contacts. Samples were submitted to SGS Kalgoorlie for gold analysis.									
	•	Metex submitted 4m composites collected by spear sampling for gold analysis to Amdel Laboratories Kalgoorlie, for 50g Fire Assay to 0.01 lower detection limit. Resampling of composite intervals where results exceeded 0.1ppm Au was carried out on an individual 1m basis.									
	•	The information of sub-sampling and sample preparation below applies to the drill holes drilled by FML only. Core samples were taken from half core, cut using an automatic core saw. The remainder of the core was retained in core trays									
		tagged with a hole number and metre mark. RC samples were cone split to a nominal 2.5kg to 3kg sample weight. The drilling method was designed to maximise sample recovery and delivery of a clean, representative sample into the calico bag.									
	•	The samples were collected in a pre-numbered calico bag bearing a unique sample ID. Samples were crushed to 75µm at the laboratory and riffle split (if required) to a maximum 3kg sample weight. Gold analysis was a 40g Fire Assay for individual samples with an ICP-OES or AAS Finish.									
	•	The assay laboratories' sample preparation procedures follow industry best practice, with techniques and practices that are appropriate for this style of mineralisation. Pulp duplicates were taken at the pulverising stage and selective repeats conducted at the laboratories' discretion.									
	•	Regular reviews of the sampling were carried out by the supervising geologist and senior field staff, to ensure all procedures were followed and best industry practice carried out.									
	•	The sample sizes are considered to be appropriate for the type, style and consistency of mineralisation encountered during this phase of exploration.									
Quality of assay data and laboratory tests	•	Hillmin ran a laboratory comparison check during the 1987 drill program comparing RDL Assay results to SGS Assay results for selected drill hole intervals. Overall, 23 drill holes (354 samples) were submitted for an AAS and Fire Assay check to a 0.001 ppm Au limit of detection. The results were generally comparable.									
	•	The information on quality of assay data and laboratory tests below applies to the drill holes drilled by FML only.									
	•	No geophysical tools, spectrometers or handheld XRF instruments were used.									
	•	For RC drilling, every 15th hole was drilled producing 2 duplicate cone split samples. For these holes both duplicate samples for the entire hole were submitted for analysis. Diamond core field duplicates were not taken. Standards were inserted every 20th sample number. All sample despatches had a minimum of 3 standards inserted.									
	•	All results from assay standards and duplicates were scrutinised to ensure they fell within acceptable tolerances.									
	•	Focus twinned several historic holes to check the location and accuracy of the historic sampling data and the results are considered to be acceptable.									
Verification	•	Significant intervals were visually inspected by company geologists to correlate assay results to logged mineralisation.									
of sampling	•	Historic sampling and assaying have been checked against hard copy WAMEX reports.									
and assaying	•	The Hillmin diamond program from 1986 was designed to twin RC holes drilled in previous years. The ATR (Annual Technical Report) notes in general diamond intersections were narrower and of lower grade. This was attributed to narrower sampling intervals and variations in grade along strike as diamond holes were drilled approx. 5m away from the RC hole they were twinning to avoid any cavities created in the drilling of the RC hole.									
	•	FML primary data is sent in digital format to the company's Database Administrator (DBA) as often as was practicable. The DBA imports the data into an acQuire database, with assay results merged into the database upon receipt from the laboratory. Once loaded, data was extracted for verification by the geologist in charge of the project.									
	•	No adjustments were made to any current or historic data. If data could not be validated to a reasonable level of certainty it was not used in any resource estimations.									
Location of data points	•	Historical surveying methods are not stated, however later Hillmin WAMEX reports note the use of registered surveyors to record the drill hole collars in a local grid.									
1	•	Ashton collar survey methods are unknown and reported in local grid.									
	•	Metex spent time re-establishing the mine grids, creating baselines and gridlines. They tied the previous local and mine grid data into AMG co-ordinates.									
	•	Focus personnel confirmed location data of original grid and resurveyed baseline stakes using DGPS. FML drill collars were surveyed upon completion, using a DGPS instrument.									
1	:	Diamond drill core was oriented by the drilling contractor using an electronic system.									
1	•	For RC, a north-seeking gyroscope tool was used to survey down hole.									
1	•	For DDH a magnetic single shot survey was completed at 30m intervals during hole advance.									
	•	All coordinates and bearings use the MGA94 Zone 51 grid system.									
1	•	Historic holes have been converted to MGA94 Zone 51 grid system in Acquire.									
	•	Historic hole collars were sometimes still visible and re-surveyed to check the accuracy of the grid conversion. The comparison was									

Criteria	Commentary
	 considered within acceptable error limits of using a DGPS unit. FML utilises Landgate sourced regional topographic maps and contours as well as internally produced survey pick-ups produced by the mining survey teams utilising DGPS base station instruments.
Data spacing and distribution	 Drill spacing along the Wedge trend is quite regular at a 25x25m spaced pattern along strike. 1m samples were collected by riffle splitter for RC holes and 4m composites were collected by spear sampling the individual 1m intervals.
Orientation of data in relation to geological structure	 Drilling was designed based on known geological models, field mapping, verified historical data and cross-sectional interpretation. Drill holes were either vertical or oriented at right angles to strike of deposit, with dip optimised for drill capabilities and the dip of the ore body.
Sample security	 All samples were reconciled against the sample submission with any omissions or variations reported to FML. All samples were bagged in a tied pre-numbered calico bag and grouped into green plastic bags. The bags were placed into bulka bags with a sample submission sheet and kept within the Laverton yard until ready for transport to Kalgoorlie by transport courier or FML staff. Historic sample security is not recorded.
Audits or reviews	 After Metex Resources acquired the WMC data, a thorough data validation of the WMC Surpac database against raw data hard copy information and Eastman photographic survey shots was conducted in the mid 1990's. Focus Minerals has purchased the Metex validated database and associated hard copies as part of the Lancefield project acquisition.

Section 2 Reporting of Exploration Results (Criteria listed in the preceding section also apply to this section.)

Section 2 Details for the Karridale Deposit from ASX Announcement "Karridale Mineral Resource increases by 60%" Dated 24/09/2020

Dated	24/09/2020									
Criteria	Commentary									
Mineral tenement and land tenure status	 The drilling was conducted on tenements E38/2032, M38/008, M38/089, M38/261 and M38/073 +91% owned by Focus Minerals (Laverton) Pty Ltd. In JV with Goldfields (GSM). Exploration expenditure by FML is continuing to increase the proportion of the JV tenement held by FML. All tenements are in good standing. The Nyalpa Pirniku claim has been lodged over the Laverton project areas. No claims have been determined at this time 									
Exploration	Karridale was originally mined by small scale shafts targeting high grade veins. The shallow shafts and drives are developed throughout									
done by other	the area and an excellent vector within the interpreted Karridale Footprint.									
parties	 Karridale has been explored by several parties including Sons of Gwalia and Crescent. Sons of Gwalia explored for oxide resources and mined an oxide resource at Burtville which was later followed into hard rock by a Crescent. 									
	 Exploration by Focus on Karridale targets the interpreted mineralised footprint which is based on: historical mining, structural interpretation, geological model, geophysics and continued success with infill of 2018 320m x 160m and 160m x 80m footprint drilling. 									
Geology	 Karridale mineralisation is hosted in an interpreted half graben on the SE side of a large Gabbro intrusion. The half graben is composed from northwest to south east by: Gabbro with dolerite chill margin. The south and south east sides of the Gabbro dip to the south and south east Structurally juxtaposed against the south and south east gabbro contacts are a series of shallow north east dipping pillow basalt flows. The basalt flows are generally 5-+10m in thickness and marked by distinct vesicle rich autobreccia tops. Laterally and down dip extensive interflow meta sediments/volcaniclastics are sandwiched between the flows. The basalt package is overlain and partly structurally interfingered with intermediate volcanic tuff and interbedded sandstone-black shale sequence. This volcano sedimentary sequence also hosts stacked shallow NW drilling mineralised shears. The shallow NW dipping shears are predominantly developed in the interflow sediments. These structures control the location of some limited 1 – 3m thick dolerite sills sourced from the Karridale gabbro. Gold mineralisation appears to postdate the Karridale gabbro intrusion but, in general is very tightly focused into the strata bound and stacked interflow meta – sediments/volcaniclastics. These interflow units preferentially take up the structural strain, alteration and mineralised veining. Additional higher-grade mineralisation is located in cross faults with north and north west strikes. 									
Drill hole information	Drill holes that have been previously reported see table below for reporting reference:									

Criteria	Commentary										
	Drill Hole	Drill Hole Number ASX Release Title ASX Release Date									
		011 – 021, 079		Significant	Increase in Karridal sit's Mineral Resou		28 January 2				
		009 – 076, 079 – 088	3, 091	High-Grade Gold Intersections from infill drilling at Karridale			30 October 2	2019			
	18KARC(066, 070,	18KARC006, 022,023, 063, 064, 066, 070, 071, 074, 075, 076, 078, 087, 089-093, 101, 102, 108 19KARC001 - 008 18KARC065, 068, 077, 080-085, 104-107, 117,119, 128 18KARC004,007-010			25% Increase in Karridale Gold Deposit's Mineral Resource More High Grade Intercepts at Laverton Gold Project Focus Advances its Karridale and Burtville Projects			9			
								9			
	1 1							2019			
	18KARC				Progress Update		31 July 2018	3			
	KARC129	9, 135		Maiden Mineral Resource for Karridale Deposit			23 February	2018			
	278, 279, KARD202	,	5,	Operationa			16 January 2	2018			
	KARD28 KARC28			Operationa Drilling Uni	•		25 July 2017	7			
				Programm			28 April 201	7			
	KARC194	4 – 201, 203 – 226, 2		and Lavert			25 January 2	2017			
	KARC16	9 – 193	- 1	Update	erals Ltd Exploration	1	28 April 2010	6			
	KARC156	5, 158, 160 - 168 6 - 157, 159 6, 717, 724, 725 - 72			Grows for Significan em at Karridale	27 January 2016					
	KARD154	KARD154			Karridale Exploration Update: Exciting Signs			5			
	KARC14	RC138 – 143 RC145 – 146 RC152 - 153		Laverton Exploration Update			30 January 2015				
	KARC123 KARC130	3 – 126		Quarterly Activities Report			30 October 2	2013			
	Collar deta	ils of 5 drill holes t	les that have not been previously reported a			ted a	re given belo	ow:			
	Hala ID	Easting		orthing			-I D 41- ()	Azim		Di- (0-II)	Tenement
	Hole ID 18KARC067	GDA94z51 466074.6		15277	RL 469.6	Tota	72 Tepth (m)	(Coll 148.7	ar)	Dip (Collar) -59.9	(Collar) M3800089
	18KARC072	466159.3		15432	471.2		78	151.1		-60.2	M3801281
	18KARC073	466139.7	68	15467	471.5		108	150		-60	M3801281
	18KARC086	466222.58		15479.6	471.27		96	151.2		-59.2	M3800073
	18KARC127	466209.9	681	15915.5	470.39		142	146.2	8	-49.66	M3800073
Data aggregation methods	Mineralised intersections are reported at a 0.5g/t Au cut-off with a minimum reporting width of 1m and up to 3m internal dilution. The length weighted average grades from diamond core can include measured intervals of core loss.										
Relationship between mineralization	Holes were drilled orthogonal to mineralisation as much as possible, however the exact relationship between intercept width and true width cannot be estimated exactly in all cases.										
widths and											
intercept lengths											
Diagrams	Accurate plans are included in this announcement. 3D perspective views and schematic cross- sections are included to illustrate						to illustrate				
Balanced	the distribu	ution of grade.									
reporting		ults are reported in the representative se				SX aı	nnouncemen	nt for HML I	noies sh	ows actual location	ons of holes
Other substantive exploration data	There is no other material exploration data to report at this time. There is no other material exploration data to report at this time.										
Further work	 FML anticipates additional drilling to follow up on encouraging results in Laverton. Focus have engaged RPMGlobal to conduct a PFS for Laverton Stage 1 mining 										

• Section 2 Details for the Burtville deposit from ASX Announcement "115% Increase to Burtville Mineral Resource" Dated 21/10/2020

Criteria	ary						
Mineral tenement	• The drilling was conducted on tenement M38/261 which is 100% owned by Focus Minerals (Laverton) Ltd						
and land tenure	he tenement is in good standing.						
status	The Burtville Deposit is covered by the 2019 Nyalpa-Pirniku Native Title Claim.						

Criteria	Commentary									
Exploration		Burtville was mined as part of the Burtville Mining Centre from the late 1890's until 1922 to a	depth of 20m -	- 40m below						
done by other	surface.									
parties	From the 19	torical drilling	was by SOG							
		 From the 1970's various companies have conducted exploration activities at Burtville. The bulk of the historical drilling was by SOG who open pit mined the deposit in the 1990's recovering 64,000 ounces @ 1.4g/t Au. 								
	Crescent Go	old and subsequently Focus conducted large scale deeper drilling programs before recomme	encing mining	in 2012 until						
	May-2013 that recovered 23,635 oz at 1.12 g/t Au.									
Geology		deposit lies within the Burtville Terrane of the Laverton Greenstone Belt.								
	Basal Basal	ts/Dolerite overlain by shales, sandstones and felsic/intermediate volcaniclastics have beer	n intruded by t	he Karridale						
	Gabbro and	Burtville Granodiorite. A swarm of brittle ductile shallow NNW dipping fault zones/shear	s over print t	ne package.						
		, a network of 200-400m spaced N-S and NNW striking cross faults extend between Burtville -								
	to Mt Lebar	non. These cross faults have been the historic focus of hundreds of shallow shafts/drive	es exploiting h	nigher grade						
		n. At Burtville a pervasive west dipping fabric hosts significant bulk mineralisation as a halo to hig	gher grade stee	p~N striking						
	and shallow I	NNW dipping mineralised structures.								
Drill hole	Historic drilling	ng information has been validated against publicly available WAMEX reports. Not all drill holes of	used in the est	mate have						
information		ed publicly. However, when Crescent Gold acquired the tenements a detailed review checking								
	those in the	database was conducted by an independent geologist. These drill holes occur mostly in the o	oxide layer tha	t has been						
	mined out. F	urthermore, just over 2/3rds of the drilling informing the remnant portion of the Burtville Mineral	Resource was	completed						
	by Crescent	and later Focus Minerals.								
			WAMEX Report A-	Donout						
	Company	Drill Hole Number	Number	Report Date						
	Aberfoyle	BTRC001, BTRC002, BTRC003, BTRC004, BTRC005, BTRC006, BTRC007, BTRC008, BTRC009,	27610	Feb-89						
		BTRC010, BTRC011, BTRC012, BTRC013, BTRC014, BTRC015, BTRC016, BTRC017, BTRC018, BTRC010, BTRC019, BTR								
		BTRC019, BTRC020, BTRC021, BTRC022, BTRC023, BTRC024, BTRC025, BTRC026, BTRC027, BTRC028, BTRC029, BTRC030, BTRC031, BTRC032, BTRC033, BTRC034, BTRC035, BTRC036,								
		BTRC037, BTRC038, BTRC040, BTRC041, BTRC043								
		BTRC062, BTRC063, BTRC064, BTRC065, BTRC066, BTRC067, BTRC068, BTRC069, BTRC070, BTRC074, BTR	31876	Sep-89						
		BTRC071, BTRC072, BTRC073, BTRC074, BTRC075, BTRC076, BTRC077, BTRC078, BTRC079, BTRC080, BTRC081, BTRC082, BTRC083, BTRC084, BTRC085, BTRC086, BTRC087, BTRC088,								
		BTRC089, BTRC090, BTRC091, BTRC092, BTRC093, BTRC094, BTRC095, BTRC096, BTRC097,								
		BTRC098, BTRC109, BTRC100, BTRC101, BTRC102, BTRC103, BTRC104, BTRC105, BTRC106, BTRC107, BTRC108, BTRC109, BTRC111, BTRC112, BTRC113, BTRC114, BTRC115, BTRC116,								
		BTRC117, BTRC118, BTRC119, BTRC120, BTRC121, BTRC122, BTRC123, BTRC124, BTRC125,								
		BTRC126, BTRC127, BTRC128								
		BTRC044, BTRC045, BTRC047, BTRC049, BTRC050, BTRC051, BTRC052, BTRC054, BTRC056, BTRC057, BTRC058, BTRC059	31884	May-89						
		BTRCDD039, BTRCDD042, BTRCDD046, BTRCDD048, BTRCDD053, BTRCDD061								
		BTRCDD143, BTRCDD144	31885	Dec-89						
	Gwalia	BTRC150, BTRC151, BTRC152, BTRC153, BTRC154, BTRC155, BTRC156, BTRC157, BTRC158, BTRC156, BTRC167, BTRC168, BTR	35752	Jun-91						
	Consolidated NL	BTRC160, BTRC161, BTRC162, BTRC163, BTRC164, BTRC165, BTRC166, BTRC167, BTRC168, BTRC169, BTRC170, BTRC171, BTRC172, BTRC173, BTRC174, BTRC175, BTRC176, BTRC177,								
		BTRC178, BTRC179, BTRC180, BTRC181								
	Sons of	BEC825, BEC826, BEC827, BEC828, BEC830	62685	Mar-01						
	Gwalia Crescent	BU011, BU012, BU013, BU014, BU016	70629	May-05						
	Gold	BURC001, BURC002, BURC003, BURC004, BURC005, BURC006, BURC007, BURC009, BURC011,	81631	Mar-09						
		BURCO12	00704	Man 11						
		BVRC001, BVRC002, BVRC003, BVRC004, BVRC007, BVRC008, BVRC009, BVRC010 BVRC018, BVRC019, BVRC020, BVRC021, BVRC022, BVRC023, BVRC024, BVRC025, BVRC026,	89791 94269	Mar-11 Mar-12						
		BVRC027, BVRC028, BVRC029, BVRC030, BVRC031, BVRC032, BVRC033, BVRC034, BVRC035,	01200	Wai 12						
		BVRC036, BVRC037, BVRC038, BVRC039, BVRC040, BVRC041, BVRC042, BVRC043, BVRC044, BVRC044, BVRC045, BVRC046, BVR								
		BVRC045, BVRC046, BVRC047, BVRC048, BVRC049, BVRC050, BVRC051, BVRC052, BVRC053, BVRC054, BVRC055, BVRC056, BVRC057, BVRC058, BVRC059, BVRC060, BVRC061, BVRC062,								
		BVRC063, BVRC064, BVRC065, BVRC066, BVRC067, BVRC046, BVRC047, BVRC048, BVRC049,								
		BVRC050, BVRC051, BVRC052, BVRC053, BVRC054, BVRC055, BVRC056, BVRC057, BVRC058, BVRC059, BVRC060, BVRC061, BVRC062, BVRC063, BVRC064, BVRC065, BVRC066, BVRC067,								
		BVRC068, BVRC069, BVRC070, BVRC071, BVRC072, BVRC073, BVRC074, BVRC075, BVRC076,								
		BVRC077, BVRC079, BVRC080, BVRC081, BVRC082, BVRC083, BVRC084, BVRC085, BVRC086,								
		BVRC087, BVRC088, BVRC089, BVRC090, BVRC091, BVRC093, BVRC094, BVRC095, BVRC096, BVRC097, BVRC098, BVRC099, BVRC100, BVRC101, BVRC102, BVRC103, BVRC104, BVRC105,								
		BVRC106, BVRC107, BVRC108, BVRC109, BVRC110, BVRC1111								
		BVRC112, BVRC113, BVRC114, BVRC115, BVRC116, BVRC117, BVRC118, BVRC119, BVRC120,	98692	Mar-13						
		BVRC121, BVRC122, BVRC123, BVRC124, BVRC125, BVRC126, BVRC127, BVRC128, BVRC129, BVRC124, BVRC124, BVRC124, BVRC126, BVRC127, BVRC128, BVRC127, BVRC128, BVR								
		BVRC130, BVRC131, BVRC132, BVRC133, BVRC134, BVRC135, BVRC136, BVRC137, BVRC138, BVRC139, BVRC140, BVRC142, BVRC143, BVRC144, BVRC145, BVRC146, BVRC147, BVRC148,								
		BVRC150, BVRC151, BVRC152, BVRC153, BVRC154, BVRC155, BVRC156, BVRC157, BVRC158,								
		BVRC159, BVRC160, BVRC161, BVRC162, BVRC163, BVRC164, BVRC165, BVRC166, BVRC167, BVRC168, BVRC169, BVRC170, BVRC171, BVRC172, BVRC173, BVRC174, BVRC174, BVRC175, BVRC176								
		BVRC168, BVRC169, BVRC170, BVRC171, BVRC172, BVRC173, BVRC174, BVRC175, BVRC176, BVRC177, BVRC178, BVRC179, BVRC180, BVRC181, BVRC182, BVRC183, BVRC184, BVRC185,								
		BVRC186, BVRC187, BVRC188, BVRC189, BVRC190, BVRC191, BVRC192, BVRC193, BVRC194,								
		BVRC195, BVRC196, BVRC197, BVRC198, BVRC199, BVRC200, BVRC201, BVRC202, BVRC203, BVRC204, BVRC205, BVRC206, BVRC207, BVRC208, BVRC209								
	Focus	BVRC210, BVRC211, BVRC212, BVRC213, BVRC214, BVRC215, BVRC216, BVRC217, BVRC218,	98692	Mar-13						
	Minerals	BVRC219, BVRC220, BVRC221, BVRC222, BVRC223, BVRC224, BVRC225, BVRC226, BVRC227,	-							
	Ltd	BVRC228, BVRC229, BVRC230, BVRC231, BVRC232, BVRC233, BVRC234, BVRC235, BVRC236,								

Criteria	Commentary										
			VRC238, BVR								
			VRC260, BVR								
			VRC271, BVR0 VRC281, BVR0								
			VRC290, BVR								
			VRC305, BVR								
			VRC314, BVR(VRC323, BVR(
			VRC332, BVR								
			VRC342, BVR								
			VRC351, BVR0 VRC361, BVR0								
			VRC370, BVR	,	,	,	,	,			
			VRC380, BVR								
			VRC391, BVR0 VRC401, BVR0								
			VRC410, BVR								
			VRC419, BVR								
			VRC428, BVR0 VRC437, BVR0								
			VRC446, BVR								
			VRC455, BVR								
			VRC464, BVR0 VRC473, BVR0								
			VRC482, BVR								
			VRC492, BVR								
			VRC527, BVR0 VRC538, BVR0	,	,	,	,	,			
			VRC553, BVR								
		BVRC581, B	VRC583, BVR	C584, BVRC5	86, BVRC	587, BVRC5	,	,			
			<u>VRC618, BVRC</u> VRC546, BVR(68 B\/DC56	0 B\/DC5	70 BVDC572	102458	Mar-14
			VRC540, BVRC							102430	IVIAI-14
		BVRC596, B	VRC601, BVR	C602, BVRC6	03, BVRC	604, BVRC6	06, BVRC61	4, BVRC6	23, BVRC624,		
			VRC628, BVR								
			VRC639, BVR0 VRC652, BVR0								
		BVRC674, B	VRC675, BVR0 VRC694, BVRC	C676, BVRC6	77, BVRC	678, BVRC6	89, BVRC69				
	The collar detai	ils of 14 holes of	drilled by Focu	ıs but not ex	ternally re	ported are g	given below.				
	Hole ID	Easting GDA94z51	Northing GDA94z51	RL	Depth (m)	Azimuth (Collar)	Dip (Collar)	Drill Type			
	BUDD0002 BUDD0004	465187.88 465221.28	6817865.2 6817863.6	437.502 436.882	75 93.5	273.8 92.8	-58.9 -60	DD DD			
	BVRC241	465517.03	6817777.8	476.071	125	256.1	-59.3	RC			
	BVRC242	465497.91	6817779.6	476.247	125	259.7	-60.7	RC			
	BVRC255	465419.79	6817980.2	477.876	125	264.6	-61.4	RC			
	BVRC256	465399.73	6817980.1	478.012	125	270.2	-61.8	RC			
	BVRC261 BVRC262	465364.04 465349.72	6817636.8 6817638.8	476.406 475.871	125 125	88.2 88.2	-60.3 -60.8	RC RC			
	BVRC282	465446.65	6817680	475.81	110	88.4	-87.6	RC			
	BVRC564	465365.01	6817891.6	427.378	54	90	-60	RC			
	BVRC608	465123.4	6817860	435.134	54	90	-60	RC			
	BVRC609	465144.2	6817861	436.363	48	90	-60	RC			
	BVRC610	465163.59	6817860	436.916	54	90	-60	RC			
	BVRC627	465355.01	6817891.7	426.907	54	90	-60	RC			
Data		dintersections								to 3m interna	al dilution. The
aggregation methods	length weig	hted average	grades from d	lamond core	can inclu	ide measure	ed intervals	of core lo	SS.		
Relationship		drilled orthog			much as p	oossible, ho	wever the e	exact relat	tionship betwe	een intercept	width and true
between	width canno	ot be estimated	d exactly in all	cases.							
mineralization											
widths and intercept											
lengths											
Diagrams	Accurate p	lans are includ	led in this and	nouncement	. 3D persi	pective view	s and sche	matic cro	ss- sections	are included to	o illustrate the
	distribution	of grade.									
Balanced reporting		ults are reporte									th only 12% of
reportina	I the remnan	nt resource utili	sıng SOG's dı	rilling condu	cted by the	e mining de	partment ar	nd not ext	ernally report	ed.	

Criteria	Commentary
Other substantive	There is no other material exploration data to report at this time.
exploration data	
Further work	Focus have engaged RPMGlobal to conduct a PFS for Laverton Stage 1 mining.

• Section 2 Details for the Beasley Creek deposit from ASX Announcement "Beasley Creek Mineral Resource Grows by 29%" Dated 20/08/2020

29%" L	Dated 20/08/2020							
Criteria	Commentary							
Mineral tenement	The drilling was conducted on tenements 100% owned by Focus Minerals (Laverton) Pty Ltd.							
and land tenure	 All tenements are 							
status	 The Beasley Creek mineral resource estimate is contained entirely within Mining Lease M38/049. The Nyalpa Pirniku claim has been lodged over the Laverton project areas. No claims have been determined at this time 							
Exploration		as formerly mined as an open pit to about 85m depth by WMC from 1						
done by other parties		has been performed by Metex/Delta Gold 1996/1997 and then Cresc		uction of 00.01	102.			
Geology	Creek SZ is deep saprolitic cla	hydrothermally brecciated sediments, conglomerates and minor black	n:	d McIntyre FZ'	s. The Beasle			
	laminated vebreccia veinCore loss typical	eins and,	ne drill bit. These ha	ard fragments r	otate with the			
Drill hole information	Company	Drill Hole Number		WAMEX Report A- Number	Report Date			
	Western Mining Corporation Ltd	BCP0002, BCP0003, BCP0004, BCP0005, BCP0007, BCP0008, BC BCP0012, BCP0013, BCP0014, BCP0021, BCP0022, BCP0023, BC BCP0026, BCP0033, BCP0034, BCD001		22647	1987			
		26696	1988					
		BCD008, BCD013, BCD018, BCD019, BCD020, BCD021, BCD023, BCD025, BCD026	31396	1989				
	Motov	BCP0328						
	Metex Resources NL	BCD028		48547	1996			
	Focus Minerals Ltd	18BSDD001, 18BSDD002, 18BSDD003, 18BSDD004, 18BSDD005 18BSDD007, 18BSDD008, 18BSDD009, 18BSDD010, 18BSDD012 18BSDD014, 18BSDD015, 18BSDD016, 18BSDD017, 18BSDD019 18BSRC001, 18BSRC002, 18BSRC003 18BSRD004, 18BSRD011, 19BSDD001, 19BSDD002, 19BSDD003, 19BSDD004, 19BSDD005 19BSRC001, 19BSRC002, 19BSRC003, 19BSRC004, 19BSRC006 19BSRC010, 19BSRC011, 19BSRC012, 19BSRD001, 19BSRD002 19BSRD005, 19BSRD006, 19BSRD007, 19BSRD008, 19BSRD010 19BSRD012, 19BSRD013, 19BSRD014, 19BSRD016, 19BSRD017 19BSRD019, 19BSRD022, 19BSRD023, 19BSRD026	, 18BSDD013 , 18BSDD020 18BSRD015 , 19BSDD006, , 19BSRC007, , 19BSRD004, , 19BSRD011,	120411	2019			
	Focus Minerals' drilled							
		ASX Ro Da						
	19BSDD009, 19BSDI 19BSDD017, 19BSDI 19BSDD024, 19BSDI 19BSDD030, 19BSDI 19BSDD037, 19BSDI 19BSRC015, 19BSRC	22/07/2019						

Criteria	Commentary					
	19BSRC035, 19BSRC040, 19BSRC043, 19BSRC044, 19BSRC045, 19BSRC053,					
	19BSRC054, 19BSRC055					
	19BSRD027, 19BSRD028, 19BSRD031, 19BSRD032, 19BSRD033, 19BSRD034					
	20BSDD027, 20BSDD030, 20BSDD032, 20BSDD038, 20BSDD050, 20BSDD051,	Laverton	28/07/2020			
	20BSDD052, 20BSDD054, 20BSDD055, 20BSDD061, 20BSDD063, 20BSDD065,	Exploration				
	20BSDD066	Update				
	20BSRC004, 20BSRC005					
	20BSRD012, 20BSRD013, 20BSRD014, 20BSRD015					
Data aggregation	 Mineralised intersections are reported at a 0.5g/t Au cut-off with up to 3m internal di 	lution. The length	weighted average grades from			
methods	diamond core can include measured intervals of core loss.					
Relationship	Wherever possible holes were drilled orthogonal to mineralisation					
between	Holes targeting the WNW extension McIntyre/BTW FZ structures and Shallow SE dip	oping footwall stru	ctures in the NW part of the			
mineralization	Beasley Creek Project often have sub-optimal orientations due to limited drilling co					
widths and	represented as true widths at this stage.					
intercept lengths	True widths can be estimated once geological/mineralisation modelling has been com.	pleted.				
	Furthermore, no intersections are represented as calculated true widths in this report					
Diagrams	Accurate plans are included in this announcement. 3D perspective views and scheme	natic cross- section	ns are included to illustrate the			
	distribution of grade					
Balanced	Historic drill results are available on WAMEX					
reporting	Drilling results are reported in a balanced reporting style. The ASX announcement for	r Focus Minerals I	noles shows actual locations of			
, ,	holes drilled, and representative sections as appropriate.					
Other substantive						
exploration data	There is no other material exploration data to report at this time.					
Further work	Focus Minerals anticipates additional drilling to follow up on encouraging results in Lar	verton.				

 Section 2 Details for the Beasley Creek South deposit from ASX Announcement "Beasley Creek South Delivers High Grade Mineral Resource" Dated 15/07/2020

Criteria	Commentary								
Mineral tenement and land tenure status	The drilling was conducted or All tenements are in good sta The Beasley Creek South min The Nyalpa Pirniku claim has	neral resource estimate is contain been lodged over the Laverton p	ed entirely within Mining Lease M roject areas. No claims have bee	n determined at th					
Exploration done by other parties	 Beasley Creek South has been drilled by numerous companies over the years, mainly WMC who mined the adjacent Beasley Creek open pit, Metex Resources and Crescent Gold NL. Drill spacing on the main shear approached 20m x 20m and was useful for guiding follow up drill depths. However, due to RC sample issues within the main shear none of these holes were used in this resource estimate. 								
Geology	 Mineralisation at Beasley South is located on the moderately east dipping Beasley Shear Zone (SZ). To date mineralisation is confirmed at Beasley South over 500m strike and to within 400m of the southern side of Beasley Creek. The Beasley SZ is deeply weathered to ~80-100% clay and drill intersections to date at 130m depth are located in completely weathered rock. The Beasley SZ is sandwiched between hanging-wall (eastern) mafic high magnesium volcanics and footwall (western) ultramafic intrusions and feldspar-hornblende porphyries. The weathered rocks within the Beasley SZ include: saprolitic clays, saprock of hydrothermally brecciated sediments, conglomerates and minor black shale, iron stone after gossan, laminated veins and, breccia vein infill. Core loss typically occurs when quartz breccia fragments become partially lodged in the drill bit. These hard fragments rotate with the bit causing grinding/washing of the soft highly oxidised shear matrix. Due to the soft nature of the oxidised shear RC sample recovery has proven to be elusive and regularly is less than 40% 								
Drill hole Information	within mineralised Beasley (Company Focus Minerals Ltd	Drill Hole Number 18BSRC009, 18BSRC010	WAMEX Report A- Number	er F	Report Date 2018				
	19BSDD044, 19BSDD045, 19BS 19BSDD061, 19BSDD062, 19BS 19BSDD068, 19BSDD069, 19BS 19BSDD076, 19BSDD077, 19BS 19BSDD085, 19BSDD086, 19BS	0, 19BSDD058, 19BSDD060, 1, 19BSDD066, 19BSDD067, 18, 19BSDD074, 19BSDD075, 2, 19BSDD083, 19BSDD084,	ASX Release Title Dutstanding Results at Beasley Creek South	ASX Release Date 30/01/2020					
	20BSDD001, 20BSDD002, 20BSI 20BSDD011, 20BSDD012, 20BSI 20BSDD018	Strong Hits at Beasley Creek South Boost	28/04/2020						

distribution of grade. Balanced	Criteria	Commentary								
Collar details of FML holes drilled during 2020 and yet to be released are given below:							Lav	verton		
BHID							Re	source Upside		
BHD							<u>'</u>	•		
2685DD0201		Collar details of FI	ML holes drilled du	ıring 2020 and yet	to be released	are given below:				
2085DO021		BHID	EAST	NORTH	RL	AZIMUTH	DIP	Drill Type		
2085D0022		20BSDD020		6837783.9		270	-60	DD		
288SDD023					432.5		-60		DD	
208SD0026		20BSDD022	433897.77	6838100.1	431.8	270	-60	61.8	DD	
2005D0026		20BSDD023		6838038.9	431.9		-60		DD	
Decision Column		20BSDD024								
208500029		20BSDD025	433966.06	6837910.5		270	-60	105	DD	
208500031					432.1	270	-60	98	DD	
208SD0033		20BSDD029	434015.9			270	-60	128	DD	
2085D0034		20BSDD031		6837876.2			-60	136.1	DD	
208SD0035		20BSDD033	434001.31	6838049.5		270	-60	124.9	DD	
208SD0036										
208SDD037									DD	
208SDD039					433.8	270	-60		DD	
208SDD040		20BSDD037				270	-60	156.4	DD	
208SDD041		20BSDD039	433966.44	6837982.7	431.8	270	-60	107	DD	
208SDD042		20BSDD040				270	-60	165.3	DD	
20BSDD043		20BSDD041	434004.72		432.8	270			DD	
20BSDD044		20BSDD042	433936.7	6837958.6	431.7	270	-60	98.1	DD	
20BSDD045		20BSDD043	433981.66	6837895.8	432.1	270	-60	115.9	DD	
208SDD046		20BSDD044	433914.19	6838045.6	431.8	270	-60	64.8	DD	
20BSDD048		20BSDD045	433965.15	6837962.3	431.7	270	-60	107	DD	
208SD049		20BSDD046	433896.06	6838073.0	431.8	270	-60	46.9	DD	
20BSDD053		20BSDD048	433919.98	6838100.0	431.8	270	-60	52.9	DD	
208SDD056		20BSDD049	434019.65	6838171.8	431.9	270	-60	128	DD	
20BSDD057		20BSDD053			433.4	270	-80		DD	
20BSDD058 434116.06 6837789.8 431.3 270 -60 238.9 DD 20BSDD064 433958.33 6838160.4 430.8 260 -60 65 DD 20BSRC002 433907.3 6838129.7 431.7 269.0 -60 30 RC 20BSRD004 434111.36 6837890.4 432.5 272.1 -60 224 RC/DD 20BSRD006 434084.52 6838114.7 432.5 267.8 -60 195.5 RC/DD 20BSRD001 434092.46 6838078.7 432.4 269.4 -60 198.5 RC/DD 20BSRD011 434090.95 683795.4 432.1 269.3 -60 207.4 RC/DD 20BSRD011 434090.95 683795.4 432.1 269.3 -60 207.4 RC/DD 20BSRD011 434090.95 683795.4 432.1 269.3 -60 207.4 RC/DD 20BSRD011 434090.95 6837965.4 432.1 269.3 -60 207.4 RC/DD 20BSRD011 434090.95 6838078.7 432.4 269.4 -60 198.5 RC/DD 20BSRD011 434090.95 6837965.4 432.1 269.3 -60 207.4 RC/DD 20BSRD011 434090.95 6837965.4 432.1 269.3 -60 207.4 RC/DD 20BSRD011 434090.95 6838078.7 432.4 269.4 -60 198.5 RC/DD 20BSRD011 43409.95 6838078.7 432.4 269.4 -60 198.5 RC/DD 20BSRD011 43409.95 6838078.7 432.4 269.4 -60 198.5 RC/DD 20BSRD011 43409.95 6838078.7 432.4 269.4 -6		20BSDD056		6837841.5						
20BSDD064 433958.33 6838160.4 430.8 260 -60 65 DD 20BSRC002 433907.3 6838129.7 431.7 269.0 -60 30 RC 20BSRD004 434111.36 6837890.4 432.5 272.1 -60 224 RC/DD 20BSRD006 434084.52 6838114.7 432.5 267.8 -60 195.5 RC/DD 20BSRD009 434110.45 6838035.1 432.3 271.9 -60 222.4 RC/DD 20BSRD010 434092.46 6838078.7 432.4 269.4 -60 198.5 RC/DD 20BSRD011 434090.95 6837965.4 432.1 269.3 -60 207.4 RC/DD 20BSRD011 434090.95 6837965.4 432.1 269.3 -60 207.4 RC/DD 20BSRD011 434090.95 6837965.4 432.1 269.3 -60 207.4 RC/DD 20BSRD011 434090.95 6837965.4 432.1 begains a compact of the second of		20BSDD057	433956.02	6837837.2	433.3	265	-60	107	DD	
20BSRC002 433907.3 6838129.7 431.7 269.0 -60 30 RC 20BSRD004 434111.36 6837890.4 432.5 272.1 -60 224 RC/DD 20BSRD006 434084.52 6838114.7 432.5 267.8 -60 195.5 RC/DD 20BSRD009 434110.45 6838035.1 432.3 271.9 -60 222.4 RC/DD 20BSRD010 434092.46 6838078.7 432.4 269.4 -60 198.5 RC/DD 20BSRD011 434090.95 6837965.4 432.1 269.3 -60 207.4 RC/DD 20BSRD011 434090.95 6837965.4 432.1 269.3 -60 207.4 RC/DD **Mineralised intersections are reported at a 0.5g/t Au cut-off with a minimum reporting width of 1m and up to 3m internal dilution.** length weighted average grades from diamond core can include measured intervals of core loss. **Wherever possible holes were drilled orthogonal to mineralisation widths and intercept lengths* **Diagrams** **Accurate plans are included in this announcement. 3D perspective views and schematic cross- sections are included to illustrate distribution of grade. **Accurate plans are included in this announcement. 3D perspective views and schematic cross- sections are included to illustrate distribution of grade. **Drilling results are reported in a balanced reporting style. The ASX announcement for FML holes shows actual locations of holes dril and representative sections as appropriate. **Other substantive** **There is no other material exploration data to report at this time.		20BSDD058		6837789.8						
20BSRD004 434111.36 6837890.4 432.5 272.1 -60 224 RC/DD		20BSDD064	433958.33		430.8	260	-60	65	DD	
20BSRD006					431.7		-60	30		
20BSRD009 434110.45 6838035.1 432.3 271.9 -60 222.4 RC/DD 20BSRD010 434092.46 6838078.7 432.4 269.4 -60 198.5 RC/DD 20BSRD011 434090.95 6837965.4 432.1 269.3 -60 207.4 RC/DD 20BSRD011 434090.95 683796.4 432.1 269.3 -60 207.4 RC/DD 20BSRD011 434090.95 683796.4 432.1 269.3 -60 207.4 RC/DD 20BSRD011 434090.95 683796. 440.1 400.1 400.1 400.1 400.1 400.1 400.1 400.1 400.1 400.1 400.1 400.1 400						272.1	-60	224		
20BSRD010 434092.46 6838078.7 432.4 269.4 -60 198.5 RC/DD 20BSRD011 434090.95 6837965.4 432.1 269.3 -60 207.4 RC/DD Data aggregation methods Relationship between mineralization widths and intercept lengths Diagrams • Accurate plans are included in this announcement. 3D perspective views and schematic cross- sections are included to illustrate distribution of grade. Balanced reporting Other substantive • There is no other material exploration data to report at this time.										
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Diagrams Accurate plans are included in this announcement. 3D perspective views and schematic cross- sections are included to illustrate distribution of grade. Balanced reporting Diagrams Diag	widths and									
Diagrams Accurate plans are included in this announcement. 3D perspective views and schematic cross- sections are included to illustrate distribution of grade. Balanced reporting Diagrams Diag	intercept lengths									
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reporting and representative sections as appropriate. Other substantive • There is no other material exploration data to report at this time.	Balanced		0	balanced reportin	a style. The AS	X announcement for	or FML holes s	hows actual loca	tions of holes drilled	
Other substantive • There is no other material exploration data to report at this time.										
באטוטומנוטוו עמנג					ort at this time.					
Further work • FML anticipates additional drilling to follow up on encouraging results in Laverton. • Focus have engaged RPMGlobal to conduct a PFS for Laverton Stage 1 mining										

 Section 2 Details for the Wedge deposit from ASX Announcement "Wedge Open Pit Resource Update" Dated 24/01/2020

Criteria	Commentary
Mineral tenement	· All exploration was conducted on tenements 100% owned by FML or its subsidiary companies Focus Operations Pty Ltd. All tenemen
and land tenure	are in good standing.
status	Various royalties may be in place as documented in the FML Annual Report 2016

Criteria	Co	mmentary
	•	The Nyalpa Pirniku claim has been lodged over the Laverton project areas. No claims have been determined at this time
	•	The tenements fall within the Laverton Water Reserve and all exploration completed complied with required regulations.
Exploration	•	The Wedge deposit has been historically mined as 3 pits by Ashton Gold (WA) Ltd between 1990 and 1992. Production figures state
done by other		262,023t @ 2.53g/t Au HG ore was mined from the pits and 260,544t @ 2.51 g/t of HG ore was Milled.
parties	•	Ashton Gold Mines Pty Ltd formerly Hillmin Gold Mines Pty Ltd conducted various exploration activities over the Wedge trend since 1984 when it gained 100% management and operation of Teck Explorations and Morrison Petroleum's JV interests. This involved geological mapping, ground magnetic surveys, soil sampling, aeromagnetics, resistivity, gradient array, induced polarisation, rock chip sampling, RC, Rotary Air Blast (RAB) and Diamond drilling.
	•	Metex acquired the Wedge tenements from Ashton Gold (WA) Ltd in September 1996, conducting various exploration activities including data validation, geological mapping, aerial photography, soil sampling, rock chip sampling, aeromagnetic surveys, RAB, Vacuum and RC drilling.
	•	The ground was subsequently acquired by Crescent Gold NL in May 2010 before being taken over by Focus Minerals Laverton in October 2012.
Geology	•	Regionally the geology comprises strongly deformed ultramafics, mafic volcanics and intercalated iron formation and sediments.
	•	The deposit is hosted by an interflow sedimentary unit within a thick Archean mafic volcanic pile. The interflow sediments consist of chert, shale and minor black shale below the oxidation horizon and contain pyrite and minor pyrrhotite.
Drill hole information	•	Historic drilling information has been validated against publicly available WAMEX reports. Not all drill holes can be found referenced in the WAMEX reports. However, cross-checking of original drill surveys was verified against the database. Most of these holes were drilled in the excavated pit area and has been depleted from the reported resource.

WAMEX Reference:

Company	Drill Hole Number	WAMEX Report A- Number	WAMEX Report Date
	LNP027 - LNP032, LNP034, LNP040 - LNP044, LNP047 - LNP050, LNP052, LNP053, LNP055, LNP056	16888	February 1985
	LNP001, LNP002, LNP007, LNP008, LNP013 - LNP015, LNP020, LNP024 - LNP026, LNP057 - LNP077, LNP083 - LNP093, LNP095, LNP096, LNP101, LNP102	Unknown	
Hillmin Gold	LNP104 - LNP123, LNP129 - LNP135, LNP138, LNP139 - LNP143	20646	February 1987
Mines Pty Ltd	LNP144 - LNP161, LNP163 - LNP215, LNP217 - LNP236, LNP238 - LNP241, LNP243, LNP245 - LNP268, LNP270, LNP271, LNP273, LNP274, LNP276, LNP278 - LNP287, LNP289, LNP291, LNP293, LNP295, LNP298 - LNP328, LNP330	23398	February 1988
	LND001 - LND009	27633	February 1989
	LNP331 - LNP347, LNP349, LNP351 - LNP357		
Ashton Gold	LND010 LNP359 - LNP361, LNP365 - LNP385	15929	January 1990
Mines Pty Ltd	LNP386 - LNP401, LNP403 - LNP406	33668	March 1991
	LNP411 - LNP418, LNP421, LNP424 - LNP432	35688	January 1992
Metex Resources NL	LNRC001, LNRC002, LNRC007 - LNRC010	48547	January 1996
Western Mining Corporation Ltd	LFP0817	22647	January 1988

FML holes WAMEX reference:

		WAMEX Report	
Company	Drill Hole Number	A- Number	WAMEX Report Date
Focus Minerals	18LNRC001, 18LNRC002, 18LNRC003, 18LNRC004,	120411	July 2019
Ltd	18LNRC005, 18LNRC006, 18LNRC007, 18LNRC008,		
Liu	18LNRC010, 18LNRC011, 18LNRC012, 18LNRC017,		
	18LNRC018, 18LNRC019, 18LNRC020, 18WDRC001,		
	18WDRC002, 18WDRC003, 18WDRC004, 18WDRC005,		
	18WDRC006, 18WDRC007, 18WDRC008, 18WDRC009,		
	18WDRC010, 18WDRC011, 18WDRC012, 18WDRC013,		
	18WDRC014, 18WDRC015, 18WDRC016, 18WDRC017,		
	18WDRC018, 18WDRC019, 18WDRC020, 18WDRC021,		
	18WDRC022, 18WDRC023, 18WDRC024, 18WDRC025,		
	18WDRC026, 18WDRC027, 18WDRC028, 18WDRC029,		
	18WDRC030, 18WDRC031, 18WDRC032,		
	18WDRC033,18WDRC034, 18WDRC035, 18WDRC036,		
	18WDRC037, 18WDRC038, 18WDRC039, 18WDRC040,		
	18WDRC041, 18WDRC042, 18WDRC043, 18WDRC044,		
	18WDRC045,		
	18WDRC047		

Criteria	FML Drilled holes not yet available on WAMEX									
			ole Number		ASX Release Title					
		9LNRC045, 19W 19WDRC024, 19' 19WDRC055		High Value	e Exploration Resu Gold Projec		22-Jul-19			
	19LNRC069 - 1 19LNRC079, 19	9LNRC0061, 19L 9LNRC070, 19Li 9LNRC089, 19LN 9WDDD002, 19V	NRC074 - IRC092	Wed	ge Open Pit Resoi	urce Update	24-Jan-20			
	Collar details of F	ML holes drilled	during 2019 are	given below: MGA 94 Zone	51		Depth	Tenement		
	Hole ID	Easting	Northing	RL	Azimuth	Dip	(m)	renement		
	19LNRC063	440997.69	6844326.7	457.35	304.64	-60.14	60	M3800159		
	19LNRC064	440918.51	6844228.1	456.83	309.92	-52.57	96	M3800159		
	19LNRC066	440974.68	6844293.2	457.27	281.26	-49.1	78	M3800159		
	19LNRC067	440919.89	6844178.6	456.47	313.09	-60.66	90	M3800159		
	19LNRC071	440942.06	6844265.3	456.88	305.64	-70.11	54	M3800159		
	19LNRC085	441026.11	6844433.2	457.31	299.9	-60.28	30	M3800159		
	19LNRC086	441010.67	6844398.9	457.61	306.57 302.52	-60.32	30	M3800159		
	19LNRC087	441003.76	6844389.1	457.53		-60.34	30	M3800159		
	19WDRC015	5 440377.35 6843239.7		455.71	320.49	-50.9	54	M3800159		
	19WDRC025	440374.13	6843274.5	455.68	310.13	-55.75	36	M3800159		
	19WDRC027	440391.37	6843288.2	455.73	321.87	-54.96	54	M3800159		
	19WDRC056	439873.36	6842975.9	453.45	323.79	-50.76	30	M3800159		
	19WDRC057	439861.23	6842964.1	453.52	324.25	-50.21	30	M3800159		
	19WDRC058	439829.81	6842931.3	453.78	319.51	-59.47	30	M3800159		
Data aggregation methods Relationship	 A statistical were applie applied to h 	review of the difed on a lode-by-louigher-grade outli	ferent mineralisa ode basis. A max er samples.	tion lodes revea imum top-cut of	25g/t was applied	de outliers to the s I to one high grade	e lode, on average	and various top cuts a 10g/t top-cap was ercept width and true		
between mineralization widths and intercept lengths			ar to militeralisati exactly in all case		possible, floweve	THE EXACT TELATION	isiip between iiti	ercept width and tide		
Diagrams			in body of the re							
Balanced			ilable on WAME							
reporting			ole in the previou							
Other substantive exploration data	There is no	other material e	xploration data to	report at this til	ne.					
Further work			ewing the explora							
	 Focus have 	engaged RPMG	Blobal to conduct	a PFS for Lave	rton Stage 1 minin	ıg				

Section 3 Estimation and Reporting of Mineral Resources

Criteria

Commentary

(Criteria listed in section 1, and where relevant in section 2, also apply to this section)

Section 3 Details for the Karridale Deposit from ASX Announcement "Karridale Mineral Resource increases by 60%"
 Dated 24/09/2020

Criteria	Commentary
Database integrity	 Data was geologically logged electronically; collar and downhole surveys were also received electronically as was the laborator analysis results. These electronic files were loaded into an acQuire database by the company in-house Database Administrator. Dat was routinely extracted to Microsoft Access during the drilling program for validation by the geologist in charge of the project. FML's database is a Microsoft SQL Server database (acQuire), which is case sensitive, relational and normalised to the Third Norma Form. Because of normalisation, the following data integrity categories exist: Entity Integrity: no duplicate rows in a table, eliminated redundancy and chance of error. Domain Integrity: Enforces valid entries for a given column by restricting the type, the format or a range of values. Referential Integrity: Rows cannot be deleted which are used by other records. User-Defined Integrity: business rules enforced by acQuire and validation codes set up by FML. Additionally, in-house validation scripts are routinely run in acQuire on FML's database and they include the following checks:

Criteria	Commentary
	Missing collar information
	 Missing logging, sampling, downhole survey data and hole diameter Overlapping intervals in geological logging, sampling, down hole surveys
	Checks for character data in numeric fields
	Data extracted from the database were validated visually in GEOVIA Surpac software and ARANZ Geo Leapfrog software. Also, when
	loading the data any errors regarding missing values and overlaps are highlighted.
Site visits	 Alex Aaltonen, the Competent Person for Sections 1 and 2 of Table 1 is FML's General Manager - Exploration and conducts regular site visits.
	 Hannah Kosovich, the Competent Person for Section 3 of Table 1 is FML's Resource Geologist and last visited site in September 2019.
Geological	All Focus drill holes, and historic mining data was used to guide the geological interpretation of the mineralisation specifically adhering
interpretation	to geological and structural controls.
	Relogging of Focus diamond core and RC chips was completed to standardise and provide a coherent data set.
	The relogging and additional drilling improved the understanding of geological controls on gold mineralisation at Karridale. The Karridale mineralisation is hosted in an interpreted half graben on the SE side of a large Granodiorite intrusion. The mineralisation is hosted
	primarily by the shallow NW dipping shears depicted by mylonitic sediment packages with intense carb-sericite alteration and by some
	NW-SE subvertical veins.
	The logging of sheared to mylonitic zones, quartz veining and/or carbonate-sericitic alteration guided the primary interpretation so that
	it was not solely controlled by mineralisation.
	• The mineralised geological interpretation was completed using Seequent Leapfrog software on a section-by-section basis. An
	 approximate 0.5g/t Au value was used to guide the interpretation. Minor deviation only of the lode geometry was noticed between drill holes along strike and down-dip.
	A number of steeply dipping NW striking cross fault features were identified and modelled. An apparent increase in grade was noted
	at the intersections of these cross faults and the shallow NW dipping lodes. The contacts of these intersections were considered a
	dilatational contacts with sharing of grades along the contact. Although in the flatter structures a grade dependent search was used to
	limit the influence of the high grades.
Dimensions	• Mineralisation extends over a 900m strike length trending NE and has been modelled from surface to a depth of 450m below surface.
	Numerous lodes have been modelled plunging 20 - 30° to the NW. Six cross-cutting faults plunging 55° to NNW and 30° to the NNE have also been interpreted. The thickness of the individual quartz veins varies from 0.25m to 6m thick. Average thickness of mineralised
	shears is 4m. In addition, an average 2m thick sub-horizontal supergene cover lode has been modelled covering most of the mineralised
	deposit area.
Estimation and	• Only RC and Diamond holes drilled by FML were used in the estimation. In total 301 holes were used, 271 RC holes for 53,270m and
modelling	30 RC pre-collar with diamond tail (RC/DD) holes for 10,934.53m.
techniques	The drill hole samples were composited to 1m within each domain, the dominant sampling interval. With a minimum 0.2m composite Composite Com
	length, intervals less than this were added to end of previous composite interval. Composited assay values of each lode were exported as text file (.csv) from Leapfrog and imported into Snowden Supervisor for
	statistical and geostatistical analysis.
	A review of histograms, probability plots and mean/variance plots for each domain revealed some outlier sample values.
	Top capping of higher Au values within each domain was carried out with Au values above the cut- off grade reset to the cut- off grade.
	Different caps were used for the lodes, an average of 10g/t Au was used; the largest cap was 30g/t Au in the cross-cutting HG fault
	lodes.
	 Variograms were modelled in Supervisor for lodes with greater than 200 samples, which was 13 lodes. Lodes with fewer than 200 samples shared the variogram of a similar orientated lode. A normal scores transformation was applied to the negatively skewed data
	in each lode. A back- transformation to original units was applied to the variogram models before being exported in Surpac readable
	format.
	GEOVIA Surpac Software was used for the estimation. An Ordinary Kriging (OK) technique was selected using the variograms modelled
	in Supervisor. Each domain was estimated separately. After a review of the geology and contact analysis in Supervisor software, it was
	considered acceptable for samples along the contact of the cross faults and flat lodes to be shared with limiting grade searches
	restricting the distance the higher grades were spread into the flat lodes. • A minimum of 8 and a maximum 14 - 16 samples were used to estimate each block with a maximum of 6 samples per drill hole. selected
	based on a Kriging Neighbourhood analysis in Supervisor.
	An elliptical search was used based on range and rotation directions of the Variograms.
	• If a block was not estimated with the initial search parameters, the minimum number of samples was reduced to 4 and the search
	distance increased by 1.5 times, with the maximum number of samples per hole reduced to 3. After the second search pass, a third
	pass was run on un-estimated blocks, increasing the search distance twice that of the second pass. After the third pass a few blocks
	in two lodes that had not estimated were assigned the average grade of the surrounding estimated blocks. • The block model had 54% blocks estimate in first search pass, 38% in the second search pass and 8% in the third search pass.
	Block sizes for the model were 20m in Y, 20m in X and 5m in Z direction. Sub celling of the parent blocks was permitted to 5m in the
	Y direction, 2.5m in the X direction and 1.25m in the Z direction. Sub-blocking was used to best fill the wireframes and inherit the
	grade of the parent block. No rotation was applied to the orientation of the blocks.
	Block size is approximately ½ of the average drill hole spacing.
	The estimate was validated by several methods. An initial visual review was done by comparing estimated blocks and raw drill holes.
	Tonnage weighted mean grades were compared for all lodes with the raw and top-capped drill hole values. There were no major
	differences.
	Swath plots of drill hole values and estimated Au grades by northing, easting and RL for the larger lodes were run in Supervisor and about the satisfactory and the transfer of the drillian date.
	showed that the estimated grades honoured the trend of the drilling data.

Criteria	Commentary
Moisture	Tonnages are estimated on a dry basis
Cut-off	• The Resources for Karridale have been reported above a 0.6g/t Au cut-off and above the 230mRL (235m below surface) for open pit
parameters	based on previous pit optimisations.
Mining factors or	The Karridale deposit would be mined by open pit extraction.
assumptions	
Metallurgical	While no metallurgical test work has been carried out specifically at Karridale, previous production and processing records for the
factors or	nearby Burtville Pit exist.
assumptions	
Environmental factors or	Karridale deposit sits near the previously mined Burtville Pit, with numerous historic workings in the area, including minor underground
assumptions	development at Boomerang.
Bulk density	Density values were assigned based on a modelled regolith category. The densities for each weathering category were calculated
	using a combination of physical bulk density and specific gravity measurements obtained from Focus diamond core.
	 A value of 1.94 was assigned to completely oxidised, 2.12 for completely weathered, 2.30 for strongly weathered, 2.53 for moderately
	weathered, 2.72 for partially weathered and 2.86 for fresh.
	 In total 512 specific gravity and bulk density measurements were used to determine the assigned densities.
	Jinning Testing and Inspections completed the bulk density measurements.
	• The water immersion technique was used for the specific gravity determinations on selected competent lengths of core greater than
	10cm.
Classification	Resources have been classified as Indicated and Inferred based primarily on drilling spacing and geological confidence in the geometry
	and continuity of the lodes. In addition, various estimation output parameters such as number of samples, search pass, kriging variance,
	and slope of regression have been used to assist in classification.
	• Shapes were created in Surpac to constrain the model within 40m x 40m spacing has been classified as Indicated and the surrounding
	40m x 80m spaced drilling for Inferred Resource down to the 230mRL
Audits or reviews	No external audits of the mineral resource have been conducted.
Discussion of	This is addressed in the relevant paragraph on Classification above.
relative	The Mineral Resource relates to global tonnage and grade estimates.
accuracy/	
confidence	

• Section 3 Details for the Burtville deposit from ASX Announcement "115% Increase to Burtville Mineral Resource" Dated 21/10/2020

Criteria	Commentary
Database	• Data was geologically logged electronically; collar and downhole surveys were also received electronically as was the laboratory analysis
integrity	results. These electronic files were loaded into an acQuire database by the company in-house Database Administrator. Data was
	routinely extracted to Microsoft Access during the drilling program for validation by the geologist in charge of the project.
	FML's database is a Microsoft SQL Server database (acQuire), which is case sensitive, relational and normalised to the Third Normal
	Form. Because of normalisation, the following data integrity categories exist:
	Entity Integrity: no duplicate rows in a table, eliminated redundancy and chance of error.
	Domain Integrity: Enforces valid entries for a given column by restricting the type, the format or a range of values.
	Referential Integrity: Rows cannot be deleted which are used by other records.
	User-Defined Integrity: business rules enforced by acQuire and validation codes set up by FML.
	Additionally, in-house validation scripts are routinely run in acQuire on FML's database and they include the following checks:
	Missing collar information
	Missing logging, sampling, downhole survey data and hole diameter
	Overlapping intervals in geological logging, sampling, down hole surveys
	Checks for character data in numeric fields Data extracted from the database were validated visually in GEOVIA Surgac software. ARANZ Geo Leapfron software and Datamine.
	But distributed from the dutabase were varieties in Sec. 11. Carpai servaria, 74.4.4.2 God Estaphog servaria and Butanimo
Site visits	software. Also, when loading the data, any errors regarding missing values and overlaps are highlighted. • Alex Aaltonen, the Competent Person for Sections 1 and 2 of Table 1 is FML's General Manager - Exploration and conducts regular site
one visits	visits.
	Michael Job, the Competent Person for Section 3 of Table 1 has not visited site.
Geological	All Focus and previous operators (Aberfoyle, Sons of Gwalia and Crescent Gold) RC and diamond drill holes and historic mining data
interpretation	was used to guide the geological interpretation of the mineralisation.
,	The gold mineralisation at Burtville is complex and is hosted within a granodiorite intrusive as well as via an extensive network of
	structurally controlled quartz veins. The stockwork of narrow quartz veins (1mm to 30cm) which cut the granodiorite, overlying sandstone
	and mafic units hosts a higher grade of gold compared with the alteration mineralisation seen in the surrounding granodiorite.
	A geological matrix analysis was conducted to determine what geological characteristics are important to assist in understanding the
	gold mineralisation. At Burtville, this study was inconclusive, with significant Au mineralisation in all rock types/altered zones except for
	the mafic volcanics.
	• Deterministic grade-based wireframes (as used in previous estimates) and running an estimate using linear methods (such as ordinary
	kriging (OK) or inverse distance (ID)) is difficult and not representative of the mineralisation. In particular, trying to tie together mineralised
	trends in such a structurally complex deposit is challenging.
	Therefore, the economic compositing function in Leapfrog software was used for the interpretation of the mineralised zone - at a cut-off
	of 0.05ppm Au, the minimum ore composite length was set to 5m, with maximum included and consecutive internal waste parameters
	set to 4m.

Criteria	Co	mmentary
	•	An intrusive geological model was constructed in Leapfrog. In the weathered zone (above the base of complete oxidation, which varies
		from 20 m to 50 m below topographic surface), a horizontal global trend was set, and used for interpolation of the geological model. In
		the transitional and fresh rock zone, a global trend of 25° towards grid west was set, which is concordant with the Au mineralisation
		trend.
	•	The geological model was designed to essentially exclude waste material and were to be used to constrain a non-linear estimation
Dii		method.
Dimensions	•	The deposit extends over a strike length of 700 mN, is about 800 mE wide and extends to 140 m below the surface. The mineralisation
Estimation and	_	is mainly around the granodiorite contact, which limits the known depth extent.
modelling	•	Estimation of the mineral resource was by the non-linear method Localized Uniform Conditioning (LUC) using Isatis software. Test work of the other major non-linear estimation method (Multiple Indicator Kriging) were not successful, as the indicator variograms above even
techniques		low thresholds were essentially nugget effect.
,	•	The LUC estimation process was as follows:
		Drill hole data selected within mineralized domains and composited to 2m downhole intervals in Datamine software – 2m was chosen
		as the best compromise between detailed information and over-smoothing using longer composites.
	•	Composited data imported into Isatis software for statistical and geostatistical analysis.
	•	Variography was done on data transformed to normal scores, and the variogram models were back transformed to original units. The
		Gaussian anamorphosis used for the normal scores transform was also subsequently used for the discrete Gaussian change of support
		model required for Uniform Conditioning. Variography was performed for separate oxidized and transitional/fresh rock mineralized
		domains.
	•	The variogram models had very high nugget effects (~80% of total sill), with a range of 200 m in fresh rock and 35 m in oxidised.
	•	Estimation (via Ordinary Kriging) was into block model that was a non-rotated model in MGA94 grid, with a panel block size of 20 mE x
		20 mN x 5 mRL – this is about the average drill spacing in the deposit. Localization of the grades was later into Selective Mining Units
		(SMU) block of 5 mE x 10 mN x 2.5 mRL (16 SMUs per panel).
	•	A 'distance limited threshold' technique was used where uncapped data was used within 5 m of the extreme values, but a capping of 10
		ppm was used beyond this This cap was based on inflections and discontinuities in the histograms and log-probability plots.
	ľ	The ellipsoid search parameters were based on the variogram ranges, with the search ellipse dimensions about 90% of the variogram range, with anisotropies retained. A minimum of 10 and maximum of 60 (2m composite) samples per panel estimate.
	•	If a panel was not estimated with these search parameters, then the ellipse was expanded by a factor of four, but less than 2% of the
		panels required this second pass.
	•	The UC process applies a Change of Support correction (discrete Gaussian model) based on the composite sample distribution and
		variogram model, conditioned to the Panel grade estimate, to predict the likely grade tonnage distribution at the SMU selectivity.
	•	The Localizing step was then run, and the resulting SMU models for the fresh and oxidised material were exported from Isatis to Datamine
	•	Estimates of Au grades were validated against the composited drill hole data by extensive visual checking in cross-section, plan and on
		screen in 3D, by global (per shoot) comparisons of input data and model, and by semi-local statistical methods (swath plots). All methods
		showed satisfactory results.
Moisture	•	Tonnages are estimated on a dry basis.
Cut-off	•	The cut-off grade of 0.6ppm Au was established for the nearby Beasley Creek pit optimisation work. Given that the mining and processing
parameters		methods would be the same for both pits, this is a reasonable assumption. However, pit optimisation work is currently underway for
Minimu forton		Burtville, and cut-off grades and other assumptions for limiting the resource should be reviewed when this work is completed.
Mining factors or assumptions	•	The Burtville deposit would be mined by open pit extraction. The previous pit design would have extended to 120m below surface (360
assumptions		mRL). The gold price used for the optimisation/pit design is unknown, but the spot price in late 2012 was ~AUD\$1700/oz.
	•	Further pit optimisation is underway but given the much higher current gold price (~AUD\$2600/oz), then it is probable that the pit shells would be deeper and reach towards the extent of the modelled mineralisation.
	•	The 340 mRL has therefore been used as the base for reporting the classified resource.
Metallurgical	•	Historical metallurgical test work and actual open cut mining showed the mineralised material had very good to excellent recoveries in
factors or		a standard CIL gold processing plant (>90% for some transitional material, but generally above 98% in fresh rock.
assumptions		d stated a St. gold processing plant (55% for some dational material, sat generally above 55% in noon root.
Environmental	•	The Burtville deposit has previously been mined by open pit methods in the 2012-2013 by Focus, and there are existing waste dumps
factors or		and open cut pits.
assumptions	•	Other operations in the area in the last 8 years have been Focus' Chatterbox – Apollo Pits south along strike and at Euro South to the
		SE and is 27 km from Goldfield's Granny Smith gold mine.
	•	Therefore, there is extensive mining history in the region, and there are no unforeseen
Dulk danate	•	environmental considerations that would preclude conventional open cut mining and waste dump construction.
Bulk density	•	Bulk density test work was gathered throughout the life of the historical open cut pit (mining ceased in 2013) with the water immersion
	_	technique used for these determinations. Average half despits values were essigned per medalled lithelega/weathering demain (1.8 t/m³ for evidiced, 2.45 t/m³ for transitional
	آ ا	Average bulk density values were assigned per modelled lithology/weathering domain (1.8 t/m³ for oxidised, 2.45 t/m³ for transitional and 2.65 t/m³ for fresh rock).
Classification	•	The Indicated Mineral Resource has a nominal drill spacing of 20 mN x 20 mE or closer (10 mE x 10 mN in grade control drilled areas),
	ĺ	is not more than 20m laterally beyond drilling, not more than 10 m below the base of drilling and blocks estimated using the first search
		pass.
	•	The Inferred Mineral Resource is material within the mineralised domain, but not meeting the criteria for Indicated.
	•	The Indicated part of the resource only extends 10 m below the limit of drilling (360 mRL maximum), and the Inferred resource only to
		the 340 mRL maximum.
	•	This classification considers the confidence of the resource estimate and the quality of the data and reflects the view of the Competent
		Person.
		No external audits of the mineral resource have conducted, although the independent consultants used for the resource estimate (Cube

Criteria	Commentary		
	Consultants) conduct internal peer review.		
Discussion of	This is addressed in the relevant paragraph on Classification above.		
relative accuracy/	The Mineral Resource relates to global tonnage and grade estimates.		
confidence			

• Section 3 Details for the Beasley Creek deposit from ASX Announcement "Beasley Creek Mineral Resource Grows by 29%" Dated 20/08/2020

29%"	Dated 20/08/2020
Criteria	Commentary
Database integrity	 Data was geologically logged electronically; collar and downhole surveys were also received electronically as was the laboratory analysis results. These electronic files were loaded into an acQuire database by the company in-house Database Administrator. Data was routinely extracted to Microsoft Access during the drilling program for validation by the geologist in charge of the project. Focus Minerals' database is a Microsoft SQL Server database (acQuire), which is case sensitive, relational and normalised to the Third Normal Form. Because of normalisation, the following data integrity categories exist: Entity Integrity: no duplicate rows in a table, eliminated redundancy/chance of error. Domain Integrity: Enforces valid entries for a given column by restricting the type, the format or a range of values. Referential Integrity: Rows cannot be deleted which are used by other records. User-Defined Integrity: business rules enforced by acQuire and validation codes set up by Focus Minerals. Additionally, in-house validation scripts are routinely run in acQuire on Focus Minerals' database and they include the following checks: Missing logging, sampling, downhole survey data and hole diameter Overlapping intervals in geological logging, sampling, down hole surveys Checks for character data in numeric fields Data extracted from the database were validated visually in GEOVIA Surpac software, ARANZ Geo Leapfrog software and Datamine software. Also, when loading the data any errors regarding missing values and overlaps are highlighted.
Site visits	 Alex Aaltonen, the Competent Person for Sections 1 and 2 of Table 1 is Focus Minerals' General Manager - Exploration and conducts regular site visits.
Geological	 Michael Job, the Competent Person for Section 3 of Table 1, has not visited site. All Focus Minerals drill holes and historic mining data were used to guide the geological interpretation of the mineralisation.
interpretation	 The mineralised shoot interpretation is based on the Beasley Creek Shear Zone and the brecciated sediments and veins within the shear. Au grades are used to assist in the interpretation. The orientation of the shoots in the southern part of the deposit reflects the known shoot geometry from the previous mining.
	 In the southern part of the deposit, the south-east plunge of the mineralised shoots is confirmed by the outcrop and mined mineralisation in the historical WMC pit, and any alternative interpretation is unlikely. However, for the northern part of the deposit away from the pit, there may be alternatives to the geometry of the shoots modelled, although the global tonnages are smaller here and unlikely to be significantly different if an alternative interpretation was adopted.
	 It is recognised that the WMC RC data in places shows down hole contamination (due to the wet ground conditions and older cross-over sub RC hammers used). Much of this data is within the historical pit and has very little influence over the resource estimate below the pit. Where this RC data is below the pit, it has not been used for the interpretation as it would create incorrect long intercepts. However, this data has been used for grade interpolation, as studies showed this data within the interpreted shoots was very similar statistically to the modern RC and DDH drilling undertaken by Focus Minerals.
	 Contiguous high-grade zones (>5 ppm Au) were modelled as separate domains. The weathering/oxidation profiles at Beasley Creek is deep, with clays and saprock extending up to 250 m below surface in the eastern part of the deposit.
	 Leapfrog software was used for the interpretation of the mineralised shoots and the regolith domains. Each mineralised shoot intercept was coded in the database before being imported into Leapfrog, so the resulting solids honour the data well.
Dimensions	• The deposit extends over a strike length of 1100m and extends to at least 280m below the surface. The deposit is arcuate in shape, striking towards the north-west in the northern part of the deposit, and to the south-west and then south in the southern part. There are numerous mineralised lodes, plunging at 30 to 50° to the south-east in the southern part of the deposit, and dipping at 50 to 60° to the north-east in the northern part.
	The individual lodes range from 5 m to 30 m thick (averaging 15 m), from 20 m to 80 m wide (averaging 30 m) and can extend up to 400 m down plunge.
Estimation and modelling techniques	 Estimation of the mineral resource was by ordinary kriging using Datamine software. The estimation process was as follows: Drill hole database including coded shoot intercepts imported into Datamine. Drill hole data composited to 1m downhole intervals, with a minimum allowable composite of 0.25 m at the shoot base. Composited data imported into Supervisor software for statistical and geostatistical analysis. Top-capping applied per mineralised shoot – caps ranged between 5 to 10 ppm Au for the main mineralised shoots, and up to 25 ppm
	Au for the high-grade shoots. The caps were based on inflections and discontinuities in the histograms and log-probability plots. • Variography was done on data transformed to normal scores, and the variogram model was back transformed to original units. Variography was only performed for mineralised shoots with more than 150 samples (seven shoots), and these were applied to the other shoots that had the closest statistical similarities.
	As the mineralised shoots have different orientations, the applied variogram rotations (for the smaller shoots) were adjusted (and checked) for each individual shoot. The prior property of the hard product to high property of the form of the control of th
	 The variogram models had moderate to high nugget effects (~30 to 50% of total sill), and with a down-plunge range of 50 to 60 m. The range across dip was small, generally 6 to 8 m. The ellipsoid search parameters were based on the variogram ranges, with the search ellipse dimensions about 90% of the variogram

Criteria	Commentary
Ontona	range, with anisotropies retained. A minimum of 8 and maximum of 14 (1m composite) samples per block were used, with a maximum
	of 4 samples per drill hole. Estimates were into parent blocks, not sub-blocks.
	Search ellipse rotation directions were the same as the variograms, for each shoot.
	If a block was not estimated with these search parameters, then the ellipse was expanded by a factor of two, using the same sample
	numbers. If a block was not estimated on the second pass, then a third pass was used – this was an expanded search of a factor of 4
	compared to the first pass, with a minimum of two and maximum of 18 samples.
	• For the block model, 66% of blocks were estimated on the first pass, 30% on the second and 3% on the third. No blocks in the mineralised shoots were left unestimated. These search volumes assisted with later resource classification.
	The block model itself was a non-rotated model in MGA94 grid, with a parent block size of 10 mE x 20 mN x 5 mRL – this is about half
	of the average drill spacing in the well-mineralised areas.
	Sub-blocking was to a minimum of 1.25 mE x 2.5 mN x 1.25 mRL for accurate volume representation, and the blocks and sub-blocks
	were coded by mineralised shoot and lithology/weathering and topography.
	Estimates of Au grades were validated against the composited drill hole data by extensive visual checking in cross-section, plan and on
	screen in 3D, by global (per shoot) comparisons of input data and model, and by semi-local statistical methods (swath plots). All methods
Majatuwa	showed satisfactory results.
Moisture	 There is significant groundwater at Beasley Creek, but bulk density determinations (see below) were made on dried core. Tonnages are therefore estimated on a dry basis.
Cut-off	The cut-off grade of 0.8 ppm Au was established from the previous pit optimisation run (see below) and gave a consistent cash flow. As
parameters	the Au price is now higher than the price used during this optimisation study (AUD\$2300/oz cf. \$1800/oz), then the reporting cut-off
	grade used is a conservative approach.
Mining factors	The Beasley Creek deposit would be mined by open pit extraction. Previous pit optimisation runs have extended to 180 m below surface
or assumptions	(250 mRL), using a gold price of AUD\$1786/oz.
	• Further pit optimisation is underway but, given the much higher current gold price (~AUD\$2300/oz), it is probable that the pit shells would
	be deeper.
Motellurgical	The 250 mRL has therefore been used as the base for reporting the classified resource. WMC and the control of the dead found of the dead of
Metallurgical factors or	WMC reported reconciled recovery of blended feed at Windarra between 1991 and 1994, although this was a blend from a number of acuracy WMC mine reconsiliation for the period report from 92%, 02%.
assumptions	sources. WMC mine reconciliation for the period ranged from 82% - 93% Test work was completed on samples by Metex/Delta in the late 1990s for heap leach and column test work and reported 94% recovery
	in 56 days and 80% in 20 days, which was considered favourable for heap leach.
	Eleven samples were further acquired by Delta Gold and subjected to bottle roll test work, returning 84-98% recovery after 48 hours.
	Nine of the 11 samples returned average 94.28% recovery after 24 hours with very low reagent consumption.
	Focus Minerals completed two new samples at ALS in September 2019. The material was considered in natural state already too fine
	to require grinding and was simple-sized post-test work.
	Later sizing showed the P80 for one sample was 54 micron and the other 75 microns. As such some of the in situ material may not need a print of a 1.1. The state of the initial may not need to be a print of the initial may not need to
	a grind at all. • The leach results for these two Beasley Creek samples were good with 96.74% and 97.74% recovery after 4 hours and, 94.44% and
	92.67% recovery at 2 hours, with low reagent consumption.
	These results confirm earlier results from Beasley Creek and indicate it will run very well in either a mill or as a heap leach.
	Metallurgical test work at Beasley Creek South shows a similar response to samples processed at ALS in 2019
Environmental	Beasley Creek was mined by open pit methods between 1987-1993 by WMC and there are existing waste dumps and open cut pits.
factors or	Other operations in the area in the past eight years have been Focus Minerals' Chatterbox Apollo Pits 8.5km south along strike and at
assumptions	Euro South, 19km to the south-east.
	Therefore, there is extensive mining history in the region, and there are no unforeseen environmental considerations that would preclude account to a local point with the property of the second proper
	 conventional open cut mining and waste dump construction. A potential heap leach would have greater environmental management burden than sending to a CIL plant but would not preclude
	mining.
Bulk density	Bulk density test work was initially on diamond core samples from different geology domains, with the water immersion technique used
	for these determinations. These results were compared with external lab results in order to develop an accurate database.
	Follow up PQ3 holes were drilled for down hole gamma logging of in situ bulk density at 0.2m downhole spacing. In additional
	available open HQ3 holes were down hole gamma logged to build a significant high-resolution dataset at Beasley Creek.
	The regolith at Beasley Creek was comprehensively modelled in Leapfrog and used to evaluate all bulk density results by regolith
	domain.
Classification	 The statistics of each domain were analysed to determine refined average bulk density values to be applied to each regolith domain. The mineralised shoots are classified as Indicated where the drilling pattern is 40 m along strike and 20 m down dip, and within 20m of
Ciassilleation	• The mineralised shoots are classified as indicated where the drilling pattern is 40 m along strike and 20 m down dip, and within 20m of the lower-most drilling in the shoot
	All the rest of the mineralised shoots outside this area are classified as Inferred.
	This classification considers the confidence of the geological interpretation and the quality of the data and reflects the view of the
	Competent Person.
Audits or	No external audits of the mineral resource have conducted, although the independent consultants
reviews	used for the resource estimate (Cube Consultants) have critically reviewed the geological interpretations provided by Focus and the
D:	quality of the WMC RC drilling.
Discussion of	This is addressed in the relevant paragraph on Classification above. The second of the relevant paragraph on Classification above.
relative accuracy/	The Mineral Resource relates to global tonnage and grade estimates.
confidence	
-	

 Section 3 Details for the Beasley Creek South deposit from ASX Announcement "Beasley Creek South Delivers High Grade Mineral Resource" Dated 15/07/2020

Criteria	Commentary
Database	 Data was geologically logged electronically; collar and downhole surveys were also received electronically as was the laboratory analysis
integrity	 bala was geologically logged electronically, collar and downlote surveys were also received electronically as was the laboratory alrahysis results. These electronic files were loaded into an acQuire database by the company in-house Database Administrator. Data was routinely extracted to Microsoft Access during the drilling program for validation by the geologist in charge of the project. FML's database is a Microsoft SQL Server database (acQuire), which is case sensitive, relational and normalised to the Third Normal Form. Because of normalisation, the following data integrity categories exist: Entity Integrity: no duplicate rows in a table, eliminated redundancy and chance of error. Domain Integrity: Enforces valid entries for a given column by restricting the type, the format or a range of values. Referential Integrity: Rows cannot be deleted which are used by other records.
	 User-Defined Integrity: business rules enforced by acQuire and validation codes set up by FML. Additionally, in-house validation scripts are routinely run in acQuire on FML's database and they include the following checks: Missing collar information
	 Missing logging, sampling, downhole survey data and hole diameter Overlapping intervals in geological logging, sampling, down hole surveys
	 Checks for character data in numeric fields. Data extracted from the database were validated visually in GEOVIA Surpac software, ARANZ Geo Leapfrog software and Datamine software. Also, when loading the data any errors regarding missing values and overlaps are highlighted.
Site visits	 Alex Aaltonen, the Competent Person for Sections 1 and 2 of Table 1 is FML's General Manager - Exploration and conducts regular site visits.
Geological	Hannah Kosovich, the Competent Person for Section 3 visited site in September 2019. All qualitable drill hale and historic mining data was used to guide the goalegiest interpretation of the mineralisation. Although persuasion
interpretation	 All available drill hole and historic mining data was used to guide the geological interpretation of the mineralisation. Although percussion drill holes were used with caution due to the poor sample recovery and quality that is inherent with the drilling method at Beasley Creek South.
	 The mineralised geological interpretation was generated in Seequent Leapfrog Geo implicit modelling software. Three larger mineralised lodes were generated by coding mineralised intervals along strike and down dip of the known trend using logged geology as a guide. An approximate 0.5g/t cut-off was used, infrequently sub 0.5g/t samples were included for continuity. Within the larger mineralised lodes, several cores of higher-grade mineralisation were modelled as separate domains. Two hanging wall lodes were modelled also with higher-grade cores within each lode.
	 Minor deviation of the lode geometry was noticed between drill holes down-dip. A gap in the main lode was modelled corresponding with less altered/weathered coarse calc – silicate mafic intrusion. Tight spaced infill drilling has been used to better define its location and extent.
Dimensions	 The deposit extends over a strike length of 450m and extends to approximately 250m below the surface. The deposit is striking towards the NNW. There are three main lodes of mineralisation and two hanging wall lodes. The bulk of the mineralisation has been modelled from surface. The lodes range from 5m to 25m wide (averaging 10m), with the internal HG shoots ranging from 1m to 15m wide (averaging 5m). The two hanging wall lodes average 3m wide.
Estimation and modelling	 The drill hole samples were composited to 1m within each domain. This is the dominant sampling interval. The boundaries between lodes and also between the HG shoots and surrounding lodes were considered "hard" boundaries and no drill
techniques	hole information were used by another domain in the estimation. Composited assay values of each domain were exported to a text file (.csv) and imported into Snowden Supervisor for geostatistical analysis.
	 A review of histograms, probability plots and mean/variance plots by domain revealed outlier sample values in some of the lodes/shoots. A maximum top-cut of 40g/t Au and an average of 25g/t Au was used for the HG shoots; maximum top-cut of 7g/t Au and an average of 4g/t Au was used for surround lodes. Assays above the top-cut were set to the top-cut value.
	 Variograms were modelled in Supervisor for the main lode and one of the smaller lodes that had the largest number of samples. Other minor lodes shared the minor lode variogram.
	• GEOVIA Surpac Software was used for the estimation and modelling process. The model was created in GDA 94 grid co-ordinates. Block sizes for the model were 10m in Y, 10m in X and 5m in Z direction. Sub celling of the parent blocks was permitted to 1.25m in the Y direction, 1.25m in the X direction and 2.5m in the Z direction. Sub-blocking was used to best fill the wireframes and inherit the grade of the parent block. No rotation was applied to the orientation of the blocks.
	 Block size is approximately ½ of the average drill hole spacing along strike and across strike was selected to best fill the wireframe volumes. An Ordinary Kriging (OK) estimation technique was selected and used the variograms modelled in Supervisor.
	The main lode was estimated using a minimum (8) and maximum (16) samples were selected based on a Kriging Neighbourhood analysis in Supervisor.
	 An elliptical search was used based on range/ratio of the Variograms. Three search passes were run in order to fill the block model with estimated Au values. After each search pass the search range was increased and the minimum number of samples was decreased.
	 The estimate was validated by several methods. An initial visual review was done by comparing estimated blocks and raw drill holes. Tonnage weighted mean grades were compared for the lodes with no major differences. Swath plots of drill hole values and estimated Au grades by northing and RL were run and showed that the estimated grades honoured
	the trend of the drilling data. Tonnages are estimated on a dry basis.
Moisture	ו טווומעשט מום פטנווומנפע טוו מ עוץ טמטוט.

Criteria	Commentary			
Cut-off	The open pit cut-off grade of 0.55 g/t Au (Gold Price AUD \$1,800/oz) was established from the 2019 Laverton Scoping Study.			
parameters	• For the purposes of reporting this open pit resource a cut-off grade of 0.8 g/t Au has been used which is in line with the recently reported			
	and nearby Beasley Creek Resource Estimate (Announced 25/10/2019).			
Mining factors or	The Beasley Creek South deposit would be mined by open pit extraction. Nearby Beasley Creek has been optimised in the scopin			
assumptions	study down to the 250mRL (approx.180m below surface) for reasonable open pit extraction the same RL cut off has been applied to the			
	Beasley Creek South open pit resource.			
Metallurgical	Beasley Creek South samples are being compiled for metallurgical test work.			
factors or	Samples are geologically / mineralogically similar to the nearby Beasley Creek deposit.			
assumptions	As stated in the Beasley Creek release 25 October 2019:			
	 Focus sent two samples for test work to ALS in September 2019. The material was considered in natural state already too fine to require grinding and was simple sized post-test work. 			
	 Later sizing showed the P80 for one sample was 54 micron and the other 75 microns. As such some of the in situ material may not need a grind at all. 			
	 The leach results for these two Beasley Creek samples were good with 96.74% and 97.74% recovery after 4hrs and, 94.44% and 92.67% recovery at 2 hrs, with low reagent consumption. 			
	These results confirm earlier results from Beasley Creek and indicate it will run very well in either a mill or as a heap leach.			
Environmental	Beasley Creek South is approximately 400m south of the existing Beasley Creek open pit which was mined by open pit methods in the			
factors or	1980s by WMC.			
assumptions	• It forms part of the Chatterbox Shear group of deposits which have been historically mined and there are no unforeseen environmental			
	considerations that would preclude conventional open cut mining and waste dump construction.			
Bulk density	Bulk density test work was routinely completed on FML diamond core samples targeting all geological/weathering domains. The water			
	immersion technique used for these determinations.			
	During May 2020, 9 whole or partial Beasley South and 2 further Beasley Creek holes were downhole logged using a bottom loading			
	gamma ray source sonde to directly measure formation density.			
	This logging method delivers bulk high-quality data with sample intervals of 0.2m.			
	The downhole logging data was categorised by modelled geological/weathering domains. This allowed direct comparison of various			
	sourced data within each relevant domain using box and whisker plots.			
	Analysis of the data showed tight correlation between downhole logging, and laboratory and company Archimedes immersion method			
	specific gravity determinations in most domains. However, some oxidised shear zone bulk density samples measured by the water			
	immersion technique fell below acceptable data ranges. An analysis of samples with very low density concluded that these samples were affected by noticeable dehydration/shrinkage cracks.			
	These types of samples can dry to form 0.2 – 0.5m sized sticks of core that can be measured but should not be measured as they			
	deliver spurious results. These samples with very low densities (<1.2 SG) were cut out of the data. Equally, anomalously high-density values were examined and were determined to be spurious were discarded from the dataset.			
	It is also noted that the immersion method requires sticks of core at least 0.2m long. Unfortunately, this creates a sample bias towards			
	more clay rich samples that tend to dry into sticks of core. These samples have lower average densities than more blocky quartz,			
	sulphidic black shale or gossan units that could not be routinely measured. It is interpreted that this is responsible for the slightly lower			
	average for oxidised shear samples measured using the immersion technique.			
	Once the data was compiled and sorted a simple average density was then assigned to each geological unit/weathering domain.			
Classification	The mineralised lodes and internal HG shoots are classified as Indicated above the 300mRL (130m depth and limit of most drilling) with			
	the bulk of the lodes filling within the first search pass.			
	Mineralised lodes below the 250mRL are classified as Inferred. The hanging wall lodes which require further delineation are classified			
	as Inferred.			
Audits or reviews	No external audits of the mineral resource have been conducted.			
Discussion	This is addressed in the relevant paragraph on Classification above.			
of relative	The Mineral Resource relates to global tonnage and grade estimates.			
accuracy/				
confidence				

 Section 3 Details for the Wedge deposit from ASX Announcement "Wedge Open Pit Resource Update" Dated 24/01/2020

Criteria	Commentary
Database integrity	 Data was geologically logged electronically; collar and downhole surveys were also received electronically as was the laboratory analysis results. These electronic files were loaded into an acQuire database by the company in-house Database Administrator. Data was routinely extracted to Microsoft Access during the drilling program for validation by the geologist in charge of the project. FML's database is a Microsoft SQL Server database (acQuire), which is case sensitive, relational and normalised to the Third Norma Form. Because of normalisation, the following data integrity categories exist: Entity Integrity: no duplicate rows in a table, eliminated redundancy and chance of error. Domain Integrity: Enforces valid entries for a given column by restricting the type, the format or a range of values. Referential Integrity: Rows cannot be deleted which are used by other records. User-Defined Integrity: business rules enforced by acQuire and validation codes set up by FML. Additionally, in-house validation scripts are routinely run in acQuire on FML's database and they include the following checks: Missing logging, sampling, downhole survey data and hole diameter Overlapping intervals in geological logging, sampling, down hole surveys

Criteria	Commentary			
	Checks for character data in numeric fields.			
	Data extracted from the database were validated visually in GEOVIA Surpac software and ARANZ Geo Leapfrog software. Also, when			
	loading the data any errors regarding missing values and overlaps are highlighted. Historic data has been validated against WAMEX reports where possible.			
Site visits	 Alex Aaltonen, the Competent Person for Sections 1 and 2 of Table 1 is FML's General Manager - Exploration and conducts regular site visits. 			
	 Hannah Kosovich, the Competent Person for Section 3 of Table 1 is FML's Resource Geologist and last visited site in September 2019. 			
Geological	All available drill hole and historic mining data was used to guide the geological interpretation of the mineralisation.			
interpretation	 The mineralised geological interpretation was generated in Seequent Leapfrog Geo implicit modelling software. A larger mineraliset trend of the entire Wedge/Lancefield North deposits was generated by coding mineralised intervals along strike and down dip of the known trend using logged geology as a guide. An approximate 0.2g/t cut-off was used, infrequently sub 0.2g/t samples were include for continuity. To the North of Lancefield North deposit an east/west running cross fault appears to terminate the mineralisation. Within the larger mineralised trend, small higher-grade shoots were modelled as separate domains. Several hanging wall lodes were modelled. 			
	 Minor deviation only of the lode geometry was noticed between drill holes down-dip. Along strike two mineralised lodes have been interpreted that appear to be cross-cutting structures. 			
Dimensions	 The entire Wedge/Lancefield North deposit strikes NE with a total strike length of approx. 2.6km. Lancefield North sits along the NE strike some 250m from the Wedge trend. The main lode of mineralisation has been modelled greater than 200m below surface, however only the top 130m of the estimate is reported. The bulk of the mineralisation has been modelled from surface. Mineralisation has an average width of 5m. 			
Estimation and modelling	 A total of 549 drill holes were used in the Estimation; 11 diamond holes, 1 diamond hole with an RC pre-collar and 537 RC holes for a total of 37,891.3m. 			
techniques	The drill hole samples were composited to 1m within each domain. This is the dominant sampling interval.			
	All domain boundaries were considered "hard" boundaries and no drill hole information were used by another domain in the estimation.			
	Composited assay values of each domain were exported to a text file (.csv) and imported into Snowden Supervisor for geostatistical			
	 analysis. A review of histograms, probability plots and mean/variance plots for the main lode domain revealed outlier sample values. A maximum 			
	top-cut of 25g/t Au and an average of 10g/t Au was used for the different lodes, with assays above the top-cut set to the top-cut value.			
	Variograms were modelled in Supervisor for the main lode and one of the smaller lodes that had the largest number of samples. Other			
	minor lodes shared the minor lode variogram. • GEOVIA Surpac Software was used for the estimation and modelling process. The model was created in GDA 94 grid co-ordinates.			
	Block sizes for the model were 12.5m in Y, 12.5m in X and 5m in Z direction. Sub celling of the parent blocks was permitted to 1.562m in the Y direction, 1.562m in the X direction and 1.25m in the Z direction. Sub-blocking was used to best fill the wireframes and inherit the grade of the parent block. No rotation was applied to the orientation of the blocks.			
	 Block size is approximately ½ of the average drill hole spacing along strike and across strike was selected to best fill the wireframe volumes. 			
	 An Ordinary Kriging (OK) estimation technique was selected and used the variograms modelled in Supervisor. The main lode was estimated using a minimum (6) and maximum (20) samples were selected based on a Kriging Neighbourhood analysis in Supervisor. 			
	The smaller lodes were estimated using a minimum (6) and maximum (14) samples.			
	 An elliptical search was used based on range/ratio of the Variograms. Three search passes were run in order to fill the block model with estimated Au values. After each search pass the search range was 			
	 increased and the minimum number of samples was decreased. The estimate was validated by a number of methods. An initial visual review was done by comparing estimated blocks and raw drill 			
	 holes. Tonnage weighted mean grades were compared for the lodes with no major differences. 			
	Swath plots of drill hole values and estimated Au grades by northing and RL were run and showed that the estimated grades honoured			
	the trend of the drilling data. • Available production figures for Wedge were used as a comparison with the estimated material within the pit shells. Production figures			
	state 262,023t @ 2.53g/t Au HG ore was mined from the pits and 260,544t @ 2.51 g/t of HG ore was Milled.			
Moisture	Tonnages are estimated on a dry basis. The state of the North Action of the Nort			
Cut-off parameters Mining factors or	 The mineral resource for the Wedge/Lancefield North deposits has been reported above a 0.8g/t Au cut-off. The Wedge/Lancefield North deposits would be mined by a cut-back on the existing open pits. 			
assumptions				
Metallurgical factors or	 Metallurgical test work was carried out by AMMTEC on behalf of Hill Minerals NL in August and September 1988. An end of mine report by Ashton Gold states mill recoveries were typically in the range of 94% - 95% 			
assumptions Environmental	Wedge has been historically mined by open pit methods.			
factors or assumptions				
Bulk density	Density values were assigned based on weathering profile and rock type, using SG test work on FML diamond core samples and historic forms and in the profile Account of the transported by the transported by the profile Account of the transported by the transported by the transported by the transported by the profile Account of the transported by the tr			
	figures used in the region. An average SG of 2.06 was used for the transported and cemented horizon, 2.0 for the highly weathered clay weathering profile, 2.49 for transitional material and 2.77 for Fresh rock were applied.			
01 17 11	The water immersion technique was used for the FML measurements.			
Classification	Material has been classified Indicated and Inferred based on a number of criteria such as geological continuity, drill hole spacing,			

Criteria	Commentary		
	estimation pass and proximity to existing open pit.		
Audits or reviews	No external audits of the mineral resource have conducted.		
Discussion	This is addressed in the relevant paragraph on Classification above.		
of relative	The Mineral Resource relates to global tonnage and grade estimates.		
accuracy/			
confidence			

Section 4 Estimation and Reporting of Ore Reserves (Criteria listed in the preceding section also apply to this section.)

	receding section also apply to this se	ction.)			
Criteria	Commentary				
Mineral	The Mineral Resources used for the estimation of Ore Reserves were previously reported as summarised in Section 3 of Table 1.				
Resource	The Mineral Recourses has been compiled by:				
Estimate for	The Mineral Resources has been compiled by: • Ms. Hannah Kosovich is the Competent Person for the Karridale, Beasley Creek South and Wedge Lancefield Mineral Resources.			South and Wedge Lancefield Mineral Resources	
conversion to	 Mr. Michael Job is the Competent Person for the Burtville and Beasley Creek Mineral Resources. 				
Ore Reserves		employee of Focus M	nerals and a Mer	nber of the Aus	stralasian Institute of Mining and Metallurgy (AusIMM).
	to the activity that they have ur The Mineral Resources are	dertaken to qualify as inclusive of these Ore	a Competent Per Reserves.	sons as define	
	applied to the reporting of	•	ay, the cut-off gra	ades applied in	the reporting of the Ore Reserve are lower than those
			logical interpreta	tion in the Min	eral Resource and the Ore Reserve, the reporting of
	Mineral Resources at a high				
Site visits					eneral Manager of Exploration and Geology, conducts
	regular site visits.	one r order rer deducin	o rana z or rabio	7 1 10 1 III E O O	onoral manager or Exploration and Goology, conducto
	· ·	petent Person for Sect	ion 3 of Table 1 is	FML's Resou	rce Geologist and has conducted site visits in the past.
					d and reviewed by Mr. Igor Bojanic, who is a Fellow of
					RPM Advisory Services Pty Ltd (RPMGlobal).
	No site visit was undertaket	n due to COVID-19 tra	vel restrictions. N	lr. Bojanic is ex	xperienced in gold operations in the Laverton region
Study status	The Mineral Resources ha	ve been converted to	Ore Reserves by	means of a P	reliminary Feasibility Study (PFS) including economic
	assessment.				
					vable, and the Project is economically viable.
Cut off	 The PFS included analysis 		• .	•	
parameters		vary by pit and materi	al type due to va	riations in haul	age costs from pit to the Run of Mine (ROM) pad and
	metallurgical recoveries.				
	Deposit	Oxide	Transition	Primary	1
	<u> </u>	g/t	g/t	g/t	_
	Karridale	0.48	0.49	-	
	Burtville	0.48	0.48	0.50	
	Beasley Creek	0.47	0.48	-	
	Beasley Creek South	0.47	0.48	-	
	Wedge	0.47	0.48	_	
	Lancefield	0.47	0.48	0.49	
	Lunconoid	0.47	0.40	0.40	1
Mining factors	Toohnigal analysis was ass	anloted in the DEC to a	otormino the med	at appropriate	mining method and estimate ore loss and dilution.
or assumptions	Selective open cut mining				
oi assumptions					ls by regularisation of the sub-blocks to the following
	sizes:	SC IIIOGCIS WOIG COIIVCI	tou to run or mine	o mining mode	is by regularisation of the sub-blocks to the following
	01200.				
	Pit	Block Dimensio	n		
	Karridale	2.5 x 5 x 2.5 m			
	Burtville	5 x 5 x 2.5 m			
	Beasley Creek	2.5 x 5 x 2.5 m			
	Beasley Creek South				
	Wedge	3.125 x 3.125 x 2.5	m		
	rreage	J. 12J A J. 12J A Z.:			
					ditioning with minimum sub-blocks dimensions of 5 x s, no further adjustments were applied.

Criteria Commentary Ore loss and dilution is reported relative to in situ Resource quantities and summarised below. Ore Avg. Avg. Ore Loss Grade of Dilution Grade of Pit Quantity Ore Loss Quantity Dilution (%) (g/t) (%)(g/t) Karridale 16% 0.25 15% 0 Beasley Creek 0.1 13% 1.66 13% 0.03 Beasley South 16% 1.26 6% Wedge 16% 0.9 20% 0.04 Minimum mining width was 20m followed by a "good-bye" cut. Minimum cut-back width is 25m. Geotechnical criteria for the design of the open pits were developed by Green Geotechnical Pty Ltd for the purpose of the PFS. The resultant overall slope angles, following pit design, are summarised below. Hanging Wall Footwall Deposit (degrees) (degrees) Karridale 36 to 42 32 to 46 Burtville 46 to 43 40 to 45 Beasley Creek 35 to 46 36 to 38 Beasley Creek South 43 37 to 43 Wedge 43 to 47 44 to 57 The economic pit shell was defined using Whittle 4X pit optimisation software ("Whittle 4X") with inputs such as geotechnical parameters, run of mine model, metallurgical recoveries and operating and sustaining capital costs. Only Measured and Indicated Resources were used to identify the economic mining limit. In defining the economic pit shell, metallurgical recoveries were not applied to Primary material from Karridale, Beasley Creek, Beasley Creek South and Wedge due to limited metallurgical test work in Primary material from these deposits. Metallurgical recoveries were applied to Primary material from Burtville and Lancefield. Inferred Mineral Resources were assumed to be waste rock for the pit shell selection using Whittle. Inferred Mineral Resources included within the selected pit shells was treated as ore in the mine scheduling and economic analysis. A breakdown of Inferred Material by pit is summarised below: Inferred Quantity MEA+IND INF Pit Mineral (M + I + I)(kt)(%) (%) Quantity (kt) 5,659 99% 1% Karridale 61 Burtville 157 96% 3.531 4% Beasley Creek 1.815 100% 0% 28% Beasley South 284 1,010 72% 99% Wedge 9 814 1% Total 510 12,828 96% 4% No specialised infrastructure is required to support the proposed mining method. The PFS mining schedule strip ratios (inclusive of Inferred Resources described above) are Karridale 7.0:1, Burtville 1.0:1, Beasley Creek 18.7:1, Beasley Creek South 19.7:1 and Wedge/Lancefield 13.5:1. Overall strip ratio for the PFS mining schedule is 8.4:1. Conventional open cut mining is a very common mining method used through the mining industry and requires no specialist The required supporting infrastructure has been included in the PFS. Major items include refurbishment of the haul roads connecting the pits to the Barnicoat mill, workshops and offices near the Barnicoat area and satellite offices and facilities near the main mining Metallurgical A reasonable quantity of test work has been conducted in several campaigns over many years. Additional metallurgical test work will factors or provide more confidence in the performance of the milling circuit and gold recoveries. The proposed flowsheet and the refurbished Barnicoat processing plant is considered capable of successfully handling the Laverton assumptions oxide and transition ores as well as selected primary ore types. Primary ores from Karridale, Beasley Creek, Beasley Creek South and Wedge have been excluded from the PFS and Ore Reserve. Ores from Burtville, Beasley Creek, Wedge and Lancefield open pits have been successfully processed in a number of processing operations, including the Barnicoat mill, providing confidence in the proposed outcomes. The Barnicoat plant will recover gold via a gravity circuit and by a carbon-in-leach process. The plant is designed to process 1.5 Mt/a

- No major presence of deleterious material has been identified.
- Head grade/recovery relationships have been estimated for each material type by pit. Life of mine average metallurgical recovery as estimated in the PFS is 91.0%

Environmental

- A review of the environmental permitting required was completed as part of the PFS.
- No environment impact statement has been completed to date.
- Environmental studies relating to the re-commencement of mining and processing operations are yet to commence.
- It is estimated to progress from commencement of additional baseline studies to project construction will require approximately 18 to 24 months.

Criteria	Commentary	
	The Project is a brown-field operation. New pits are proposed at Beasley Creek South and Karridale which are nearby previously	
	operated open pits. All other proposed open pits are extensions of previously mined pits. The Barnicoat mill, already in place, is not	
	proposed to be relocated.	
	No major environmental or permitting risks have been identified for the Project. RPMGlobal considers that following completion	
	of the required baseline studies and assessments it is likely the Project will receive relevant permits and approvals. These	
Infrastruatura	approvals will outline the conditions under which the Project will need to be operated.	
Infrastructure	The Project is located approximately 8km East of the town of Laverton. Site infrastructure requirements have been defined as part of the PES.	
	 Site infrastructure requirements have been defined as part of the PFS. There is existing infrastructure and facilities on-site, including the de-commissioned Barnicoat mill, buildings, workshops and pit to mil 	
	haul roads. These will require upgrading prior to being re-commissioned.	
	The PFS proposes the following infrastructure and services for the Project:	
	Power to be generated via a diesel power station with a Peak Power load of 4.7 MW.	
	Re-commission of water bores.	
	 Accommodation camp of 200 to 250 persons potentially located in Laverton. 	
	 Satellite crib areas, offices, workshops and go-bays to support mining operations at the Karridale/Burtville area and the Beasley 	
	Creek/Beasley Creek South/Wedge area.	
	Tailings to be stored in previously mined pits adjacent to the Barnicoat mill.	
	 Some additions to the existing haul roads between the pits and mill are required. Sufficient land is available for the placement of all required, infrastructure, including one processing plant, waste rock storage, explosives 	
	 Sufficient land is available for the placement of all required. infrastructure, including ore processing plant, waste rock storage, explosives magazine and accommodation village 	
	Further studies are required to confirm the site water balance and capital and operating costs associated with water supply to the project	
Costs	The estimating of capital and operating costs was supported by engineering commensurate with a preliminary feasibility study.	
	Mobile plant (mining equipment) capital costs for major items were based on recent quotes from equipment providers.	
	Fixed plant capital costs were primarily based on in-house data and benchmarking. An average contingency of 18% was applied to initia	
	capital costs.	
	Some capital items, such as the diesel power plant and accommodation camp were cost based on a Build Own Operate Transfe	
	(BOO/T) contract basis. Additionally, mining facilities, such as satellite facilities, workshop plant, diesel generators were costed on a life	
	of mine leasing basis.	
	Capital costs were based on an AUD to USD exchange rate of 0.7.	
	• Mining, processing and G&A operating costs were largely derived from a first principal engineering basis, with cost inputs, such as	
	operating consumables, based on in-house data and benchmarking.	
	Off-site costs such as refining were provided by Focus. Provided by Focus.	
	• Royalties were assessed on a tenement basis. These included royalties for Government (2.5%) and tenement specific royalties. Total	
Revenue factors	royalties vary by tenement and range from 6.5 to 7.5% of revenue. • Gold is the only revenue generating product considered in the Ore Reserves.	
Nevenue lactors	A gold price of AUD 2,207/oz was provided by Focus and confirmed by Mr. Bojanic as reasonable estimate for a long-term price	
	using published metal price forecasts	
Market	The demand for gold is considered in the gold price used.	
Assessment	It was considered that gold will be marketable for beyond the processing life of these Reserves.	
	The commodity is not an industrial metal.	
Economic	An economic model has been prepared from the outcomes of the preliminary engineering and costing associated with the PFS. The	
	economic modelling demonstrates that the Project is cash flow positive.	
	• The base case results in a positive economic outcome as assessed by an NPV calculation (@5.0% DCF). The NPV is most sensitive to	
	the gold price.	
	Focus has advised the Project carries sufficient tax credits to cover forecast tax payable from the PFS. RPMGlobal completed economic	
	analysis on both a pre- and post-tax basis.	
0 ' '	• The project break-even gold price is approximately AUD\$1,856/oz (pre-tax) or AUD\$1,900/oz (post-tax).	
Social	• There is currently a native title application (Nyalpa Pirniku WC2019/002) over most of the Project area. The claim has been accepted to	
	be considered for determination but is yet to be determined (Wood 2020). Traditional owners of the area are the Wongatha people. • Focus holds an Aboriginal Land Access agreement with the Wongatha people, who had a native title claim over the entire Project Area	
	 Focus holds an Aboriginal Land Access agreement with the Wongatha people, who had a native title claim over the entire Project Area The Wongatha claim was dismissed. It is yet to be determined whether the ongoing heritage interest over the Project area by the parties 	
	to the Wongatha agreement will be influenced by the progression of Nyalpa Pirniku native title claim.	
Other	No naturally occurring material risks have been identified through the PFS.	
· 	Mining Leases covering most of the areas to be affected by the proposed operations are in place.	
	Parts of existing and proposed haul roads are not covered by Mining Act tenements so appropriate tenure will need to be sought	
	to facilitate their development and use.	
	The Barnicoat Mill is a prescribed premise (Category 5), licenced under L8490/2010/2, which permits processing of up to 1.5Mt	
	of ore per annum.	
Classification	The Ore Reserve is classified as Probable in accordance with the JORC Code, corresponding to the resource classifications of Measured	
	and Indicated Resources.	
	1	
	There are no Measured Resources at the Project.	
	There are no Measured Resources at the Project. Indicated Resources have been converted to Probable status.	

Criteria	Commentary			
and Reviews	 exploration results, Mineral Resources and Ore Reserves. Within the JORC Code is a "Checklist of Assessment and Reporting Criteria" (Table 1 – JORC Code). This checklist has been used as a systematic method to undertake a review of the underlying Study used to report in accordance with the JORC Code. RPMGlobal has completed an internal review of the Ore Reserve estimate, deriving results using two separate methods, and believes the estimate accurate. 			
Discussion of relative accuracy/confidence	 The proposed gold mine will be employing conventional mining and ore processing techniques. The PFS has been supported by engineering and costing to provide a level of service targeting +/-25% accuracy. The marginal cut-off grades used to derive the Ore Reserve estimates were calculated from the final outcomes of the PFS. The ultimate pit limits were selected based on a Revenue Factor of 85% to provide a 15% margin at the limit and based on Measured and Indicated Resources. Pit designs were undertaken based on the preferred pit shells. Ore Reserve quantities and grades were derived based on the mining model, the cut-off grade and with the detailed ultimate pit shell. An internal audit checked the estimation of quantities. Sensitivity analyses were undertaken on the economic model to test robustness of the economic outcomes The Project is most sensitive to gold price. Un-discounted cash-flows are break-even at a gold price of AUD\$1,753/oz (post tax). The accuracy of the underlying Mineral Resources is defined by the Resource Category that the Mineral Resources are assigned to. Only Indicated Resources have been used for estimating Ore Reserves. Exploration targets have recently been reported in the immediate vicinity of the reported Ore Reserves at the Karridale, Burtville and Beasley Creek South areas. Additional metallurgical test work is recommended to increase the confidence in the performance of the milling circuit and gold recoveries Primary ores from Karridale, Beasley Creek, Beasley Creek South and Wedge have been excluded from the PFS and Ore Reserve. The current reserve pit shell at Karridale extends to the boundary between transitional and fresh material. Further metallurgical testing and studies are required to determine the potential metallurgical properties and likely capital and operating costs for the processing of this material. 			

JORC Code, 2012 Edition – Table 1 Euro Deposit Section 1 Sampling Techniques and Data

	tion apply to all succeeding sections.)
Criteria	Commentary
Sampling techniques	This report relates to results from Reverse Circulation (RC) and diamond core (DDH) drilling.
techniques	• Euro north and South have been drilled by various companies over the years, this report contains information on holes drilled by:
	Ashton Gold Ltd (Ashton) who were part of a joint venture with Dominion Mining Ltd (Dominion), Sons of Gwalia Ltd (SOG),
	Crescent Gold NL (Crescent) and Focus Minerals Ltd (Focus).
	 Ashton collected 1m RC samples via a riffle splitter. A spear sample was also taken of the intervals and 4m composites submitted for analysis Where composite assays exceeded 0.25 ppm Au, the corresponding 1m sample was submitted.
	Ashton recorded duplicate samples in the assay files.
	Dominion submitted 1m RC samples for analysis for the entire drill hole.
	 Crescent collected 1m RC percussion samples in a plastic bag off the drill cyclone. The sample was then put through a 75/25 riffle splitter resulting in a 3-4kg sample that was submitted for analysis. HQ3 diamond core was placed in core trays, marked up, logged geologically and geotechnically then photographed. Core samples were submitted as either 1m samples or to geological contacts from surface to SGS Perth for analysis.
	SOG submitted 1m RC samples from surface.
	The information of sampling techniques below applies to the drill holes drilled by Focus Minerals (FML) from 2019 onward.
	RC percussion drill chips were collected through a cyclone and cone splitter. Samples were collected on a speared 4m composite and cone split 1m basis.
	RC chips were passed through a cone splitter to achieve a sample weight of approximately 3kg.
	4m composite samples were taken by spear or scoop sampling the bulk 1m sample. Where results returned greater than 0.2g/t Au, the 1m samples were submitted.
	 At the assay laboratory all samples were oven dried, crushed to a nominal 10mm using a jaw crusher (core samples only) and weighed. Samples in excess of 3kg in weight were riffle split to achieve a maximum 3kg sample weight before being pulverized to 90% passing 75µm.
Drilling	Only Reverse Circulation (RC) and Diamond (DD) drilling methods have been included in the resource estimate.
techniques	Ashton reports state drilling was by a face sampling hammer RC rig.
	Dominion drilling was by Drillex using an RC rig.
	SOG used a Reverse Circulation drill rig.
	Crescent gold used various drill contractors over the years. Rigs were either RC with face hammer sampling techniques or HQ3
	tube diamond coring rigs.
	All FML, 2019 onward, drilling was completed using an RC face sampling hammer. Most holes were surveyed upon completion of
	drilling initially using an electronic multi-shot (EMS) camera.
Drill sample	Ashton recorded drill sample recovery in their logs as a percentage.
recovery	Dominion did not record sample recovery in their logs.
	SOG did not record sample recovery in the logs

Criteria	Commentary
	Crescent recorded sample recovery in the geology logging and noted samples were recovered dry.
	FML Sample recovery was recorded by a visual estimate during the logging process.
Logging	Ashton logged the entire hole for weathering, rock type, structure, texture, alteration, veining, mineralisation and colour.
	Dominion logged the entire hole for rock type, structure, texture, alteration, veining and mineralisation.
	 Crescent logged the entire drill hole for colour, weathering, regolith, rock type, texture, alteration, veining, mineralisation. Drill core was photographed.
	 Not all the SOG holes have geological logs in the SQL database, limited logging of rock type, texture and alteration has been captured.
	 FML RC samples were geologically logged to record weathering, regolith, rock type, veining, alteration, mineralisation, structure and texture and any other notable features that are present.
	The logging information was transferred into the company's drilling database once the log was complete.
	 Logging was qualitative, however the geologists often recorded quantitative mineral percentage ranges for the sulphide minerals present.
	The entire length of all holes is logged.
Sub-sampling techniques and sample	 Ashton submitted 4m composite samples to SGS Kalgoorlie, samples were dried, jaw crushed, hammer milled, split and pulverised. Samples were analysed for gold by fire assay on a 50g charge to a lower limit of detection of 0.01 ppm Au. Where the composite assay exceeded 0.25ppm, the relevant 1m interval was submitted to SGS for analysis.
preparation	Dominion submitted 1m samples from surface to Genalysis in Kalgoorlie for Au analysis by 50g Fire Assay.
	 SOG submitted 1m RC samples from surface to LLAL Leonora for analysis by Aqua Regia and leachwell (CN leach) on all samples returning assays above 0.6g/t Au.
	 Crescent submitted 1m RC samples from surface to SGS Leonora for Au analysis by 50g fire assay with AAS finish. Diamond samples were submitted as either 1m intervals or to geological contacts to SGS Perth for Au analysis by 50g fire assay.
	• FML RC samples were cone split to a nominal 3 - 5kg sample weight. The drilling method was designed to maximise sample recovery and delivery of a clean, representative sample into the calico bag.
	 Where possible all RC samples were drilled dry to maximise recovery. The use of a booster and auxiliary compressor provide dry sample for depths below the water table. Sample condition and recovery percentage was recorded (wet, dry, or damp) at the time of
	sampling and recorded in the database.
	 The samples were collected in a pre-numbered calico bag bearing a unique sample ID. Samples were crushed to 75µm at the laboratory and riffle split (if required) to a maximum 3kg sample weight. Gold analysis was initially by 40g aqua regia for the composite samples then 30g Fire Assay for individual samples with an ICP-OES or AAS Finish.
	• The assay laboratories' sample preparation procedures follow industry best practice, with techniques and practices that are appropriate for this style of mineralisation. Pulp duplicates were taken at the pulverising stage and selective repeats conducted at the laboratories' discretion.
Quality of assay data and	Dominion drill logs contain multiple assays for some mineralised intervals suggesting duplicates or repeats of higher-grade assays was undertaken.
laboratory tests	 As part of the Dominion RC drill campaign in 1994, a high-grade Ashton drilled intersection of mineralisation was followed up. Dominion confirmed the mineralisation width and assay results with similar return.
	 SOG submitted resamples from every hole at an approximate ratio of 1 every 20m drilled as a quality check. Returned assays verified the original assays.
	Crescent utilised numerous checks for the quality of its assay data taking field duplicate samples, submitting standard reference
	samples, laboratory check assays, leachwell analysis, BLEG analysis and reviewing the laboratory quality control reports.
	 Earlier FML QAQC checks involved inserting a standard or blank every 20 samples in RC and taking a field duplicate every 20 samples in RC. Field duplicates were collected from the cone splitter on the rig. A minimum of 3 standards were inserted for every sample batch
	 submitted. Sampling was carried out by the supervising geologist and senior field staff, to ensure all procedures were followed and best industry
	 The sample sizes are considered to be appropriate for the type, style and consistency of mineralisation encountered during this
	phase of exploration. Laboratory repeat checks were also run on the assay data.
Verification of	Historic sampling and assaying have been checked against hard copy WAMEX reports or company reports.
sampling and assaying	 No adjustments were made to any current or historic data. If data could not be validated to a reasonable level of certainty it was not used in any resource estimations.
Location of data	Ashton collars surveyed by Mt Morgan Mine Surveyors and reported in local grid.
points	Dominion state all RC holes were surveyed using a theodolite.
	SOG have reports from contract surveying companies for the resurvey of the tenement boundary, establishment of a new Euro local grid and verification of previously drilled holes by re-surveying their collars. Drill collars were surveyed in local Euro grid and also
	converted to AMG co-ordinates
	Crescent surveyed drill collars in MGA94 Zone 51 grid co-ordinates using site survey personnel. Downhole surveys were taken by either an electronic multi-shot camera or gyroscope tool by Surtron Technologies Pty Ltd. Description of the survey were taken by either an electronic multi-shot camera or gyroscope tool by Surtron Technologies Pty Ltd. Description of the survey were taken by either an electronic multi-shot camera or gyroscope tool by Surtron Technologies Pty Ltd. Description of the survey were taken by either an electronic multi-shot camera or gyroscope tool by Surtron Technologies Pty Ltd. Description of the survey were taken by either an electronic multi-shot camera or gyroscope tool by Surtron Technologies Pty Ltd. Description of the survey were taken by either an electronic multi-shot camera or gyroscope tool by Surtron Technologies Pty Ltd. Description of the survey were taken by either an electronic multi-shot camera or gyroscope tool by Surtron Technologies Pty Ltd. Description of the survey were taken by either an electronic multi-shot camera or gyroscope tool by Surtron Technologies Pty Ltd. Description of the survey were taken by either an electronic multi-shot camera or gyroscope tool by Surtron Technologies Pty Ltd. Description of the survey were taken by electronic multi-shot camera or gyroscope tool by Surtron Technologies Pty Ltd. Description of the survey were taken by electronic multi-shot camera or gyroscope tool by survey were taken by electronic multi-shot camera or gyroscope tool by ele
	• FML drill collars were surveyed after completion, using a DGPS instrument. Most holes were surveyed upon completion of drilling. An electronic multi-shot camera was used, holes were surveyed open hole.

Criteria	Commentary
	All coordinates and bearings use the MGA94 Zone 51 grid system.
	Historic holes have been converted to MGA94 Zone 51 grid system in Acquire.
	Focus utilises Landgate sourced regional topographic maps and contours as well as internally produced survey pick-ups produced by
	the mining survey teams utilising DGPS base station instruments. Historic drill collars were assessed to see if they plotted on the
	topographic maps within an acceptable tolerance.
Data spacing and	Drill spacing along the Euro South deposit is quite regular at a 25m x 10m - 15m spaced pattern along strike.
distribution	Drill spacing along the Euro North trend is 30m x 30m with the average depth of RC holes 82m below surface.
Orientation of	Drilling was designed based on known geological models, field mapping, verified historical data and cross-sectional interpretation.
data in relation to	Drill holes were either vertical or oriented at right angles to strike of deposit, with dip optimised for drill capabilities and the dip of the
geological structure	ore body.
Sample security	Historic sample security is not recorded.
	FML samples were reconciled against the sample submission with any omissions or variations reported.
	All samples were bagged in a tied numbered calico bag, grouped into green plastic bags. The bags were placed into cages or bulk
	bags or pods with a sample submission sheet and delivered directly from site to the Kalgoorlie laboratories by FML personnel.
Audits or reviews	In March 2004 Apollo Gold Mining Ltd validated hard copy company reports and compact discs from previous tenement holders
	against the Apollo held drill database. A visual check of the original company data against the database compared, location co-
	ordinates, down hole survey readings and assays.
	Euro South was mined by Crescent Gold between 2009 and 2010, where monthly reconciliations were undertaken.

Section 2 Reporting of Exploration Results (Criteria listed in the preceding section also apply to this section.) Criteria Commentary

Criteria	Commentary				
Mineral	Euro deposit is loc	cated within Mining Lease M38/143 and M38/143 which are registered to	Focus Minerals	Ltd. and Focus	
tenement and	Operations Pty Lt	d of Perth, Western Australia.			
land tenure	The Nyalpa Pirnik	u claim cover the Laverton Project tenure. At this stage no Laverton clair	ms have progre	ssed to determined s	tatus.
status					
Exploration		en historically mined since 1895 when gold was first discovered in the regi			
done by other	from 1898 to 1911	with a recorded production of 35,707 ounces of gold from 94,826 tonne	s of ore at an av	verage grade of 11.7	g/t Au.
parties	 Euro South was n 	nined by Crescent Gold from an open pit between 2009 and 2010 produci	ing 843kt at 1.4	2g/t Au for 38.6 koz.	
Geology	The Euro trend co	overs a sequence of tholeiitic meta-basalt flows with intrusive dolerite plus	minor felsic po	rphyries and	
		nded chert interflows sediments. The prospect is bound to the west by the			
		he east and N-0S striking AMAG high that may be another BIF sequence		•	nce
		d conglomerates, unconformably overlying the eastern side of the package			
	• •	e clastics units may be a local equivalent of Wallaby type conglomerates	•	•	
		erate west dipping faults have been sericite-carb-chl altered. Mineralisat			
	-	o than quartz veining.	io ou origiy t	TOTO SIGNOG TITLE	
Drill hole		· · · · · · · · · · · · · · · · · · ·	Not all drill bala	s oon he found refere	nood
Information	_	ormation has been validated against publicly available WAMEX reports.			
in or madon	· ·	ports. However, cross-checking of original drill surveys was verified again	ist the database	e. Most of these notes	s were
	drilled in the exca	vated pit area and has been depleted from the reported resource.			
	EURO SOUTH WAM	EV Deference:			
	EURU SUUTH WAIVI	EX Relefence:	WAMEY		1
			WAMEX	WAMEX Report	
	Company	Drill Hole ID	Report A- Number	Date	
	Ashton		17955	January 1990	
		E48, E49, E50, E51, E53, E54, E55, E57, E58 E62, E63, E64, E65, E66, E67, E68, E69, E70, E71, E72, E74, E75	17333	January 1990	
		E02, E03, E04, E03, E00, E07, E00, E09, E70, E71, E72, E74, E73	33653	February 1991	
		E80, E81, E82, E83, E84, E85	35633	December 1991	
	Dominion	EURC004, EURC005, EURC006, EURC007, EURC008,	43121	November 1994	
		EURC009, EURC011, EURC012, EURC013, EURC014,			
		EURC015, EURC016, EURC017, EURC018, EURC019,			
		EURC021, EURC022, EURC023, EURC024, EURC025,			
		EURC026, EURC027, EURC028			
	Crescent Gold NL	GTDE5, GTDE7	69877	November 2004	
		SL02, SL04, SL05, SL07, SL09, SL12, SL13, SL14_SOUTHLAV,			
		SL15_SOUTHLAV, SL16_SOUTHLAV, SL17_SOUTHLAV,			
		SL18_SOUTHLAV, SL19_SOUTHLAV, SL20_SOUTHLAV,			
		SL21_SOUTHLAV, SL22_SOUTHLAV, SL24, SL25,			
		SL26_SOUTHLAV, SL27_SOUTHLAV, SL29, SL30_SOUTHLAV,			
		SL31_SOUTHLAV, SL32, SL33_SOUTHLAV, SL34_SOUTHLAV,			
		SL35_SOUTHLAV, SL36_SOUTHLAV, SL37_SOUTHLAV, SL40_SOUTHLAV, SL41_SOUTHLAV, SL42_SOUTHLAV,			
		SL40_SOUTHLAV, SL41_SOUTHLAV, SL42_SOUTHLAV, SL50,			
		SL51, SL55, SL57, SL58, SL59, SL60, SL62, SL63, SL64, SL65,			
		SL66, SL67, SL68, SL69, SL70			
	L	1,,,,			_

Criteria	Commentary							
		EURRC0		TIBBOOK TO	DD064 E::=	2005 51155111	74177	December 2006
						0005, EUDD006, C102, EURC103,	81229	February 2009
						C102, EURC103,		
		EURC110), EURC111, E	EURC112, EU	RC113, EURC	C114, EURC115,		
						C121, EURC122,		
		EURC123	3, EURRC007, 12, EURRC			:URRC011, RRC016,		
		EURRC0				RRC020,		
		EURRC0		,	,	RRC024,		
		EURRC0			DO 100 EUD	2400 =1150404		
		EURC124	4, EURC125, I	EURC126, EU	RC128, EUR	C129, EURC131	86387	February 2010
	0-11 1-4-114.004	N I - I						
	Collar details of SO	noies arilled	auring 1996 a					
				MGA 94 Zo			Depth	
	Hole ID	Easting	Northing	RL	Azimuth	Dip	(m)	Tenement
	ERC002	441456.18		460.146	70	0	-90	M3800143
	ERC003	441477.68		460	25	0	-90	M3800143
	ERC004	441381.09	6821022.6	462.361	101	0	-90	M3800143
	ERC004A	441382.64	6821022.8	462.384	90	91	-60	M3800143
	ERC005A	441431.59	6821023.2	461.809	45	91	-61.5	M3800143
	ERC006	441331	6821073.5	462.729	95	92	-60	M3800143
	ERC007	441354.51	6821074.6	463.149	80	90	-60	M3800143
	ERC008	441282.66	6821198.1	465.38	80	92	-62	M3800143
	ERC009	441308.05	6821198.5	465.953	60	91	-63	M3800143
	ERC010	441283.61	6821296	470.578	65	87	-59	M3800143
	ERC011	441308.66	6821295.7	471.79	50	94	-60	M3800143
	ERC012	441309.27		474.628	50	92	-60	M3800143
	ERC013	441333.98		474.129	50	92	-60	M3800143
	ERC015	441354.59		473.314	50	92	-60.5	M3800143
	ERC016	441309.09		476.744	50	92	-60.5	M3801187
	ERC017	441285.55		475.13	75	92	-60	M3800143
	ERC018	441406.09		461.593	85	94.5	-61.5	M3800143
	ERC019	441431.15		461.236	60	94	-60	M3800143
	ERC020	441455.46		460.937	40	99	-61	M3800143
	ERC021	441356.38		462.658	105	93	-60	M3800143
	ERC021		6821047.9	463.026	85	93	-60.5	M3800143
								M3800143
	ERC023		6821048.6	463.096	65 45	90	-60.5	
	ERC024	441431.99		462.675	45	91	-59 60	M3800143
	ERC025	441331.92		463.398	95	90	-60	M3800143
	ERC026	441356.47		463.708	80	93	-60	M3800143
	ERC027	441382.6	6821097.5	464.037	65	91	-60	M3800143
	ERC028	441405.82		464.352	45	95	-59.5	M3800143
	ERC030	441456.38		460	50	94	-59	M3800143
	ERC031	441480.01		459.644	36	95	-57	M3800143
	ERC032	441272.58		471.037	75	95	-59	M3800143
	ERC033	441308.9	6821348.8	473.744	45	92	-59	M3800143
	ERC034	441332.75		471.671	30	95	-57.5	M3800143
	ERC035	441259.95		471.296	50	92	-60	M3800143
	ERC036	441282.89	6821393.4	473.35	50	92	-59	M3800143
	ERC037	441309.47	6821396.4	473.361	50	91	-60	M3800143
	ERC038	441333.97	6821397.5	472.498	30	93	-62	M3800143
	ERC039	441260.64	6821450.3	472.137	49	90	-58	M3800143
	ERC040	441284.43	6821450.4	474.175	75	97	-62	M3800143
	ERC041	441301.95	6821450	476	50	90	-62	M3800143
	ERC042	441334.7	6821451.3	475	50	90	-62	M3800143
	ERC043	441259.47		471.916	40	92	-60.5	M3800143
	21.00-10		3321100			, v <u>-</u>	55.0	

Criteria	Commentary									
	ERC044	441285.8	6821495.6	475.538	30	95	-60	M38	300143	
	ERC049	441286.78		468.59	35	80	-61.5		300143	
	ERC051	441444.89		459.883	75	91	-60.5	M38	300143	
	ERC052	441469.21		459.436	50	91	-61.5		300143	
	ERC054	441455.03		459.365	75	90	-59.5		300143	
	ERC057	441468.15		459	75	90	-60		300143	
	ERC060	441333.71		467.046	90	93	-60		300143	
	ERC061	441358.21		467.152	65	95	-58.5		300143	
	ERC062		6821322.1	472.437	70	90	-60		300143	
	ERC063	441335.52	6821322.7	471.885	38	98	-58	M38	300143	
	ERC066		6821472.1	475.117	30	90	-60	M38	300143	
	EURO NORTH WA	AMEX Referen	ce:							
	Company	,		Drill Hol	e Number		WAMEX Rep Numb		WAMEX Report Dat	
	Hillmin Gold Mine		3, E4, E8	Dilli ilo	C Hullibei		20642		Feb-87	
		E		E23, E24, E25	, E15, E16, E1 5, E26, E27, E 6, E37		28072		Dec-88	
	Ashton Gold Mine	É	49, E50, E51,	E52, E53, E5	3, E44, E45, E 4, E55, E56, E	57, E58, E59	17955		Jan-90	
	0 1011	E	71, E72, E73,	E74, E75, E7	•		33653		Feb-91	
	Crescent Gold	S	L93), SL91, SL92,	69877		Nov-04	
	Focus Miner	E	URRC005, EL	JRRC006	RRC003, EUR		74177		Dec-06	
	Focus Miner	E	EURC158, EURC159, EURC160, EURC161, EURC162, EURC163, EURC164, EURC165, EURC166, EURC167, EURC168, EURC169, EURC170, EURC171				98404		Jun-13	
	Holes not availa	able through W	VAMEX but pre	eviously report	ed:					
	Company	у			le Number		ASX Release		ASX Release I	
	Company	У		1, 21EURC002 5, 21EURC00	ole Number 2, 21EURC003	3, 21EURC004, 3, 21EURC009,	ASX Release Exploration Up Laverton Gold	pdate -	ASX Release I 29 October 202	
			21EURC009 21EURC010	1, 21EURC00 5, 21EURC00)	ole Number 2, 21EURC003 7, 21EURC008	3, 21EURC009,	Exploration Up Laverton Gold	pdate - Project		
	The details of Focu	us Minerals dri	21EURC009 21EURC010 lled RC holes	1, 21EURC00: 5, 21EURC00:) in 2019 and 20	ole Number 2, 21EURC003 7, 21EURC008 021 not previou	3, 21EURC009,	Exploration Up Laverton Gold	pdate - Project		219
	The details of Focu HOLEID 21EURC006	us Minerals dri EAST 441032	21EURC000 21EURC010 Illed RC holes NC 6822	1, 21EURC00: 5, 21EURC00:) in 2019 and 20 DRTH	ole Number 2, 21EURC008 7, 21EURC008 021 not previou RL 472	usly reported are to AZIMUTH 88	Exploration Up Laverton Gold abulated below:	pdate - Project IP 60	29 October 202 DEPTH (m 126	219
	The details of Focu HOLEID 21EURC006 21EURC011	us Minerals drii EAST 441032 441046	21EURC000 21EURC010 Illed RC holes NC 6822 6822	1, 21EURC00: 5, 21EURC00: 0 in 2019 and 20 DRTH 084 091	De Number 2, 21EURC008 7, 21EURC008 D21 not previou RL 472 472	usly reported are to AZIMUTH 88 92	Exploration Up Laverton Gold abulated below:	Project IP 60 60	29 October 202 DEPTH (m 126 60	219
	The details of Focu HOLEID 21EURC006 21EURC011 19EURC001	us Minerals dri EAST 441032 441046 441107	21EURC000 21EURC010 1 lled RC holes 80 6822 6821	1, 21EURC00: 5, 21EURC00:) in 2019 and 20 DRTH 084 091 930	ole Number 2, 21EURC003 7, 21EURC008 021 not previou RL 472 472 467	21EURC009, usly reported are to AZIMUTH 88 92 81	Exploration Up Laverton Gold abulated below:	pdate - Project IP 60 60 71	29 October 202 DEPTH (m 126 60 168	219
	The details of Focu HOLEID 21EURC006 21EURC011 19EURC001 19EURC002	us Minerals drii EAST 441032 441046 441107 441049	21EURC000 21EURC010 1 Ed RC holes 8822 6821 6821	1, 21EURC00: 5, 21EURC00:) in 2019 and 20 DRTH 084 091 930 969	ble Number 2, 21EURC008 7, 21EURC008 021 not previou RL 472 472 467 470	3, 21EURC009, usly reported are to AZIMUTH 88 92 81 63	Exploration Up Laverton Gold abulated below:	Project Project IP 60 60 71	29 October 202 DEPTH (m 126 60 168 180	219
	The details of Focu HOLEID 21EURC006 21EURC011 19EURC001 19EURC002 19EURC003	EAST 441032 441046 441107 441049 441021	21EURC000 21EURC010 1lled RC holes 8822 6822 6821 6821 6822	1, 21EURC00: 5, 21EURC00: 0 in 2019 and 20 0RTH 084 091 930 969 026	Delle Number 2, 21EURC008 7, 21EURC008 D21 not previou RL 472 467 470 471	23, 21EURC009, usly reported are to AZIMUTH 88 92 81 63 81	Exploration Up Laverton Gold abulated below: Di -(Project Projec	29 October 202 DEPTH (m 126 60 168 180 204	219
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aggregation methods Relationship between mineralisation widths and intercept lengths Diagrams	The details of Focu HOLEID 21EURC006 21EURC011 19EURC002 19EURC003 19EURC004 19EURC005 19EURC006 19EURC007 19EURC008 19EURC009 • Mineralised interto 1m. • Holes were drill width cannot be	## Minerals drii ## EAST ## 441032 ## 441046 ## 44107 ## 44094 ## 440932 ## 440933 ## 40992 ## 40859 ## ersections are ## ed orthogonal ## e estimated ex ## and Tables in	21EURC000 21EURC010 NC	1, 21EURC00: 5, 21EURC00: 0 in 2019 and 20 DRTH 084 091 930 969 026 063 112 174 227 290 242 0.5g/t Au cut-o ion as much a es.	Della Number 2, 21EURC003 7, 21EURC008 D21 not previous RL 472 467 470 471 471 471 472 471 472 471 471	Section Sect	Exploration Up Laverton Gold abulated below:	pdate - Project IP 60 60 71 61 60 63 77 62 59 diamond	29 October 202 DEPTH (m 126 60 168 180 204 174 167 192 186 198 210 holes, composit	219 nn)
aggregation methods Relationship between mineralisation widths and intercept lengths	The details of Focu HOLEID 21EURC006 21EURC011 19EURC001 19EURC002 19EURC003 19EURC005 19EURC006 19EURC007 19EURC008 19EURC009 • Mineralised interto 1m. • Holes were drill width cannot be expected to 1m.	## Minerals dri ## EAST ## 41032 ## 41046 ## 41049 ## 41021 ## 40962 ## 40932 ## 40910 ## 40892 ## 40859 ## ersections are ## and Tables in the results available in the results availab	21EURC000 21EURC010 1	1, 21EURC00: 5, 21EURC00: 0) in 2019 and 20 0RTH 084 091 930 969 026 063 112 174 227 290 242 0.5g/t Au cut-o ion as much ales.	ble Number 2, 21EURC008 7, 21EURC008 021 not previous RL 472 467 470 471 470 471 471 472 471 ff, composited	Section Sect	Exploration Up Laverton Gold abulated below:	pdate - Project IP 60 60 71 70 71 61 60 63 77 62 59 diamond	29 October 202 DEPTH (m	219 nn)

Criteria	Commentary
	company data.
Other	There is no other material exploration data to report.
substantive	
exploration data	
Further work	The company is further reviewing the exploration results.

Section 3 Estimation and Reporting of Mineral Resources

	ion 1, and where relevant in section 2, also apply to this section)
Criteria	Commentary
Database integrity	FML's database is a Microsoft SQL Server database (acQuire), which is case sensitive, relational and normalised to the Third Normal Form. Because of normalisation, the following data integrity categories exist:
	Entity Integrity: no duplicate rows in a table, eliminated redundancy and chance of error.
	Domain Integrity: Enforces valid entries for a given column by restricting the type, the format or a range of values.
	Referential Integrity: Rows cannot be deleted which are used by other records.
	User-Defined Integrity: business rules enforced by acQuire and validation codes set up by FML.
	Additionally, in-house validation scripts are routinely run in acQuire on FML's database and they include the following checks:
	Missing collar information
	Missing logging, sampling, downhole survey data and hole diameter
	Overlapping intervals in geological logging, sampling, down hole surveys
	Checks for character data in numeric fields
	Data extracted from the database were validated visually in GEOVIA Surpac software and ARANZ Geo Leapfrog software. Also,
	when loading the data any errors regarding missing values and overlaps are highlighted.
011 1 11	Historic data has been validated against WAMEX reports where possible.
Site visits	Alex Aaltonen, the Competent Person for Sections 1 and 2 of Table 1 is FML's General Manager - Exploration and conducts regular site visits.
	 Hannah Kosovich, the Competent Person for Section 3 of Table 1 is FML's Resource Geologist and last visited site in September 2019.
Geological	All available drill hole and historic mining data was used to guide the geological interpretation of the mineralisation.
interpretation	Knowledge and information generated from the proximal Euro mining operations also guided the interpretation.
	An approximate cut-off grade of 0.4g/t was implemented.
	The mineralised geological interpretation was constructed in Seequent Leapfrog Geo software on a sectional basis.
	At Euro South 21 stacked moderate west dipping lodes were modelled. The main zone of Euro South mineralisation comprises 3
	longer strike WSW dipping lodes that extend over much of the open pit.
	Euro North consists of two main NNW striking, WSW dipping main lodes. An additional 5 minor lodes shallow westerly dipping lodes were identified and modelled.
	Voids from historic underground mining were modelled.
Dimensions	The entire Euro trend strikes NNW over more than 1.7km
	Euro North sits approx. 180m to the NNW of the current Euro Pit.
	Euro North has been modelled over a 480m strike; lodes have been interpreted from near surface to approximately 185m below
	surface. The average thickness of the main lodes is 3m and the minor lodes 2.5m.
	The Euro South deposit has been interpreted to trend towards the NNW over 780m strike. Mineralisation has been
	modelled from surface to approx. 120m below surface.
	The width of interpreted mineralization varies from 1m to approx. 11m, with an average width of 4m.
Estimation and modelling	The drill hole samples were composited to 1m within each domain. This is the dominant sampling interval.
techniques	 All domain boundaries were considered "hard" boundaries and no drill hole information was used by another domain in the estimation.
	Composited assay values of each domain were imported into Snowden Supervisor for geostatistical analysis.
	A review of histograms, probability plots and mean/variance plots for each domain revealed some outlier sample values.
	Top capping of higher Au values within each domain was carried out with Au values above the cut-off grade reset to the cut-off grade. Not all lodes were top-cut.
	The different lodes have different top-cuts as required, a maximum top-cap of 15ppm was used with an average of 7-8ppm.
	Variograms were modelled in Supervisor on the larger domains that had over 100 samples and this variogram was then shared with the other lodes of similar orientation and proximity. Due to the skewed nature of the dataset a Normal Scores transformation was applied to obtain better variograms. A back-transformation was then applied before being exported.
	At Euro North one variogram was modelled and had moderate nugget effect ~ 35% of total sill and a down plunge range of 200m for the main N-S lodes, across dip was small, 10m.
	At Euro 6 variograms were modelled, the variograms had a moderate nugget effect ~ 25% up to 52% of the total sill and a down
	plunge range of up to 60m for the main NNW cross-cutting lodes, across dip was small, 10m.
	 No "unfolding" of the mineralised wireframes was required. Datamine Software was used for the estimation and block modelling process. The model was created in GDA 94 grid co-ordinates. Block sizes for the model were 10m in Y, 10m in X and 5m in Z direction. Sub celling of the parent blocks was permitted to 2.5m in

Criteria	Commentary
	 the Y direction, 2.5m in the X direction and 1.25m in the Z direction. Sub-blocking was used to best fill the wireframes and inherit the grade of the parent block. No rotation of the block model orientation was applied. An Ordinary Kriging (OK) estimation technique was selected and used the variograms modelled in Supervisor. Each domain was estimated separately using only its own sample values.
	At Euro North with less drill density a "grade restricted search" method was used, whereby high- grades have reduced search distances. This helps to reduce the spread of higher values into areas of low sampling.
	Minimum (6-8) and maximum (14-16) sample numbers was selected for the first estimation pass, this was dropped to a minimum (4) samples on the second and third search pass.
	An elliptical search was used based on range of the Variograms. The different lodes had different search ellipses modelled based on their individual orientations.
	After the first estimation pass and second and third pass were run to ensure all mineralised blocks estimated. The search distance was doubled between the first and second search pass and doubled again between the second and third search pass.
	 Euro South after the first pass, 73% of blocks had estimated, 23% in the second and 4% of blocks estimated in the third pass. Euro North 71% of blocks had estimated in the first pass, 29% in the second and 0.2% of blocks estimated in the third pass.
	The estimate was validated by a number of methods. An initial visual review was done by comparing estimated blocks and raw drill holes.
	 Tonnage weighted mean grades were compared for all lodes, there were no major differences. Swath plots of drill hole values and estimated Au grades by northing, easting and RL were generated for all domains in Supervisor software and showed that the estimated grades honoured the trend of the drilling data.
Moisture	Tonnages are estimated on a dry basis.
Cut-off parameters	The mineral resource for the Euro deposits has been reported above a 0.6g/t Au cut-off. This figure is based on recent Feasibility studies.
Mining factors or assumptions	The Euro deposit would be mined by a cut-back on the existing open pit.
Metallurgical factors or assumptions	 Metallurgical test work was carried out by AMMTEC in June 2004 on a composite of Euro ore samples. Three different gold extraction tests were run (direct leach, gravity and CIL leach) all with reasonably high gold recoveries or 98.3 for direct leach, 97.9 for gravity and 91.9 for CIL leach. When mined Euro ore was blended with other sources so actually recoveries are unknown.
Environmental factors or assumptions	 Euro has been historically mined by open pit methods and associated ground disturbances such as haul roads and waste dumps exist in the area. There are no unforeseen environmental considerations that would prevent open pit mining from re-commencing in the area.
Bulk density	• Density values were assigned based on weathering surfaces generated from the logging and were based on values historically used in the area and test work conducted on drill core by Crescent Gold. Oxide = 1.80 t/m³, Transitional = 2.4 t/m³, Fresh = 2.75 t/m³
Classification	Material has been classified Indicated and Inferred based on a number of criteria such as geological continuity, drill hole spacing, estimation pass and proximity to the existing open pit.
Audits or reviews	No external audits of the mineral resource have been conducted.
Discussion of relative accuracy/ confidence	 This is addressed in the relevant paragraph on Classification above. The Mineral Resource relates to global tonnage and grade estimates

JORC Code, 2012 Edition - Table 1 Craigiemore - Mary Mac Trend, West Laverton -**Bulldog Trend, Chatterbox Trend and Gladiator Trend Follows**

Section 1 Sampling Techniques and Data (Criteria in this section apply to all succeeding sections.)

Criteria	Commentary
Sampling techniques	 This report relates to results from Reverse Circulation (RC) and diamond core (DDH) drilling. Unless specifically mentioned Chatterbox deposits Apollo - Whisper, Eclipse – Garden Well, Innuendo and Rumor are referred to the Chatterbox in this table. West Laverton refers to the deposits West Laverton, Rega and Bulldog in this table. Gladiator deposits Gladiator Pit, Murrays, Cousin Murray and Gladiator West are referred to as Gladiator in this table. Deposits Craigiemore, Mary Mac, Mary Mac North and Golden Pinnacles trend are referred to as CM/MM in this report. The deposits covered in this release have been drilled by various companies over the years. Most companies held multiple tenements during their tenure with similar drill practices were applied at each deposit. This includes Focus Minerals Ltd (FML), Crescent Gold NL (Crescent), Metex Resources (Metex) and its Laverton Exploration Joint Venture (LEJV) with Delta Gold NL(DGL) and Placer Dome Asia Pacific (PDAP), Sons of Gwalia Ltd (SOG), Western Mining Corporation (WMC), Hillmin Gold Mines Pty Ltd (Hillmin), which was renamed Ashton Gold Mines Pty Ltd (Ashton) in October 1989. This was dissolved in December 1990 with all rights and obligations assumed by Ashton Gold (WA) Ltd. Chatterbox Trend was drilled by FML, Crescent, Metex/LEJV and WMC. West Laverton trend was drilled by FML, Crescent, SOG and Hillmin/Ashton. Gladiator trend was drilled by FML, SOG, WMC, Hillmin/Ashton, Metex/LEJV, Teck Explorations Ltd, Technomin Australia NL

Criteria	Commentary
	CM/MM trend was drilled by FML, Crescent, SOG and Hillmin/Ashton.
	Early Crescent RC holes were sampled at 1m intervals with the sample from the cyclone being collected in a plastic bag then put
	through a 75/25 riffle splitter, resulting in a 3-4kg sample. Later, larger programs collected 1m RC samples automatically using a cone
	splitter off the drill rig producing 3kg samples.
	Crescent diamond core was sampled across geologically identified zones of mineralisation, the sample lengths varied between a
	minimum of 0.1m and a maximum of 1.3m. The diamond core was marked up for sampling by the supervising geologist during the core
	logging process, with sample intervals determined by the presence of lithology, alteration, and where applicable core loss. The core
	was cut in half using a core saw and the same half of the core (RHS looking downhole) was routinely sent to the laboratory for
	analysis. Infrequent whole core samples were submitted at CM/MM.
	• FML and more recent Crescent RC percussion drill chips were collected through a cone splitter from the drill rig. The bulk sample from
	drilling was placed in neat rows directly on the ground (not bagged) with the nominal 2-3kg calico split sub- sample placed on top of the corresponding pile. RC chips were passed through a cone splitter to achieve a nominal sample weight of approximately 3kg. The
	splitter was levelled at the beginning of each hole. Geological logging defined whether a sample was to be submitted as a 1m cone
	split sample or a 4m spear composite sample. Split samples (1m) were transferred to sample numbered calico bags for submission to
	the laboratory. Composite samples were spear sampled using a scoop to obtain a small representative sample and deposited into
	numbered sample bags. Mineralised 4m composite sampled where resampled at 1m intervals using stored original 1m cyclone split
	samples.
	• FML diamond core was sampled across identified zones of mineralisation and vary from 0.2m to a maximum of 1.2m. The core was
	cut in half using a core saw and the ½ core samples submitted for assay.
	WMC RC samples were collected on 1m intervals.
	Hillmin/Ashton collected 1m RC samples via a riffle splitter, some programs also concurrently collected 4m composite samples. Where
	composite assays exceeded 0.1 ppm Au, the corresponding 1m samples for the entire composite interval were submitted for assay.
	Hillmin/Ashton recorded duplicate samples in the assay files. • Hillmin diamond core was sampled after diamond sawing to ½ core and mineralised intervals sampled to lithological contacts while non
	ore grade host rock was submitted as 4m filleted composites.
	Ashton diamond drilling was either with an RC pre-collar followed by HQ diamond coring or PQ diamond core from surface, which was
	reduced to HQ in earlier holes. Diamond core was either quarter or half core sampled in 1m intervals within the mineralised zones or
	composited to 4m outside known mineralisation zones.
	Teck Exploration collected samples in 1m intervals that were composited to 2m for analysis with anomalous values and/or chert
	intersections assayed at 1m intervals.
	Technomin submitted 1m or 2m samples for analysis.
	SOG RC holes were sampled as 1m samples from surface using a riffle splitter to generate ~ 3kg samples with later programs
	collecting samples at 3m-4m composites and submitting 1m split samples where "significant gold" was intersected.
	 Metex / LEJV collected RC samples in 1m intervals in plastic bags. All dry sample were riffle split to return a representative 1m split sample for analysis. Any wet/Moist samples where 50mm PVC spear sampled. Samples were 4m composites with corresponding 1m
	intervals resampled via the same method from composites that returned assay values greater than 0.1ppm.
	 Metex Diamond holes had an RC pre-collar that was generally composite sampled in 4m intervals, the core was half core samples with
	sample lengths from only a handful of 4m composites to 0.5m length with the majority of core sampled to 1m intervals.
Drilling	Only Reverse Circulation (RC) and Diamond drilling (DD) methods have been included in the resource estimate.
techniques	• FML RC drilling was conducted using a 5 3/8inch face sampling hammer for RC drilling or NQ2 triple tube diamond drilling. At hole
	completion, downhole surveys for RC holes were completed at 30m intervals using a True North Seeking Gyro tool.
	Crescent completed RC using a face sampling hammer or HQ diamond drilling.
	Hillmin used rotary mud pre-collars or existing RC holes for its diamond drilling using a PQ or HQ diameter drill bit.
	Ashton RC reports state drilling was by a face sampling hammer RC rig. A https://dx.doi.org/10.1001/j.j.ge/10.1001/j.j.ge/10.1001/j.j.ge/10.1001/j.j.ge/10.1001/j.j.ge/10.1001/j.j.ge/10.1001/j.j.ge/10.1001/j.j.ge/10.1001/
	 Ashton used a PQ or HQ diameter drill bit, with coring either from surface or with an RC pre-collar. SOG used RC face sampling hammer drilling techniques.
	 SOG used RC face sampling hammer drilling techniques. Metex/LEJV RC drilling was conducted using 5 3/8inch bits and face sampling hammers with 900cfm/350psi of air boosted to
	1200cfm/700psi where necessary by an auxiliary compressor.
	Metex Diamond drilling was by NQ sized core barrels at Gladiator and PQ or HQ triple tube core barrels at Chatterbox all with RC pre-
	collars.
	Metex reported that WMC RC holes were drilled using a conventional cross-over sub.
	Teck used a variety of RC drilling hammers depending on the rock type using a Schramm rig with 425cfm/250 psi.
Drill sample	Historic RC drill sample recovery is not well documented.
recovery	FML/Crescent RC sample recovery was recorded by a visual estimate % during the logging process.
	Crescent diamond core recovery was reported as a percentage of the core run.
	• FML diamond core recovery was measured and recorded as a percentage of the core "run". That is, the length of core between the run
	blocks against the increase in hole depth.
	 Hillmin early RC drill logs do not document drill recovery, however later drill logs have a percentage estimate recorded. Hillmin Diamond core recovery is recorded as a % of the core in the drill logs and varies from 73% to 100% with majority of recovery.
	 Hillmin Diamond core recovery is recorded as a % of the core in the drill logs and varies from 73% to 100% with majority of recovery above 90%.
	 Ashton Diamond core recovery is recorded as a % of the core in the drill logs and overall was good.
	Metex/LEJV sample recovery is not well documented in their WAMEX reports. In a Chatterbox report diamond core recoveries were
	generally good. Core loss was recorded in limited areas with significant jointing/fractures or weathered clays.
	Along the Chatterbox trend the high water table issues prevalent at Beasley Creek also impacted samples.
	Work by Crescent in 2011 to establish unreliable samples based on logging of Wet samples or poor recovery from sample weights were

Criteria	Commentary
	flagged and excluded from the Resource estimate.
	Metex developed a sample quality matrix to log sample return and moisture when logging. Sample recovery/return was split into 0-25%, as a second of the sample recovery and the sample recovery return was split into 0-25%.
	25-65%, 65- 100%; whilst moisture was Wet, Damp, Dry. A record of 1 had the lowest recovery and was wet, 9 was considered high recovery and dry. Samples logged with a Quality ranking of wet, regardless of % return was set to absent and ignored during the grade
	estimation process but used in the guidance of mineralisation.
Logging	FML/Crescent RC samples were geologically logged to record weathering, regolith, rock type, colour, alteration, mineralisation,
	structure, texture and any other notable features that are present. All data is entered directly into validating digital software directly.
	• In addition to parameters logged over RC chips, all diamond core was also logged for structure. If an orientation line was available,
	structure orientation measurements were taken and recorded. Core holes were oriented where possible and marked into metre intervals with relation to hole depth. Any loss of core was noted and
	 Core holes were oriented where possible and marked into metre intervals with relation to hole depth. Any loss of core was noted and recorded in the drilling database. Recovery and RQD measurements were recorded.
	The logging information was transferred into the company's drilling database once the log was complete.
	Logging was qualitative, however the geologists often recorded quantitative mineral percentage ranges for the sulphide minerals
	present.
	 Diamond core was photographed one core tray at a time wet and dry using a standardised photography jig. The entire length of all holes was logged.
	Hillmin/Ashton logged the entire drill hole for colour, weathering, rock type, texture, structure, alteration, veining and mineralisation.
	Ashton diamond holes were also geologically logged for colour, weathering, rock type, texture, structure, alteration, veining and
	mineralisation.
	SOG logged holes from surface for weathering, lithology, texture, grain size, colour, alteration and veining. WMO DC complex years logged to record colour, grain size, accessingly yearth size, at wat well fabric and colour size.
	 WMC RC samples were logged to record colour, grain size, occasional weathering, structural fabric and rock type. Metex/LEJV RC and DD holes were logged for colour, weathering, structural fabric, alteration, veining, mineralisation, sample quality
	and lithology. Diamond core was also logged for recovery and RQD.
	Teck and Technomin RC holes were logged for colour, weathering, rock type, quartz veining.
Sub-sampling	FML All samples were collected in a pre-numbered calico bag bearing a unique sample ID. Jinning Testing & Inspection completed the
techniques and sample	assay testing, with sample preparation and assay completed in Kalgoorlie. All samples were oven dried, crushed to a nominal 10mm
preparation	using a jaw crusher (core samples only) and weighed. Samples in excess of 3kg in weight were riffle split to achieve a maximum 3kg sample weight before being pulverized to 90% passing 75µm. Gold analysis was by 40g Fire Assay with an AAS Finish.
	• Crescent submitted 1m RC samples or ½ core diamond samples. Samples were collected in pre-numbered bags weighing approx. 3kg
	and submitted to various laboratories for fire assay or screen fire assay with an ICP-OES or AAS Finish.
	Hillmin/Ashton submitted either 1m samples or 4m composite samples in numbered bags that corresponded to the 1m intervals they'd
	composited. Samples were sent to AAL Laboratories in Leonora, SGS in Kalgoorlie or Ultratrace in Perth for Fire Assay on a 50g charge with an AAS finish. Where the composite sample exceeded 0.1 ppm Au, the pre-numbered individual 1m samples were
	submitted for Fire Assay to a lower detection limit of 0.01ppm Au.
	Ashton reports state samples submitted to SGS Kalgoorlie, samples were dried, jaw crushed, hammer milled, split and pulverised.
	Samples were analysed for gold by fire assay on a 50g charge to a lower limit of detection of 0.01 ppm Au. Where the composite
	assay exceeded 0.1 ppm, the relevant 1m interval was submitted to SGS for analysis. • Hillmin/Ashton diamond core was sampled as either 4m filleted composites or a sawn core sampled to lithological contacts. Samples
	were submitted to Genalysis or SGS Kalgoorlie for gold analysis by screen fire assay method.
	WMC sub-sampling and assay preparation not documented. Samples were submitted to WMC labs at its Windarra or Kalgoorlie
	operations.
	SOG submitted 1m or 2-4m composite samples for analysis to ALS Laboratories for analysis by aqua regia digest with an AAS finish SOURCE Trace Porth for fire access.
	or Ultra Trace Perth for fire assay. • Teck submitted 2m composite samples to Analabs Kalgoorlie by aqua regia digest with an AAS, subsequent 1m samples submitted
	were analysed by fire assay.
	Technomin submitted 1m or 2m composite samples weighing approx. 2-3kg to Australian Assay Laboratories for a 50g fire assay.
	Metex/ LEJV RC samples were submitted to Amdel or Genalysis Kalgoorlie for analysis with either an aqua regia digest or 50g fire assay: At Cladister a multiplement analysis was run as camples.
	 assay. At Gladiator a multielement analysis was run on samples. Metex diamond samples were submitted to Genalysis for multielement analysis with Aqua regia analysis and fire assay on the re-split.
Quality of assay	FML inserted 2 standards and collected 4 duplicates for every 100 samples. Diamond core field duplicates were not taken. Laboratory
data and	replicates were also taken in the sample preparation stage by the responsible laboratory.
laboratory tests	All results from assay standards, duplicates and lab repeats were scrutinised to ensure they fell within acceptable tolerances.
	Crescent submitted Certified Standards, blanks, field duplicates and laboratory repeats at regular intervals over the drill programme. Crescent logged the sample quality as wet, moist or dry and reviewed sample weights to flag holes as being unreliable and evoluded.
	 Crescent logged the sample quality as wet, moist or dry and reviewed sample weights to flag holes as being unreliable and excluded from the estimation.
	Crescent also twinned 5 RC holes at Innuendo with diamond to ascertain the effects of the high water content encountered whilst drilling.
	The report concluded there is reliability issues with down hole contamination in wet samples. This has been taken into account with the
	estimation by removing all samples logged as wet or unreliable.
	 All results from assay standards, duplicates and lab repeats were scrutinised to ensure they fell within acceptable tolerances. Crescent resource geologists also reviewed the available QAQC data for pre- Crescent drilling and generated Q-Q plots to compare the
	data within flagged lodes and filtered by reliability. The data distribution between companies was comparable and considered acceptable
	to use.
	Hillmin/Ashton took field duplicate samples in the RC.
	Hillmin ran a laboratory comparison check during the 1987 drill program comparing RDL Assay results to SGS Assay results for selected drill belo integrals. The a comparing with Mindels
<u> </u>	drill hole intervals. Then comparing with Minlab.

Criteria	Co	mmentary
	•	At CM/MM Hillmin twinned a selection of RC holes with Diamond holes in 1988.
	•	Ashton also ran a laboratory comparison check during the 1989 drill program comparing SGS Assay results for selected drill hole intervals
		another laboratory, Minlab using a 50g fire assay. Results were found to be comparable.
	•	SOG used Field duplicates and laboratory replicates to check repeatability of results.
		WMC sample checks and laboratory information is not well documented however the drilling techniques and assay method are
		appropriate for this style of mineralisation. Previous Crescent and Metex resource estimates have reviewed and plotted QQ plots to
		confirm the tenor of mineralisation is comparable.
		Teck Minerals also ran a re-assaying program with comparable results.
		Technomin submitted duplicates as a check on repeatability.
	•	Metex submitted field duplicates at a rate of 1:50 for RC drilling and also used laboratory repeats and standards in their quality checks.
		In 1998 along the Chatterbox trend Metex drilled diamond holes to twin previously drilled Metex RC holes as a check. Results showed
		similar widths and grades of mineralisation were intersected by both drilling methods.
		No geophysical tools, spectrometers or handheld XRF instruments were used.
Verification	•	Historic sampling and assaying have been checked against hard copy WAMEX reports.
of	•	FML primary data is sent in digital format to the company's Database Administrator (DBA) as often as was practicable. The DBA
sampling		imports the data into an acQuire database, with assay results merged into the database upon receipt from the laboratory. Once
and		loaded, data was extracted for verification by the geologist in charge of the project.
assaying		
	٠	No adjustments were made to any current or historic data. If data could not be validated to a reasonable level of certainty it was not
l a antion of data		used in any resource estimations.
Location of data points	•	FML utilises Landgate sourced regional topographic maps and contours as well as internally produced survey pick-ups produced by
ροιπισ		the mining survey teams utilising DGPS base station instruments.
	•	FML drill collars were surveyed upon completion, using a DGPS instrument. Drill core was oriented by the drilling contractor using an
		Ezy-mark system. For RC a north-seeking gyroscope tool was used to survey down hole. For DDH a magnetic single shot survey was
		completed at 30m intervals on advance.
	•	The majority of Crescent Gold holes were surveyed by Electronic Multi-shot down hole or gyroscopic survey with collars surveyed by
		site survey personnel.
	•	Drill core was oriented by the drilling contractor using an Ezy-mark system.
	•	Hillmin WAMEX reports note the use of registered surveyors to record the drill hole collars in a local grid.
	•	Ashton collar survey methods are unknown and reported in local grid.
	•	SOG holes were surveyed using an Eastman Single Shot camera at the base of holes over 60m depth.
	•	WMC holes were collar surveyed by WMC survey staff in a local grid.
	•	Teck and Technomin do not state their survey methods. Down hole dips are the planned dip.
	•	Metex/LEJV holes were surveyed by a consultant survey company. Diamond core samples were surveyed by Single Shot Eastman
		camera. Later RC holes drilled in the JV were gyroscopic down-hole surveyed.
	•	All coordinates and bearings use the MGA94 Zone 51 grid system.
	•	Historic holes have been converted to MGA94 Zone 51 grid system in Acquire.
	•	Historic hole collars were sometimes still visible and re-surveyed to check the accuracy of the grid conversion. The comparison was
		considered within acceptable error limits of using a DGPS unit.
Data spacing and	•	Drill spacing along the Chatterbox Trend within the deposit areas is nominally on a grid spacing of 25m x 25m, although at Rumor the
distribution		grid spacing is closer to 50m x 25m. Apollo within and proximal to the existing open pits has been drilled down to 12.5m x 12.5m
		spacing in places. Between deposits spacing increases to 50m x 25m and 100m x 50m at the extremes.
	•	West Laverton drill spacing within and immediately surrounding the existing open pits is tight grid spacing 10m x 10m to 25m x 15m.
		Further out from the pits it extends to a more irregular spacing 25m x 30m-60m. Between West Laverton and Bulldog the drill spacing
		is irregular, a 350m gap in RC or DD drilling exists between the West Laverton and Bulldog deposits.
	•	Gladiator drill spacing within the existing open pits of Gladiator and Murrays is 15m x 25m, extending out along strike of the pits to 15m
		x 35m – 50m for a couple of drill lines before becoming a single drill line. Cousin Murray has a 25m x 25m spaced drill pattern.
		Gladiator West is more irregular spaced pattern of ~ 25m x 25m to 40m x 50m.
	•	CM/MM trend drill spacing is tightly spaced within pit area's along known mineralisation trends. Within the pit area's drill spacing is
		10m x 15m. Near pit drill spacing extends to an irregular 25m x 20m, which has been infilled down to 12.5m x 10m in certain target
		areas.
	•	Between Craigiemore and Mary Mac deposits the drill spacing is irregular. There is an 80m gap at the end of the known Craigiemore
		trend before a small 140m strike of two RC "fence lines" of drilling 25m x 50m spaced. It is then another 90m from the end of the small
		cluster of RC to the start of the Mary Mac trend and more regular spaced drilling. A gap of 200m exists between the end of Mary Mac
		Hill and the Golden Pinnacles drilling which focuses on two out-crops and has an irregular drill spacing.
Orientation of	•	Drilling was designed based on known geological models, field mapping, verified historical data and cross-sectional interpretation.
data in relation		Drill holes were oriented at right angles to strike of deposit, with dip optimised for drill capabilities and the dip of the ore body.
to geological		Sim following and an expect to carrie of deposit, with alp optimized for all supposition and an expectation
structure		
Sample security	•	All samples were reconciled against the sample submission with any omissions or variations reported to FML.
	•	All samples were bagged in a tied numbered calico bag, grouped into green plastic bags. The bags were placed into bulka bags with a
		sample submission sheet and kept within the Laverton yard until ready for transport to Kalgoorlie by transport courier.
	•	Historic sample security is not recorded.
Audits or reviews	•	Early Crescent Resource Models were completed by external consultants who undertook data validation as part of the scope of works.
		No external audit or review of the Resource Models has been undertaken.
		The state of the s

Section 2 Reporting of Exploration Results

(Criteria in this section apply to all succeeding sections.)

Criteria Mineral tenement and land tenure status

Commentary

- All exploration was conducted on tenements 100% owned by FML or its subsidiary companies Focus Operations Pty Ltd and Focus Minerals Laverton. All tenements are in good standing.
- Various royalties may be in place as documented in the FML Annual Report.
- Native title determination for Nyalpa Pirniku was announced on 31 October 2023. The Laverton Gold Project includes regions that are
 variously classified in this determination. The Central Laverton deposits and Mineralised Trends detailed in this report are within regions
 now classified as: Native title exists (non exclusive).
- Chatterbox deposits occur across tenements M 38/535 and M 38/101.
- West Laverton deposits are within tenement M 38/345.
- Adjacent Gladiator deposits occur across tenements M 38/363, M 38/364, M 38/342, E 38/3424.
- CM/MM deposits are within tenement M38/270

Exploration done by other parties

- Various stakeholders over the years have engaged in activities over the deposits including but not limited to geological mapping, ground
 magnetic surveys, soil sampling, aeromagnetic surveys, costean sampling and rock chip sampling.
- Drilling campaigns have been completed over the area by various parties detailed in Section 1.
- Along with RC and DD drilling, Air Core and RAB drilling methods have been used to delineate the deposits.
- Focus Minerals Laverton successful acquired Crescent Gold in October 2012.
- Along the Chatterbox, in 2004 the JV between Metex Resources and PDAP mined a trial pit at Apollo (formerly known as Whisper at time
 of mining) as part of a pre-feasibility study. A figure of 68Kt @ 2.44g/t Au for 5,351 ounces has been recorded. Crescent Gold commenced
 open cut mining along the Apollo trend as four discrete pits, separate from the original Whisper pit, from November 2011 to September
 2012. The pits varied from 20m deep to 75m total depth. A total of 1.05Mt @ 1.76g/t Au for 59,500 ounces was mined from the four pits.
 Eclipse (Garden Well) was also mined by Crescent during this time to a depth of 60m for a total of 103Kt @ 2.86g/t Au for 9,443 ounces.
- All three deposits along the West Laverton trend have been historically mined as discrete open pits by Ashton Gold in the early 1990's, with West Laverton the largest. West Laverton was excavated from December 1990 through to May 1992. A reported 116Kt @ 3.15g/t Au for 11,791 ounces was mined from the pit. The final pit reached a depth of 61m. Whilst mining West Laverton, Rega Pit was mined by Ashton from November 1991 until May 1992. A total of 120Kt @ 3.53g/t Au for 13,709 ounces was extracted. The final pit depth was 62.5m. Bulldog was also mined during this period producing 158Kt @ 2.15g/t Au for 10,940 ounces. The final pit reached a depth of 50m. In November 2010 Crescent Gold commence pre-strip waste mining in the 230m long region between West Laverton and Rega open pits. A 247-drill hole campaign of 10m x 10m grid shallow RC grade control was conducted in the region prior to waste mining commencing. In December 2010, trial of mining mineralised waste for two benches was conducted to test the proposed mining methods to account for the shallow dip of the mineralisation. A reported 9Kt @ 0.57g/t Au of mineralised waste was excavated and stockpiled. No further mining was conducted at West Laverton.
- Gladiator open pit was mined as a North and South pit by Ashton from Sept 1990 through until February 1992 to a depth of 72.5m. Milling data reports 409Kt @ 2.49g/t Au for 32,771 ounces was processed from both pits. Nearby Murrays open pit was also mined by Ashton from January 1991 to May 1992 for a final depth of 35m. Milling data for Murrays reported 144Kt @ 1.94g/t Au for 8,967 ounces processed. A historic underground mine also known as Gladiator (at the southern extent of Gladiator West) was actively mined between 1897 and 1942 when WW2 impacted mining. It was reportedly mined to about 200m vertically with a strike length of ~ 200m producing 139Kt @ 12g/t Au for 53,600 ounces.
- The Craigiemore deposit has been historically mined as underground drives and shafts in the early 1900's through to the late 1930's, with
 a recorded production of 135Kt @ 9.60g/t Au for 41,774 ounces. Minor open cut mining occurred by a private entity in the late 1970's to
 early 1980's with a recorded production of 4Kt @ 1.84g/t Au for 240 ounces. In 1988 Hillmin commenced mining at Craigiemore by open
 cut methods until 1993 producing 592Kt @ 2.13g/t Au for 38,000 ounces.
- Crescent Gold recommenced open cut mining at Craigiemore in June 2010 through until July 2011. An unreconciled mining production of 619Kt @ 1.67g/t Au for 33,178 ounces was recorded in the Crescent mining database.
- The Mary Mac deposit was historically mined in the early 1900's by underground drives and shafts mostly from 1909 until 1913, a figure of 42Kt @ 9.21g/t Au for 12,440 ounces has been reported. In August 2010, Crescent commenced open cut mining at Mary Mac South (MMS) until April 2011, reportedly mining 692Kt @ 1.26g/t Au for 28,034 ounces. Whilst still mining MMS, open pit excavation of Mary Mac Hill (MMH) to the North of MMS commenced in February 2011. Mining open pit continued until July 2012, reportedly 494Kt @ 1.84g/t Au for 29,230 ounces was open cut excavated from MMH.
- Numerous historical shafts exist on the Golden Pinnacles deposit, production figures are unknown.

Geology

- Regionally the deposits are part of the Laverton Greenstone Belt in the Eastern Yilgarn Craton. Lying within the Kurnalpi Terrane which is
 dominated by andesitic volcanics with erosional remnants of siliclastic sequences, the deposits are located on N to NE striking shear
 zones between the Mt Margaret Dome in northwest and the Kirgella Dome in the southeast.
- Locally the Chatterbox Trend of deposits is hosted by a large-scale structural feature of the region the Chatterbox Shear Zone, from
 which its name is derived. This moderately ESE dipping ductile/brittle fault zone separates the Laverton Lithostructural domains from the
 Mount Margaret Lithostructural domains. Rock units within the deposit areas are strongly altered and sheared sediments and
 metasediment rocks, felsic intrusives and ultramafics to the east, in the footwall. Mineralisation is commonly associated with increased
 goethite/manganese/hematite alteration or the intrusions.
- The West Laverton deposits consists of two north-south trending banded iron formation (BIF) ridges within a sequence of mafic and ultramafic volcanic and intrusive rocks with interflow sediments. The Laverton Shear Zone, a major north-south trending shear that delineates the western boundary of the Laverton Tectonic Zone, is interpreted as extending through the West Laverton trend from Rega in the North to Bulldog in the south. The footwall of the shear is dominated by a dolerite. The hanging wall comprises basalt. Gold mineralisation is associated with the shear zone overprinting an ultramafic gabbro. West Laverton mineralisation is generally associated within dilational jogs along the shear zone and shallowly dipping quartz veins. Rega mineralisation is interpreted as hosted within two shear zones within massive mafic and pillow basalt units. Bulldog mineralisation is losted within ductile shear zones with quartz veining.
- Gladiator Underground and Murrays deposits are closely related to a bending NNE, SW to NNW

Criteria Commentary

- striking, east dipping banded iron formation (BIF). Mineralisation is associated with quartz reefs parallel to the BIF and dips in multiple directions. The stratigraphy is dominated by a basalt unit on the west and gabbroic units of varying compositional and granulometry on the east with felsic porphyry units intruding sporadically. Gladiator West sits on the basaltic footwall of Gladiator Underground. Cousin Murray is within the gabbroic hanging wall of Murrays Open Pit and is striking NW, mineralization is associated with silica-sericite alteration. Gladiator Open pit mineralization is associated with quartz feldspar porphyry intrusives. Geology is dominated by basalt with a corridor of dolerite and a felsic tuff on the SW side of the pit and a NE striking BIF on the NE side of the pit. The Central BIF between Cousin Murray and Gladiator South and the NE BIF on the NE of Gladiator South hold mineralization.
- The CM/MM trend is hosted in a meta-sedimentary/ mafic volcanic package of rocks that has been highly deformed through late stage
 folding and faulting. A central steeply dipping Banded Iron Formation (BIF) unit has been associated with the gold mineralisation. Gold
 commonly occurring within the quartz veining and disseminated pyrite of the silicified chert horizons of the BIF. Evidence of a supergene
 enrichment zone near the vicinity of the
- water table was noted during mining by Crescent. The BIF horizon strikes north south and has been traced northwards to Laverton townsite, approximately 4km away.

Drill hole Information

Chatterbox:

 Historic drilling information has been validated against publicly available WAMEX reports. Not all drill holes can be found referenced in the WAMEX reports. However, cross-checking of original drill surveys was verified against the database:

WAMEA reports. However, cross-checking or original drill surveys was verified a	•	
Drill Hole Number	WAMEX Report A- Number	WAMEX Report Date
GDWC009, GDWC010, GDWC012, GDWC013, GDWC014, GDWC030, GDWC031, GDWC034, GDWC035, GDWC037, GDWC038, GDWC048, GDWC049, GDWC052, GDWC053, GDWC056, GDWC057, GDWC058	31396	Jun-89
GDWC086, GDWC091, GDWC093, GDWC094, GDWC100, GDWC101, GDWC102, GDWC103,	35126	Feb-92
GDWC105, GDWC106, GDWC107, GDWC109, GDWC110, GDWC1111, GDWC113, GDWC114, GDWC115, GDWC125, GDWC121, GDWC122, GDWC123, GDWC126, GDWC127,		
GDWC128, GDWC130, GDWC131, GDWC132, GDWC133, GDWC134, GDWC135, GDWC136, GDWC137, GDWC138, GDWC139, GDWC140, GDWC145, GDWC146, GDWC147		
GWD001_W, GWD002_W, GWRC004, GWRC005, GWRC007, GWRC008, GWRC009, GWRC010, GWRC015, GWRC018, GWRC019, GWRC020, GWRC021, GWRC025, GWRC026,	54899	Mar-98
GWRC027, GWRC028, GWRC029, GWRC031, GWRC032, GWRC033, GWRC034, GWRC035,		
GWRC036, GWRC037, GWRC038, GWRC039, GWRC040, GWRC041, GWRC042, GWRC043, I GWRC045, GWRC046, GWRC047, GWRC048, GWRC049, GWRC050, GWRC051, GWRC052, I		
GWRC055, GWRC056, GWRC057, GWRC058, GWRC059, GWRC060, GWRC061, GWRC062, GWRC063, GWRC064, GWRC065, GWRC066, GWRC067, GWRC068, GWRC069, GWRC070,		
GWRC072, GWRC074, GWRC075, GWRC076, GWRC077, GWRC078, GWRC079, GWRC080, GWRC081, GWRC082, GWRC084, GWRC085, GWRC087, GWRC088, GWRC089, GWRC090,		
GWRC092, GWRC093, GWRC094, GWRC095, GWRC097, GWRC099, GWRC100, GWRC101,		
GWRC102, GWRC103, GWRC104, GWRC105, GWRC106, GWRC107 GWD004_W, GWD005_W, GWD006_W, GWD007_W, GWD008_W, GWD009_W, GWD010_W,	57921	Mar-99
GWD011_W, GWD012_W, GWD013_W, GWD014_W, GWD015_W, GWD016_W, GWD017_W, GWD018_W, GWD019, GWRC108, GWRC109, GWRC110, GWRC111, GWRC113, GWRC114,		
GWRC117, GWRC120, GWRC121, GWRC122, GWRC124, GWRC125, GWRC126, GWRC127,		
GWRC128, GWRC129, GWRC132, GWRC133, GWRC137, GWRC138, GWRC140, GWRC141, GWRC142, GWRC143, GWRC145, GWRC146, GWRC147, GWRC148, GWRC149, GWRC150,		
GWRC151, GWRC152, GWRC153, GWRC154, GWRC155, GWRC157, GWRC158, GWRC159, GWRC160, GWRC161, GWRC163, GWRC164, GWRC165, GWRC166, GWRC167,		
GWRC168, GWRC170, GWRC171, GWRC173, GWRC174, GWRC175, GWRC176, GWRC177, GWRC178, GWRC179, GWRC180, GWRC185, GWRC186, GWRC187, GWRC189, GWRC190,		
GWRC191, GWRC192, GWRC193, GWRC194, GWRC195, GWRC196, GWRC197, GWRC198,		
GWRC199, GWRC200, GWRC201, GWRC202, GWRC204, GWRC205, GWRC207, GWRC210, GWRC211, GWRC212, GWRC213, GWRC217, GWRC218, GWRC219, GWRC220, GWRC221,		
GWRC223, GWRC225, GWRC226, GWRC227, GWRC229, GWRC230, GWRC231, GWRC232, GWRC233, GWRC234, GWRC236, GWRC237, GWRC239, GWRC240, GWRC241, GWRC242,		
GWRC243, GWRC244, GWRC245, GWRC246, GWRC247, GWRC248, GWRC249, GWRC250,		
GWRC251, GWRC252, GWRC253, GWRC254, GWRC255, GWRC256, GWRC257, GWRC258, GWRC259, GWRC260, GWRC261, GWRC263, GWRC264, GWRC266, GWRC267, GWRC268,		
GWRC269, GWRC270, GWRC271, GWRC272, GWRC273, GWRC277, GWRC281, GWRC282, GWRC283, GWRC285, GWRC286, GWRC289, GWRC290, GWRC291, GWRC292, GWRC293, J		
GWRC294, GWRC295, GWRC296, GWRC297, GWRC298, GWRC299, GWRC300, GWRC301, GWRC302, GWRC303, GWRC304, GWRC305, GWRC306, GWRC307, GWRC308, GWRC309,		
GWRC310, GWRC311, GWRC313, GWRC314, GWRC315, GWRC316, GWRC317, GWRC318,		
GWRC320, GWRC321, GWRC322, GWRC323, GWRC324, GWRC325, GWRC326, GWRC329, GWRC330, GWRC331, GWRC332, GWRC333, GWRC334, GWRC335, GWRC336, GWRC338, GWRC336, GWRC36, GWRC36		
GWRC340, GWRC341, RFRC002, RFRC005, RFRC006, RFRC008, RFRC009, RFRC010, RFRC011		
GWD023, GWD024	65027	Feb-02
GWD025, GWD027, GWD028, GWD030, GWD031, GWD032, GWRC410, GWRC411 GWD035	66477 68953	May-03 Mar-04
GWRC348, GWRC349, GWRC350, GWRC352, GWRC355, GWRC356, GWRC357, GWRC358, GWRC369, GWRC360, GWRC361, GWRC362, GWRC363, GWRC364, GWRC365, GWRC366,	65027	Feb-02
GWRC367, GWRC368, GWRC369, GWRC370, GWRC371, GWRC372, GWRC373, GWRC374, GWRC375, GWRC376, GWRC378, GWRC379, GWRC380, GWRC381, GWRC383, GWRC384, GWRC385, GWRC386, GWRC387, GWRC388, GWRC389, GWRC391, GWRC392, GWRC393,		
GWRC394, GWRC395, GWRC397, GWRC398, GWRC399, GWRC400, GWRC401, GWRC403, GWRC404, GWRC405, GWRC406		
GWRC420, GWRC421, GWRC422, GWRC423, GWRC424, GWRC425, GWRC426, GWRC427, GWRC428, GWRC429, GWRC430, GWRC431, GWRC432, GWRC433, GWRC434, GWRC435, GWRC436, GWRC437, GWRC438, GWRC439	68953	Jul-04

Criteria	Commentary		
Criteria	Commentary WHDD001, WHDD002, WHDD003, WHDD004, GWRC468, GWRC469, GWRC470, GWRC471, GWRC472, GWRC475, GWRC476, INRC005, INRC006, INRC007, INRC008, INRC009, INRC010, INRC011, INRC013, INRC015, INRC015, INRC016, INRC017, INRC018, INRC021, INRC022, INRC023, INRC024, INRC026, INRC026, INRC027, INRC028, INRC029, INRC030, INRC031, INRC033, INRC033, INRC034, INRC035, INRC035, INRC036, INRC037, INRC040, INRC041, INRC042, INRC043, INRC044, INRC046, INRC048, INRC036, INRC037, INRC040, INRC018, WHRC018, WHRC029, WHRC020, WHRC022, WHRC023, WHRC024, WHRC025, WHRC026, WHRC027, WHRC028, WHRC029, WHRC030, WHRC031, WHRC032, WHRC033, WHRC034, WHRC034, WHRC034, WHRC035, WHRC036, WHRC037, WHRC036, WHRC037, WHRC036, WHRC037, WHRC046, WHRC047, WHRC048, WHRC049, WHRC050, WHRC041, WHRC045, WHRC045, WHRC046, WHRC047, WHRC048, WHRC046, WHRC050, WHRC051, WHRC056, WHRC059, WHRC063, WHRC063, WHRC064, WHRC056, WHRC066, WHRC069, WHRC061, WHRC063, WHRC063, WHRC064, WHRC065, WHRC066, WHRC069, WHRC067, WHRC066, WHRC067, WHRC068, WHRC067, WHRC069, WHRC067, WHRC068, WHRC067, WHRC068, WHRC067, WHRC068, WHRC067, WHRC068, WHRC069, WHRC079, WHRC080, WHRC081, WHRC082, WHRC083, WHRC086, WHRC087, WHRC088, WHRC091, WHRC081, WHRC083, WHRC086, WHRC087, WHRC086, WHRC087, WHRC088, WHRC091, WHRC081, WHRC093, WHRC094, WHRC085, WHRC086, WHRC087, WHRC088, WHRC091, WHRC091, WHRC092, WHRC093, WHRC094, WHRC095, WHRC096, WHRC097, WHRC081, WHRC097,	93988	Apr-11
	APRC152, APRC153, APRC154, ECRC001, ECRC002, ECRC003, ECRC004, ECRC005, ECRC006, ECRC007, ECRC009, ECRC013, ECRC014, ECRC015, ECRC016, ECRC017, ECRC018, ECRC019, ECRC020, ECRC021, ECRC022		
	ECRC016, ECRC019, ECRC020, ECRC021, ECRC022 ECRC024, EMRC001	98404	Jun-13

Chatterbox collar det	ails of holes not pre	eviously externally r	eported:				
COMPANY	BHID	EAST	NORTH	RL	AZIMUTH	DIP	DEPTH
FOCUS	GWRC478	433639.44	6830109.6	441.61	262.6	-60	152
METEX	GWRC001	434067.78	6831338.3	440.77	270	-60	20
	GWRC002	434137.79	6831338.3	442.8	270	-60	90
	GWRC003	434157.79	6831338.3	442.8	270	-60	96
	GWRC006	434087.79	6831378.3	441.82	270	-60	30
METEX / PDAP	GWRC450	433637.55	6830432.5	438.18	270	-60	60
	GWRC451	433668.63	6830482.5	437.8	270	-60	60
	GWRC452	433638.63	6830483.3	437.59	90	-60	60
	GWRC453	433697.76	6830833.2	434.8	270	-60	65
	GWRC454	433709.15	6830833.3	434.88	270	-60	69
	GWRC455	433721.06	6830832.8	434.94	270	-60	80
	GWRC456	433697.49	6830858.3	434.12	270	-60	65
	GWRC457	433709.94	6830858	434.39	270	-60	70
ĺ	GWRC458	433721.67	6830858.1	434.4	270	-60	80
	GWRC459	433662.87	6830858	434.05	90	-60	65
ĺ	GWRC460	433697.56	6830883.5	434.19	270	-60	65
	GWRC461	433709.58	6830882.9	434	270	-60	75
	GWRC462	433693.32	6830908.2	434.33	270	-60	60
ĺ	GWRC463	433702.43	6830908.3	434.35	270	-60	70
	GWRC464	433647.76	6830908.2	434.13	270	-60	50

Chatterbox RC Grade control holes drilled by Crescent / FML remaining beneath the current pit floors are tabulated below:

BHID	EAST	NORTH	RL	AZIMUTH	DIP	DEPTH
AP410101	433722.71	6831508.7	435.34	270	-60	18
AP410102	433712.81	6831507.5	435.27	270	-60	12
AP410105	433743.4	6831522.6	435.68	270	-60	36
AP410106	433735.37	6831522.6	435.58	270	-60	24
AP410107	433727.38	6831522.7	435.51	270	-60	24
AP410108	433718.52	6831522.4	435.43	270	-60	21
AP410109	433709.16	6831522.6	435.28	270	-60	15
AP410110	433758.02	6831532.8	435.35	270	-60	30
AP410111	433749.71	6831533	435.71	270	-60	30
AP410113	433733.59	6831532.9	435.55	270	-60	30
AP410114	433717.84	6831532.9	435.34	270	-60	24
AP410115	433761.75	6831545.4	435.85	270	-60	27
AP410116	433754.12	6831545.4	435.78	270	-60	27
AP410118	433737.73	6831545.5	435.54	270	-60	30

Criteria	Commentary						
	AP410119	433729.6	6831545.6	435.46	270	-60	30
	AP410120	433721.26	6831545.7	435.44	270	-60	24
	AP410121	433745.44	6831559.1	435.65	270	-60	36
	AP410122	433727.25	6831556.8	435.47	270	-60	30
	AP410123	433701.15	6831557.4	435.14	270	-60	18
	AP410124	433693.1	6831557.4	435.11	270	-60	18
	AP410128 AP410129	433735.04 433726.72	6831569.5 6831569.5	435.65 435.48	270 270	-60 -60	30 30
	AP410132	433748.39	6831583.5	435.78	270	-60	24
	AP410134	433720.18	6831583.4	435.33	270	-60	24
	AP410136	433691.92	6831509.6	435.04	270	-90	10
	AP410138	433696.29	6831522.4	435.15	0	-90	10
	AP410140	433680.43	6831522.5	434.98	0	-90	10
	AP410149	433688.77	6831545.8	435.05	0	-90	10
	AP410163	433746.96	6831596.2	435.82	0	-90	12
	AP410164	433739.47	6831595.5	435.7	0	-90	12
	AP410165 AP410166	433731.62 433723.63	6831595.3 6831595.5	435.56 435.56	0	-90 -90	15 15
	AP410166 AP410169	433699.82	6831595.5	435.14	0	-90 -90	12
	AP410109 AP410172	433744.17	6831608.4	435.68	0	-90 -90	10
	AP410173	433729.85	6831608.1	435.56	0	-90	10
	AP410174	433720.32	6831608.1	435.43	0	-90	10
	AP410175	433704.95	6831608.3	435.21	0	-90	10
	AP410177	433731.21	6831620.8	435.61	0	-90	10
	AP410180	433706.91	6831620.3	435.27	0	-90	10
	AP410188	433699.02	6831645.6	435.14	0	-90	10
	AP410191	433731.51	6831527.7	435.48	270	-60	18
	AP410192	433739.35	6831527.7	435.56	270	-60	21
	AP410193	433746.95	6831527.7	435.34	270	-60	25
	AP410195	433762.97	6831527.4	435.62	270	-60	25
	AP410196	433770.89	6831527.6	435.93	270	-60	25
	AP410202	433754.61	6831539	435.7	270	-60	25
	AP410203	433762.33	6831538.8	435.52	270	-60	25
	AP410204	433770.56	6831538.8	435.96	270	-60	25
	AP410208	433739.35	6831551.3	435.47	270	-60	25
	AP410209	433747.52	6831551.1	435.71	270	-60	25
	AP410210	433755.26	6831551.1	435.6	270	-60	25
	AP410214	433749.82	6831557.1	435.73	270	-60	25
	AP410216	433723.68	6831562.9	435.36	270	-60	21
	AP410217	433731.58	6831562.8	435.36	270	-60	25
	AP410218	433739.69	6831563	435.54	270	-60	25
	AP410226	433702.89	6831627	435.14	0	-90	6
	AP410229	433727.77	6831626.9	435.52	0	-90	6
	AP410232	433717.14	6831614.2	435.36	0	-90	6
	AP410234	433732.44	6831614.3	435.59	0	-90	6
	AP410236 AP410237	433701.39 433717.02	6831602 6831602.1	435.18 435.4	0	-90 -90	6
	AP410237 AP410238	433717.02	6831602.1	435.4 435.44	0	-90 -90	10
	AP410238 AP410240	433725.34	6831602.5	435.44 435.67	0	-90 -90	10
	AP410240 AP410244	433679.38	6831589.5	434.72	0	-90 -90	7
	AP410250	433727.8	6831589.5	434.72	0	-90 -90	10
	AP410250 AP410251	433735.54	6831589.4	435.15	0	-90 -90	10
	AP410251	433743.01	6831589.4	435.38	0	-90	10
	AP410261	433676.81	6831516	434.75	0	-90 -90	6
	AP410263	433691.52	6831515.8	434.82	0	-90	6
	AP410437	433635.65	6830443.1	437.86	289	-50	41
	AP410443	433652.62	6830463.4	437.77	290	-50	45
	AP410457	433652.94	6830504.7	437.1	279.6	-60	35
	AP410462	433675.25	6830512.9	437.19	280	-50	48
	AP410464	433670.05	6830527.5	437.03	280	-50	36
	AP410465	433680.36	6830524.5	437.19	279.6	-50	45
	AP410467	433670.33	6830540.2	436.91	280	-50	29
	AP410468	433677.38	6830540	437.09	279.6	-50	36
	AP410469	433683.14	6830539.6	437.03	279.6	-50	44
	AP410471	433685.75	6830550.6	436.93	280	-50	48
	AP410476	433692.86	6830561.8	436.81	279.6	-50	46
	AP410481	433697.57	6830572.9	436.76	280	-50	48
	AP410485	433695.69	6830585.8	436.67	280	-50	45
	AP410503	433225.77	6829388.3	439.65	290	-50	48
	AP410504	433233.79	6829385.4	439.78	290	-50	54

Criteria	Commentary						
	AP410505	433226.18	6829401.3	439.64	290	-50	36
	AP410506	433233.92	6829398.3	439.67	290	-50	46
	AP410507	433248.01	6829393.3	439.74	289.6	-50	60
	AP410510	433241.1	6829409.3	439.54	290	-50	48
	AP410511	433248.92	6829406.5	439.59	290	-50	54
	AP410515	433256.38	6829416.6	439.38	290	-50	54
	AP410516	433264.53	6829413.6	439.54	290	-50	60
	AP410520	433256.77	6829430.6	439.38	289.6	-50	46
	AP410523	433258.43	6829442.8	439.15	289.6	-50	40.19
	AP410524	433265.68	6829440.4	439.33	289.6	-50	43
	AP410526	433258.78	6829456.2	439.01	289.6	-50	36
	AP410527	433265.88	6829453.8	439.11	289.6	-50	39
	AP410528	433269.61	6829452.1	439.19	289.6	-60.1	42
	AP410531	433274.19	6829464.1	439.01	289.6	-50	42
	AP410533	433271.96	6829478	438.9	289.6	-60	35
	AP410536	433282.85	6829487.3	438.84	289.6	-50	36
	AP410537	433289.28	6829484.9	438.93	289.6	-50	44
	AP410538	433281.03	6829500	438.99	289.34	-60.29	36.33
	AP420105	433655.87	6830520.8	419.77	270	-60	24
	AP420106	433664.68	6830520.6	419.51	270	-60	36
	AP420107	433672.83	6830520.6	419.81	270	-60	36
	AP420108	433680.46	6830520.2	419.93	270	-60	36
	AP420110	433657.7	6830508.1	419.57	270	-60	36
	AP420111	433669.66	6830508.1	419.81	270	-60	48
	AP420113	433696.79	6830505.9	422.24	270	-60	54
	AP420115	433680.9	6830495.7	422.38	270	-60	42
	AP420116	433685.85	6830495.3	422.3	270	-60	42
	AP420117	433692.98	6830495.3	422.28	270	-60	48
	AP420118	433701.53	6830495.2	422.42	270	-60	48
	AP420119	433646.22	6830482.9	422.37	270	-60	36
	AP420120 AP420121	433679.61 433688.03	6830482.7 6830482.5	422.38 422.35	270 270	-60 -60	42 48
	AP420121 AP420122	433646.91	6830470.9	422.49	270	-60	36
	<u> </u>			_	270		
	AP420124	433680.03	6830470.6	422.47		-60	48
	AP420125	433687.7	6830470.4	422.45	270	-60	48
	AP420126 AP420130	433651.67 433665.79	6830458.1 6830445.8	422.39 422.7	270 270	-60 -60	48 48
	AP420130 AP420131	433672.9	6830445.6	422.7	270	-60	48
	APC420002	433713.91	6830989.1	419.5	280	-55	21
	APC420002 APC420003	433723.22	6830987.5	419.72	280	-55	28
	APC420004	433758.45	6830979.8	419.57	280	-60	19
	APC420007	433722.82	6830974.2	419.9	280	-50	29
	APC420008	433730.42	6830973.2	420.05	280	-50	38
	APC420015	433684.96	6830968.6	419.71	280	-70	25
	APC420017	433699.91	6830966.4	419.77	280	-70	27
	APC420018	433707.84	6830965	419.79	280	-65	29
	APC420019	433719.69	6830963	420.01	280	-60	30
	APC420020	433738.04	6830959.4	420	277.2	-59.7	46
	APC420022	433766.5	6830953.9	420.42	280	-65	30
	APC420023	433652.99	6830962.5	419.52	0	-90	25
	APC420026	433718.69	6830950.1	419.96	280	-50	39
	APC420027	433726.45	6830948.7	420	272.9	-49	41
	APC420028	433735.17	6830947.3	419.91	272	-49	44
	APC420029	433743.12	6830945.8	419.9	268.8	-49.4	47
	APC420030	433750.13	6830944.5	419.9	270.5	-50	52
	APC420034	433644.2	6830951.7	419.77	0	-90	31
	APC420035	433651.72	6830949.7	419.55	0	-90	34
	APC420036	433659.98	6830947.8	419.71	0	-90	34
	APC420037	433667.23	6830946.5	419.77	0	-90	34
	APC420038	433676.9	6830944.4	419.75	280	-80	34
	APC420039	433686.93	6830942.9	419.84	280	-70	35
	APC420040	433698.12	6830940.6	419.7	280	-60	38
	APC420041	433706.89	6830939.1	419.96	280	-60	39
	APC420042	433720.43	6830936.5	420.02	270	-60	44
	APC420043	433733.22	6830934.7	419.96	270	-60	48
		433659.4	6830935.5	419.61	270	-60	38
	APC420046				_		
	APC420046 APC420047 APC420048	433670.49 433682.73	6830932.9 6830930.3	420.04 420.21	270 270	-60 -60	41 43

Criteria	Commentary						
	APC420051	433714.38	6830925.5	420.09	270	-50	49
	APC420052	433721.11	6830924.1	420.09	270	-50	53
	APC420053	433730.41	6830922.4	420.26	270	-50	56
	APC420054	433736.89	6830921.4	420.27	270	-50	53
	APC420056	433754.7	6830918.5	421.08	270	-50	58
	APC420057	433762.68	6830916.9	420.93	270	-50	65
	APC420059	433657.92	6830922.9	419.85	270	-60	37
	APC420060	433665.2	6830921.9	419.97	270	-60	39
	APC420061	433672.53	6830920.7	419.66	270	-60	40
	APC420062	433680.32	6830919.1	419.8	270	-60	42
	APC420063	433695.61	6830915.5	420.57	270	-60	43
	APC420065 APC420066	433711.92 433737.42	6830913.2 6830908.1	419.96 419.89	270 270	-60 -60	43 51
	APC420060 APC420071	433718.29	6830904.1	419.84	250	-60 -55	48
	APC420071	433732.66	6830902.9	419.64	250	-50	57
	APC420072	433745.12	6830894.7	420.11	270	-50	60
	APC420074	433756.69	6830892.4	420.03	270	-50	61
	APC420076	433645.54	6830900.3	419.87	270	-65	31
	APC420077	433655.88	6830899.2	419.78	270	-65	34
	APC420078	433664.92	6830897	419.77	270	-65	38
	APC420079	433675.87	6830895.1	419.77	270	-64	41
	APC420080	433695.74	6830890.7	419.78	270	-60	50
	APC420081	433705.15	6830889.5	419.88	270	-60	54
	APC420082	433723.98	6830885.5	419.98	270	-60	56
	APC420083	433737.48	6830883.5	420.08	270	-60	52
	APC420084	433745.61 433673.03	6830881.9 6830882.3	420 419.28	270 270	-60	55 38
	APC420087 APC420088	433673.03	6830882.3	419.28	270	-60 -60	38 42
	APC420080 APC420090	433730.83		420.18	270		57
	APC420090 APC420091	433740.88	6830871.9 6830870.1	419.98	270	-60 -60	59
	APC420091 APC420094	433767.85	6830865.1	419.84	270	-60	41
	APC420098	433703.73	6830863.7	419.79	280	-60	51
	APC420100	433721.91	6830860.6	420.13	270	-60	58
	APC420101	433735.59	6830857.8	420.36	270	-60	65
	APC420104	433735.23	6830845.7	420.25	270	-60	68
	APC420107	433761.81	6830840.9	420.2	280	-60	40
	APC420111	433695.02	6830838.8	420	280	-50	52
	APC420112	433702.88	6830837.5	420.02	280	-50	59
	APC420113	433710.41	6830836.3	420.08	280	-50	66
	APC420114	433718.57	6830834.6	420.18	279	-50	72
	APC420115	433724.04	6830834.1	420.19	280	-60	67
	APC420120	433761.28	6830762.1	420.13	280	-55 -50	56
	APC420122 APC420123	433702 433698.83	6830695.9 6830687.9	419.94 419.74	280 280	-50 -50	17 23
	APC425001	433714.29	6830684.5	424.64	280	-50	38
	APC425003	433723.73	6830695.4	424.92	280	-50	40
	APC425004	433733.63	6830693.1	425.11	280	-50	36
	APC425007	433736.03	6830705.2	424.91	280	-60	44
	APC425012	433738.62	6830717.6	424.84	280	-50	44
	APC425013	433746.09	6830716.5	424.8	280	-50	51
	APC425014	433754.51	6830714.8	424.92	280	-50	58
	APC425018	433736.12	6830731.2	424.95	280	-60	43
	APC425019	433747.12	6830729.2	424.97	280	-60	50
	APC425027	433759.23	6830739.5	425.18	280	-50	57
	APC425042	433691.41	6830790.5	424.88	280	-50 60	38
	APC425043A APC425051	433699.26 433708.62	6830788.4 6830799.6	424.97 424.85	280 280	-60 -50	42 53
	APC425051 APC425062	433708.62	6830813.2	424.86 424.86	280	-50 -50	62
	APC425063	433701.00	6830811.8	424.85	280	-50 -50	63
	APC425064	433726.01	6830809.3	425.05	280	-50	74
	APC425071	433708.14	6830824.8	425.07	280	-60	62
	APC425072	433711.17	6830823.9	424.77	280	-70	65
	APC425079	433771.99	6830813.9	424.77	280	-50	46
	APD410103	433521.59	6830120.5	409.88	270	-60	42
	APD410104	433529.04	6830120.5	409.99	270	-60	42
	APD410106	433519.9	6830107.9	409.87	270	-60	42
	APD410107	433527.04	6830107.9	409.89	270	-60	42
i .	APD410111	433521.6	6830095.6	409.63	270	-60	48

Criteria	Commentary						
	APD410112	433528.93	6830095.4	409.7	270	-60	54
	APD410114	433524.69	6830082.9	409.78	270	-60	60
	APD410115	433532.41	6830082.8	409.87	270	-60	66
	APD410118	433526.02	6830070.3	409.9	270	-60	66
	APD410119	433533.17	6830070.4	410.1	270	-60	73
	APD410122	433525.55	6830058	410.19	270	-60	66
	APD410123	433540.58	6830058.1	410.23	270	-60	66
	APD410127	433513	6830045.5	410	270	-60	60
	APD410128	433529.91	6830045.4	409.96	270	-60	72
	APD410133	433522.62	6830032.9	410.09	270	-60	66
	APD410139	433512.63	6830020.5	410.05	270	-60	60
	APD410140	433519.79	6830020.4	410	270	-60	60
	APD410141	433526.81	6830020.5	410.12	270	-60	60
	APD410144	433501.74	6830008.1	409.56	270	-60	60
	APD410147	433486.61	6829995.5	409.81	270	-60	48
	APD410148	433501.15	6829995.5	409.9	270	-60	60
	APD410149	433509.59	6829995.4	409.98	270	-60	60
	APD410151	433499.14	6829983.2	409.9	270	-60	60
	APD410152	433507.69	6829983.3	410.02	270	-60	60
	APD410154	433475.86	6829970.5	409.98	270	-60	30
	APD410155	433483.81	6829970.5	410.05	270	-60	36
	APD410156	433491.67	6829970.5	410.14	270	-60	36
	APD410157	433499.62	6829970.5	410.01	270	-60	45
	APD415101	433669.13	6830530.4	415.17	270	-60	30
	APD415102	433678.86	6830508.2	414.76	270	-59.1	50
	APD415103	433654.18	6830495.3	414.83	270	-60	30
	APD415106	433631.7	6830483	414.9	270	-60	18
	APD415107	433697.18	6830482.9	414.82	270	-60.9	48
	APD415108	433691.38	6830470.2	414.92	270	-59	42
	APD415109	433678.14	6830458	414.73	270	-58.7	42
	APD415111	433675.76	6830445.7	414.72	270	-58.7	42
	APD415113	433654.94	6830433.2	414.94	270	-60	36
	APD415116	433647.99	6830420.5	414.82	270	-60	36
	APD415117	433656.14	6830420.5	414.9	270	-57.3	42
	APD415118	433664.07	6830420.4	414.91	270	-59.1	54
	APD415120	433607.05	6830408	414.82	270	-60	24
	APD415121	433622.95	6830408	414.54	270	-60	36
	APD415123	433651.61	6830407.9 6830407.8	414.66	270 270	-60	60
	APD415124 APD415125	433666.93 433681.44	6830407.9	415.07 414.92	270	-60 -60	60 60
	APD415125 APD415127	433604.5	6830395.3	414.69	270	-60	30
	APD415127 APD415128	433612.81	6830395.4	414.63	270	-60	30
	APD415135	433663.79	6830395.6	414.76	270	-60	60
	APD415137	433605.93	6830383	414.79	270	-60	36
	APD415138	433612.56	6830382.9	414.75	270	-60	36
	APD415141	433652.84	6830383	414.75	270	-60	60
	APD415141	433667.12	6830382.8	414.92	270	-60	60
	APD415143	433678.7	6830382.9	414.96	270	-59.1	42
	APD415146	433594.01	6830370.4	414.81	270	-60	30
	APD415147	433601.74	6830370.4	414.68	270	-60	30
	APD415153	433649.72	6830370.4	415.07	270	-59.1	54
	APD415156	433673.86	6830370.5	414.97	270	-60	30
	APD415157	433681.61	6830370.4	414.98	270	-60	30
	APD415163	433641.71	6830358	414.56	270	-60	60
	APD415168	433587.26	6830345.5	414.7	270	-60	30
	APD415173	433628.62	6830345.7	414.71	270	-60	54
	APD415174	433636.39	6830345.8	414.8	270	-60	54
	APD415180	433583.66	6830333.1	414.71	270	-60	36
	APD415181	433591.42	6830333.5	414.81	270	-60	42
	APD415182	433598.24	6830333.3	414.86	270	-60	42
	APD415183	433615.49	6830333.2	414.99	270	-60	54
	APD415184	433632.23	6830333.2	414.9	270	-60	60
	APD415189	433575.42	6830320.5	414.88	270	-60	36
	APD415190	433584.09	6830320.5	414.83	270	-60	42
	APD415191	433592.09	6830320.5	414.81	270	-60	54
	APD415192	433599.8	6830320.5	414.86	270	-60	54
	APD415194	433615.95	6830320.6	414.68	270	-60	54
	APD415195	433624.03	6830320.4	414.81	270	-60	54
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### APP415559 \$450614 \$650261 \$44.88 \$70 \$40 \$4 ### APP415506 \$4507204 \$650266 \$414.54 \$70 \$40 \$4 ### APP415068 \$4507204 \$650266 \$414.54 \$70 \$40 \$4 ### APP41507 \$450623 \$650266 \$444.72 \$70 \$40 \$4 ### APP41507 \$450623 \$650266 \$444.72 \$70 \$40 \$6 ### APP41507 \$450623 \$650266 \$444.72 \$70 \$40 \$6 ### APP41507 \$450623 \$650266 \$444.72 \$70 \$40 \$6 ### APP41507 \$450626 \$650266 \$444.72 \$70 \$40 \$6 ### APP41507 \$450626 \$650266 \$444.72 \$70 \$40 \$4 ### APP41507 \$45067 \$4 ### APP41507 \$4 ### APP41	Criteria	Commentary						
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## APPLISTOR 43597 82 883278-6 414.67 270 461 54 54 54 54 54 54 54 5		APD415200	433600.15	6830308	414.88	270	-60	48
APPH15210 433958 22 8832056 44172 270 450 60 APPH15212 433803 8832056 3 4118 3 270 450 46 APPH15212 433803 8832056 3 4118 3 270 450 46 APPH15216 433857 8832053 1418 3 270 450 46 APPH15216 433957 48 8332053 1418 270 450 46 APPH15216 433957 48 8332053 1418 270 450 46 APPH15216 433957 48 8332053 1418 270 450 46 APPH1522 433950 8532053 1418 270 450 46 APPH1522 433950 8532053 1418 270 450 46 APPH1522 433956 8532053 1418 270 450 46 APPH1523 433958 48 8332053 1418 270 450 46 APPH1523 433958 79 6832705 4418 270 450 46 APPH1523 433958 79 6832705 4418 270 450 46 APPH1523 433958 79 6832705 4418 570 450 46 APPH1523 433958 79 6832705 4418 570 450 46 APPH1523 433958 79 6832705 4418 570 450 46 APPH1523 43395 79 6832705 4418 570 70 450 46 APPH1523 43395 79 6832705 4418 570 70 450 46 APPH1523 43395 79 6832275 4418 570 70 450 46 APPH1523 43395 79 6832275 4418 570 70 450 46 APPH1523 43395 79 6832275 4418 570 70 450 46 APPH1523 43395 79 6832275 4418 70 70 450 46 APPH1524 43395 79 6832275 4418 70 70 450 46 APPH1524 43395 79 6832275 4418 70 70 450 46 APPH1524 43395 79 6832275 4418 70 70 450 46 APPH1524 43395 79 6832275 4418 70 70 450 46 APPH1524 43395 79 6832275 4418 70 70 450 46 APPH1524 43395 79 6832275 4418 70 70 450 46 APPH1524 43395 79 6832275 4418 70 70 450 46 APPH1524 43395 79 6832275 4418 70 70 450 46 APPH1524 43395 79 6832285 4418 70 70 450 47 APPH1524 43395 79 6832285 4418 70 70 450 47 APPH1524 43395 79 6832285 4418 70 70 450 47 APPH1524 43395 79 6832285 4418 70 70 450 47 APPH1524 43395 79 6832285 4418 70 70 450 47 APPH1524 43395 79 6832285 4418 70 70 450 50 50 50 50 50 50 50 50 50 50 50 50 5		APD415208	433572.04	6830295.6	414.54	270	-60	42
APPH 1521 4 43395 69 881205 3 4 414 69 270 4-50 49 APPH 1521 4 43395 19 881205 3 415 270 4-50 49 APPH 15217 4 43395 19 881205 3 415 270 4-50 54 APPH 15219 4 43395 14 6 812023 1 414 62 270 4-50 54 APPH 15219 4 43395 14 6 812023 1 414 62 270 4-50 54 APPH 15219 4 43395 14 6 812023 1 414 62 270 4-50 49 APPH 15220 4 43395 89 881207 5 414 41 42 270 4-50 42 APPH 15220 4 43395 89 881207 5 414 41 42 270 4-50 42 APPH 15220 4 43395 89 881207 5 414 42 270 4-50 42 APPH 15220 4 43395 89 881207 5 414 41 5 270 4-50 42 APPH 15220 4 43395 89 881207 5 414 41 5 270 4-50 48 APPH 15220 4 43395 89 681207 5 414 41 5 270 4-50 48 APPH 15220 4 43395 81 681207 5 414 41 5 270 4-50 48 APPH 15220 4 43395 81 681207 5 414 5 270 4-50 48 APPH 15220 4 43395 81 4 881207 5 414 5 270 4-50 58 APPH 15220 4 43395 81 4 44 45 270 4-50 58 APPH 15220 4 43395 81 4 44 45 270 4-50 58 APPH 15220 4 43395 81 4 44 45 270 4-50 58 APPH 15220 4 43395 81 4 44 45 270 4-50 58 APPH 15220 4 43395 81 4 44 45 270 4-50 58 APPH 15220 4 43395 81 4 44 45 270 4-50 58 APPH 15237 4 4359 81 50 681205 81 44 47 7 270 4-50 58 APPH 15237 4 4359 81 50 681205 81 44 47 7 270 4-50 58 APPH 15237 4 4359 81 50 681205 81 44 47 7 270 4-50 58 APPH 15237 4 4359 81 50 681205 81 44 47 7 270 4-50 42 APPH 15237 4 4359 81 50 681205 81 44 47 7 270 4-50 42 APPH 15237 4 4359 81 50 681205 81 44 47 7 270 4-50 42 APPH 15237 4 4359 81 50 681205 81 44 47 7 270 4-50 42 APPH 15237 4 4359 81 50 681205 81 44 47 7 270 4-50 42 APPH 15237 4 4359 81 50 681205 81 44 47 7 270 4-50 42 APPH 15237 4 4359 81 50 681205 81 44 47 7 270 4-50 42 APPH 15237 4 4359 81 50 681205 81 44 47 7 270 4-50 42 APPH 15237 4 4359 81 50 681205 81 44 47 7 270 4-50 42 APPH 15237 4 4359 81 50 681205 81 44 47 7 270 4-50 42 APPH 15237 4 4359 81 50 681205 81 44 47 7 270 4-50 42 APPH 15237 4 4359 81 50 681205 81 44 47 7 270 4-50 42 APPH 15237 4 4359 81 50 681205 81 44 47 7 270 4-50 42 APPH 15237 4 4359 81 50 681205 81 44 44 47 7 270 4-50 42 APPH 15237 4 4359 81 50 681205 81 44 44 5 270 4-50 42 APPH 15239 4 4359 81 50 681205 81 4 44 47 7 270 4-50 42		APD415209	433579.82	6830295.4	414.67	270	-60	54
## APPH 15217 4335075 8830283 4416 270 660 42 ## APPH 15218 43350548 68502831 4144 49 270 660 42 ## APPH 15218 43350548 68502831 4144 49 270 660 49 ## APPH 15218 43350548 68502831 4144 49 270 660 49 ## APPH 15229 4335016 8830283 415 270 660 49 ## APPH 15229 4335016 8830283 415 270 660 49 ## APPH 15228 435085 19 8830270 4444 4144 3 270 660 49 ## APPH 15228 435085 19 8850270 5 4444 3 270 660 49 ## APPH 15231 435085 19 8850270 5 4444 5 270 660 49 ## APPH 15233 435085 19 8850270 6 4445 12 270 660 49 ## APPH 15233 435085 19 8850270 6 4445 12 270 660 49 ## APPH 15234 435085 19 8850270 6 4445 12 270 660 49 ## APPH 15235 435085 19 8850270 6 4445 12 270 660 49 ## APPH 15235 435085 10 8850270 6 4445 12 270 660 49 ## APPH 15236 435085 10 8850270 6 4445 12 270 660 49 ## APPH 15236 435085 10 8850270 6 4445 12 270 660 49 ## APPH 15236 435085 10 8850270 6 4445 12 270 660 49 ## APPH 15236 435085 10 8850270 6 4445 12 270 660 49 ## APPH 15236 435085 10 8850270 6 4445 12 270 660 49 ## APPH 15236 435085 10 885020 6 4445 12 270 660 49 ## APPH 15236 435085 10 885020 6 4445 12 270 660 49 ## APPH 15237 435085 10 885020 6 4445 12 270 660 49 ## APPH 15237 435085 10 885020 6 4445 12 270 660 49 ## APPH 15237 435085 10 885020 6 4445 12 270 660 49 ## APPH 15237 435085 10 885020 6 4445 12 270 660 49 ## APPH 15237 435085 10 885020 6 4445 12 270 660 49 ## APPH 15237 435085 10 885020 6 4445 12 270 660 49 ## APPH 15237 435085 10 885020 6 4445 12 270 660 49 ## APPH 15237 435085 10 885020 6 4445 14 49 270 660 49 ## APPH 15237 435085 10 885020 6 4445 14 49 270 660 49 ## APPH 15237 435085 10 885020 6 4445 14 49 270 660 49 ## APPH 15237 435085 10 885020 6 4445 14 49 270 660 49 ## APPH 15237 435085 10 885020 6 4445 14 49 270 660 49 ## APPH 15238 435085 10 885020 6 885020 6 4445 14 49 270 660 49 ## APPH 15236 435085 10 885020 6 885020 6 4445 14 49 270 660 49 ## APPH 15236 435085 10 885020 6 885020 6 4445 14 49 270 660 49 ## APPH 15236 435085 10 885020 6 885020 6 4445 14 49 270 660 49 ## APPH 15236 435085 10 885020 6		APD415210	433588.23	6830295.5	414.72	270	-60	60
APPHIG217 4335576 8636283 415 270 46 42 APPHIG219 4339514 83362831 41449 270 46 554 APPHIG219 4339514 83362831 4145 270 46 48 APPHIG229 43395174 83362831 4145 270 46 48 APPHIG229 4339519 83627015 41442 270 46 48 APPHIG229 4339519 83627015 41442 270 46 48 APPHIG229 4339517 83627015 41442 270 46 48 APPHIG229 4339517 83627015 41455 270 46 48 APPHIG229 4339517 83627015 41455 270 46 48 APPHIG229 4339517 83627015 41455 270 46 36 APPHIG229 4339517 83627017 41455 270 46 36 APPHIG229 4339517 83627017 41455 270 46 36 APPHIG229 4339517 83627017 41455 270 46 36 APPHIG229 4339518 41451 270 46 36 APPHIG229 4339518 8362701 41451 270 46 36 APPHIG239 4339518 8362701 41451 270 46 42 APPHIG240 4339548 8362701 41451 270 46 42 APPHIG241 4339548 8362701 41451 270 46 6 42 APPHIG242 4339548 8362701 41451 270 46 6 42 APPHIG241 4339548 8366701 8366701 41451 270 46 6 6 6 6 6 6 6 6 6 6 6 6 6 6 6 6 6 6		APD415211	433595.66	6830295.4	414.69		-60	
APPHS278		APD415212	433603.8	6830295.3	414.83	270	-60	48
APPOMISSIP APPOMI					_	_		
APPH14220 413506 8630231 415 270 40 48 APPH14224 433566 8630231 41442 270 40 54 APPH14225 4335767 86302701 41443 270 40 54 APPH14225 4335767 86302701 41443 270 40 48 APPH14227 433927.5 86302701 41443 270 40 48 APPH14227 433927.5 86302701 41445 270 40 48 APPH14227 433927.5 86302701 4145 570 40 48 APPH14221 433572 186302701 4145 570 40 48 APPH14223 433566 98 86302701 4145 770 40 48 APPH14223 433567 97 86302701 4145 770 40 48 APPH14223 4335767 86302701 4145 770 40 48 APPH14223 4335767 86302501 4145 770 40 48 APPH14224 433570.7 86302501 4145 770 40 59 APPH14224 433586.8 863026 4146 770 40 59 APPH14224 433586.8 863026 41447 770 40 70 70 70 70 70 70 70 70 70 70 70 70 70								
APPH18224 433968.99 68020715 4144.2 270 40 42 APPH18226 43395.1 68020715 414.5 270 40 48 APPH18226 43395.1 68020715 414.5 270 40 48 APPH18227 43396.7 68020715 414.5 270 40 48 APPH18228 43396.1 68020715 414.5 270 40 48 APPH18224 43396.1 68020715 414.5 270 40 48 APPH18224 43396.1 68020715 414.5 270 40 48 APPH18224 43396.1 680208.1 414.7 270 40 12 APPH18227 4396.1 33 680208.1 414.7 270 40 12 APPH18227 43396.1 38 680208.1 414.7 270 40 12 APPH18228 43396.8 680208.1 414.7 270 40 12 APPH18228 43396.8 680208.1 414.7 270 40 42 APPH18228 43396.8 680208.1 414.8 270 40 36 APPH18228 43396.8					-			-
APPH 19225 4355/6-71 6803020-5 414-55 270 6-0 48 APPH 19227 435962-75 6803020-5 414-55 270 6-0 48 APPH 19227 435962-75 441-55 270 6-0 36 APPH 19223 43596-74 6803027-8 414-55 270 6-0 36 APPH 19233 43596-79 6803027-7 414-5 270 6-0 48 APPH 19235 43596-79 7 6803027-7 414-5 270 6-0 48 APPH 19235 43596-79 7 6803027-7 414-5 270 6-0 48 APPH 19235 43596-79 7 6803027-7 414-5 270 6-0 48 APPH 19235 43596-70 7 6803027-7 414-5 270 6-0 48 APPH 19235 43596-70 7 6803025-7 414-5 270 6-0 48 APPH 19235 43596-70 7 6803025-7 414-5 270 6-0 49 APPH 19235 43596-70 7 6803025-7 414-6 7 270 6-0 12 APPH 19237 43591-7 7 6803025-7 414-7 270 6-0 12 APPH 1924 43596-7 7 6803025-7 414-7 270 6-0 12 APPH 1924 43596-7 7 6803025-7 414-7 270 6-0 12 APPH 1924 43596-7 7 6803025-7 414-7 270 6-0 12 APPH 1924 43596-7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7						-		
## APCH19226 #\$3589.1 \$680270.9 414.55 270 40 48 APCH19272 43396.17 5802071.8 414.5 270 40 48 APCH19272 43396.17 5802071.8 414.5 270 40 48 APCH19272 43396.17 414.5 270 40 48 APCH19273 43386.17 580207.5 414.5 270 40 48 APCH19273 43386.17 580207.5 414.5 270 40 48 APCH19277 43386.13.5 580207.5 414.5 270 40 48 APCH19277 43386.13.5 580207.5 414.5 270 40 48 APCH19277 43386.13.5 580208.8 414.5 270 40 48 APCH19277 43386.13.5 580208.8 414.5 270 40 40 42 APCH19277 43386.13.5 580208.8 414.5 270 40 40 42 APCH19274 43386.8 580208.8 414.5 270 40 42 APCH19247 43386.8 580208.8 414.5 270 40 42 APCH19248 43386.9 580208.8 414.5 270 40 42 APCH19248 43386.9 580208.8 414.5 270 40 42 APCH19247 43386.9 580208.3 414.5 270 40 42 APCH19258 43386.1 580208.3 444.5 270 450 42 APCH19258 43386.1 580208.3 444.5 270 450 38 APCH19258 43386.1 58						_		
APOH15227 433694.74 6830273 414.65 270 40 48 APOH15232 433664.74 6830273 414.65 270 40 36 APOH15233 433665.99 6830273 414.64 270 40 48 APOH15235 433665.91 6830265 414.65 270 40 48 APOH15236 43367.04 6830265 414.65 270 40 48 APOH15257 43861.35 6830265 414.67 270 40 12 APOH1527 43861.35 6800268 414.67 270 40 12 APOH1527 43861.35 6800268 414.67 270 40 12 APOH1527 43861.35 6800268 414.67 270 40 12 APOH1528 43365.60 6800268 414.67 270 40 12 APOH1528 43365.60 6800268 414.67 270 40 42 APOH1528 43365.60 43365.76 6800268 414.67 270 40 42 APOH1528 43365.81 6800263 414.64 270 40 36 APOH1528 43365.81 6800233 414.64 270 40 36 APOH1529 43366.81 6800233 414.69 270 40 36 APOH1529 43366.81 6800233 414.69 270 40 36 APOH1529 43366.60 433620 414.69 270 40 36 APOH1529 43366.60 433620 414.69 270 40 36 APOH1529 43366.60 480020 414.60 270 40 40 40 40 40 40 40 40 40 40 40 40 40								
## APPHIS222 43355474 8830297 8 414 61 270 40 48 ## APPHIS234 433579 7 4 416 61 270 40 48 ## APPHIS234 433579 7 4 416 61 270 40 48 ## APPHIS235 433887 04 6830298 414 34 270 40 48 ## APPHIS237 433513 5 8380298 1 414 51 270 40 48 ## APPHIS237 433513 5 6830298 1 414 51 270 40 35 ## APPHIS237 433513 6 6830298 1 414 51 270 40 35 ## APPHIS237 433513 6 6830298 1 414 51 270 40 35 ## APPHIS234 43362 00 6830298 1 414 52 270 40 42 ## APPHIS234 43362 00 6830298 1 414 52 270 40 42 ## APPHIS234 43362 00 6830298 1 414 52 270 40 42 ## APPHIS234 43362 00 6830298 1 414 52 270 40 42 ## APPHIS234 43362 00 6830298 1 414 52 270 40 42 ## APPHIS234 43362 00 6830298 1 414 52 270 40 36 ## APPHIS234 43362 00 6830298 1 414 52 270 40 36 ## APPHIS234 43362 00 6830298 1 414 52 270 40 36 ## APPHIS235 43365 00 6830298 1 414 52 270 40 36 ## APPHIS235 433650 1 6830298 1 414 52 270 40 36 ## APPHIS235 433650 1 6830298 1 414 56 270 40 30 ## APPHIS235 433650 1 6830208 1 414 56 270 40 30 ## APPHIS235 433650 1 6830208 1 414 56 270 40 30 ## APPHIS235 433650 1 6830208 1 414 56 270 40 30 ## APPHIS235 433650 1 6830208 1 414 56 270 40 30 ## APPHIS235 433650 1 6830208 1 414 59 270 50 50 36 ## APPHIS235 433650 1 6830208 1 414 59 270 50 50 36 ## APPHIS235 433650 1 6830208 1 414 59 270 50 50 36 ## APPHIS235 433650 1 6830208 1 414 59 270 50 50 142 ## APPHIS236 43368 1 6830208 1 414 59 270 50 50 142 ## APPHIS236 43368 1 6830208 1 414 59 270 50 50 142 ## APPHIS236 43368 1 6830208 1 414 59 270 50 50 142 ## APPHIS236 43368 1 6830208 1 414 59 270 50 50 142 ## APPHIS236 43368 1 6830208 1 414 59 270 50 50 142 ## APPHIS236 43368 1 6830208 1 414 59 270 50 50 142 ## APPHIS236 43368 1 6830208 1 414 59 270 50 50 142 ## APPHIS236 43368 1 6830208 1 414 59 270 50 50 142 ## APPHIS236 43368 1 6830208 1 414 59 270 50 50 142 ## APPHIS236 43368 1 6830208 1 50 50 50 50 50 50 50 50 50 50 50 50 50						_		
APPHIS233 43596 98 6800278 4146 31 270 60 48 APPHIS255 43597 7 6800279 4143 31 270 60 48 APPHIS256 4358104 6800268 4146 81 270 60 54 APPHIS256 4358104 6800268 4146 81 270 60 36 54 APPHIS261 435840 60 6800268 4146 81 270 60 36 APPHIS261 435840 60 6800268 4146 81 270 60 36 APPHIS262 435854 80 6800268 4146 81 270 60 36 APPHIS262 435854 80 6800268 4146 81 270 60 42 APPHIS262 435854 80 6800268 4146 81 270 60 42 APPHIS262 435854 80 6800268 4146 81 270 60 42 APPHIS263 435854 81 6800268 4146 81 270 60 42 APPHIS264 43586 94 6800268 4146 270 60 42 APPHIS264 43586 94 6800268 4146 270 60 36 APPHIS264 435852 81 6800268 4146 41 270 60 36 APPHIS264 435852 81 6800268 4146 41 270 60 36 APPHIS264 435852 81 6800268 4146 41 270 60 36 APPHIS264 435859 81 6800268 4146 41 270 60 36 APPHIS265 435857 81 6800268 4146 41 270 60 36 APPHIS260 435859 81 6800233 4146 80 270 60 36 APPHIS260 435859 81 6800233 4146 81 270 60 36 APPHIS260 435859 81 6800233 4146 81 270 60 36 APPHIS260 435859 81 6800233 4146 81 270 60 36 APPHIS260 435859 81 6800233 4146 81 270 60 30 APPHIS260 435859 81 6800233 4146 81 270 60 30 APPHIS260 435859 81 6800203 4146 91 270 60 30 APPHIS260 435859 81 6800203 4146 91 270 60 30 APPHIS260 435864 91 6800200 4146 91 270 60 30 APPHIS260 435864 91 6800200 4146 91 270 60 36 APPHIS260 435864 91 6800200 4146 91 270 60 36 APPHIS260 435864 91 6800200 4146 91 270 60 36 APPHIS260 435864 91 6800200 4146 91 270 60 36 APPHIS260 435864 91 6800200 4146 91 270 60 36 APPHIS260 435864 91 6800200 4146 91 270 60 36 APPHIS260 435864 91 6800200 4146 91 270 60 91 42 APPHIS260 435864 91 6800200 4146 91 270 60 91 42 APPHIS260 435864 91 6800200 4146 91 270 60 91 42 APPHIS260 435864 91 6800200 4146 91 270 60 91 42 APPHIS260 435864 91 6800200 4146 91 270 60 91 42 APPHIS260 435864 91 6800200 4146 91 270 60 91 42 APPHIS260 435864 91 6800200 4146 91 270 60 91 42 APPHIS260 435864 91 6800200 4146 91 270 60 91 42 APPHIS260 435853 71 680076 91 4146 91 270 60 91 42 APPHIS260 435853 71 680076 91 4146 91 270 60 91 42 APPHIS260 435853 71 680076 91 4146 91 270 60 9								
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APPM 15243		APD415241	433548.06	6830245.4	414.8	270	-60	36
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APDHIS245 433875.76 6830246.4 414.52 270 -60 42 APDHIS247 43389.81 6830246.4 414.47 270 -60 36 APDHIS247 43389.81 6830246.3 414.54 270 -60 36 APDHIS250 43389.83 6830233.3 414.66 270 -60 30 APDHIS251 43385.81 6830233.3 414.66 270 -60 30 APDHIS256 43385.27 6830231 414.66 270 -60 30 APDHIS256 43385.27 6830220.4 414.54 270 -60 30 APDHIS256 43385.27 6830220.5 414.99 270 -60 30 APDHIS257 433847 6830220.5 414.99 270 -60 36 APDHIS258 43385.27 6830220.4 414.92 270 -60 36 APDHIS259 43385.27 6830220.4 414.92 270 -60 36 APDHIS269 43386.16 6830220.4 414.92 270 -60 42 APDHIS269 43386.10 6830220.4 414.8 270 -86.1 42 APDHIS268 43385.21 6830220.4 414.9 270 -86.1 42 APDHIS268 43385.20 683020.4 414.9 270 -86.1 42 APDHIS268 43385.21 683027.8 414.8 270 -86.4 42 APDHIS268 43385.20 683027.8 414.9 270 -86.4 42 APDHIS268 43385.20 683027.8 414.9 270 -86.4 42 APDHIS268 43385.20 683027.8 414.9 270 -86.4 42 APDHIS268 43385.20 683027.8 414.5 270 -86.0 36 APDHIS268 43385.6 1 683027.9 414.4 52 270 -86.0 36 APDHIS268 43385.6 1 683027.9 414.4 52 270 -86.0 36 APDHIS268 43385.8 683019.3 414.4 52 270 -86.0 36 APDHIS268 43385.8 683019.3 414.4 8 270 -86.1 42 APDHIS268 43385.8 683019.3 414.4 8 270 -86.1 42 APDHIS268 43385.3 6 683019.3 414.4 8 270 -86.0 36 APDHIS268 43385.3 6 683019.3 414.4 8 270 -86.0 36 APDHIS270 43385.8 683019.3 414.4 8 270 -85.1 42 APDHIS278 43385.8 683017.1 414.8 270 -86.1 36 APDHIS278 43385.8 683017.1 414.8 270 -86.1 36 APDHIS278 43385.8 683017.1 414.8 270 -86.1 36 APDHIS288 43385.3 6 683017.1 414.8 270 -86.1 36 APDHIS288 43385.3 6 683017.1 414.8 270 -86.1 36 APDHIS281 43385.3 6 683017.3 414.8 270 -86.1 36 APDHIS281 43385.3 6 683017.3 414.8 3 270 -86.1 42 APDHIS281 43385.3 6 683017.3 414.8 3 270 -86		APD415243	433562.02	6830245.4	414.77	270	-60	42
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APD415324 433544.19 6829995.7 414.82 270 -59.5 42 APD415325 433556.9 6829995.7 414.92 270 -59.6 42 APD415326 433529.47 6829982.9 412.03 270 -60 36 APD415327 433551.52 6829983.7 411.28 270 -60 36 APD415329 433529.7 6829970 412.75 270 -60 40 APD415330 433540.83 6829970.4 412.34 270 -60 40 ECGC0007 434102.07 6831327.6 441.48 270 -60 48 ECGC0029 434112.15 6831387.4 442.67 270 -60 48 ECGC0030 434101.8 6831387.8 442.4 270 -60 48								
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APD415326 433529.47 6829982.9 412.03 270 -60 36 APD415327 433551.52 6829983.7 411.28 270 -60 36 APD415329 433529.7 6829970 412.75 270 -60 40 APD415330 433540.83 6829970.4 412.34 270 -60 40 ECGC0007 434102.07 6831327.6 441.48 270 -60 48 ECGC0029 434112.15 6831387.4 442.67 270 -60 48 ECGC0030 434101.8 6831387.8 442.4 270 -60 48					-	-		
APD415327 433551.52 6829983.7 411.28 270 -60 36 APD415329 433529.7 6829970 412.75 270 -60 40 APD415330 433540.83 6829970.4 412.34 270 -60 40 ECGC0007 434102.07 6831327.6 441.48 270 -60 48 ECGC0029 434112.15 6831387.4 442.67 270 -60 48 ECGC0030 434101.8 6831387.8 442.4 270 -60 48								
APD415329 433529.7 6829970 412.75 270 -60 40 APD415330 433540.83 6829970.4 412.34 270 -60 40 ECGC0007 434102.07 6831327.6 441.48 270 -60 48 ECGC0029 434112.15 6831387.4 442.67 270 -60 48 ECGC0030 434101.8 6831387.8 442.4 270 -60 48								
APD415330 433540.83 6829970.4 412.34 270 -60 40 ECGC0007 434102.07 6831327.6 441.48 270 -60 48 ECGC0029 434112.15 6831387.4 442.67 270 -60 48 ECGC0030 434101.8 6831387.8 442.4 270 -60 48						_		
ECGC0029 434112.15 6831387.4 442.67 270 -60 48 ECGC0030 434101.8 6831387.8 442.4 270 -60 48		APD415330	433540.83	6829970.4	412.34	270	-60	40
ECGC0030 434101.8 6831387.8 442.4 270 -60 48		ECGC0007	434102.07	6831327.6	441.48	270	-60	48
		ECGC0029	434112.15	6831387.4	442.67	270	-60	48
ECGC0044 434084.84 6831417.6 442.25 270 -60 30								
		ECGC0044	434084.84	6831417.6	442.25	270	-60	30

Criteria	Commentary						
	ECGC41505	434105.98	6831310.9	414.28	270	-60	24
	ECGC41506	434098.84	6831311.8	414.47	270	-60	24
	ECGC41508	434114	6831319.4	414.6	270	-70	36
	ECGC41509	434105.98	6831320	414.5	270	-60	33
	ECGC41510	434098.18	6831320.5	414.66	270	-60	30
	ECGC41513	434111.51	6831329.4	414.86	270	-65	38
	ECGC41515	434102.44	6831329.5	415.02	270	-60	30
	ECGC41516	434098.02	6831329.4	414.98	270	-60	30
	ECGC41517	434091.43	6831329.8	414.99	270	-60	24
	ECGC41520	434114.89	6831340.1	415.22	270	-65	36
	ECGC41521	434098.54	6831339.8	415.18	0	-90	36
	ECGC41526	434115.64	6831350	415	270	-60	40
	ECGC41527	434109.64	6831350	415	270	-60	36
	ECGC41532	434116.79	6831359.8	415.09	270	-60	36
	ECGC41534	434094.33	6831360.2	414.89	270	-60	24
	ECGC41536	434109.56	6831369.6	414.82	270	-60	30
	ECGC41537	434104.2	6831369.9	414.82	270	-60	30
	ECGC41538	434098.63	6831370.2	414.93	270	-60	24
	ECGC41541	434102.26	6831380.1	415.04	270	-60	24
	ECGC41542	434095.6	6831379.8	414.78	270	-60	24
	ECGC41544	434102.3	6831390	414.96	270	-60	24
	ECGC41545	434090.75	6831390	414.93	270	-60	18
	ECGC41548	434086.08	6831399.5	414.85	270	-60	12

Vest Laverton

Historic drilling information has been validated against publicly available WAMEX reports. Not all drill holes can be found referenced in the WAMEX reports.

Company	Drill Hole Number	WAMEX Report A-Number	WAMEX Report Date
HILLMIN	WL10, WL11, WL12, WL13, WL15, WL18, WL7, WL8, WL9	17871	Dec-85
	WL26, WL30, WL31, WL32, WL33	20650	Feb-87
	BD1, BD3, BD4, BD5	23452	Feb-88
	WL34, WL35, WL36, WL37, WL38, WL39, WL40, WL41, WL42, WL43, WL44	23455	Feb-88
	BD8, BD9, BD10, BD11, BD12, BD13, BD14, BD15, BD16, BD17, BD18, BD19, BD20, BD21, BD22, BD23, BD24, BD25, BD26, BD27, BD28, BD31, BD32, BD33, BD34, BD35, BD37, BD38, BD39, BD40, BD41, BD42, BD7, WL101, WL102, WL103, WL105, WL106, WL107, WL110, WL111, WL112, WL113, WL114, WL115, WL116, WL117, WL118, WL119, WL120, WL121, WL51, WL52, WL53, WL54, WL55, WL56, WL58, WL59, WL60, WL61, WL62, WL63, WL64, WL65, WL66, WL67, WL68, WL70, WL71, WL72, WL73, WL75, WL76, WL77, WL78, WL80, WL81, WL82, WL83, WL84, WL86, WL88, WL89, WL90, WL91, WL92, WL95, WL96, WL97, WL98, WL99, WLD1, WLD10, WLD11, WLD12, WLD2,	27622	Feb-06 Jun-88
ASHTON	WLD3, WLD4, WLD5, WLD6, WLD7, WLD8, WLD9 BD100_WESTLA, BD101_WESTLA, BD103_WESTLA, BD105_WESTLA, BD107_WESTLA, BD108_WESTLA, BD109_WESTLA, BD110_WESTLA, BD1110_WESTLA, BD1111_WESTLA, BD1112_WESTLA, BD113_WESTLA, BD114_WESTLA, BD115_WESTLA, BD116_WESTLA, BD117_WESTLA, BD116_WESTLA, BD112_WESTLA, BD122_WESTLA, BD123_WESTLA, BD126, BD127, BD45, BD46, BD47, BD50, BD51, BD52, BD56, BD57, BD58, BD59, BD62, BD67, BD69, BD70, BD73, BD74, BD75, BD76, BD78, BD79, BD80, BD81, BD82, BD83, BD85, BD86, BD94, BD95, BD97, BD99, BDD1, BDD2, BDD3, BDD4, WL122, WL123, WL124, WL125, WL126, WL127, WL128, WL129, WL130, WL131, WL132, WL133, WL134, WL135, WL136, WL137, WL138, WL139, WL140, WL141, WL142, WL143, WL144, WL145, WL146, WL147, WL148, WL149, WL150, WL151, WL152, WL163, WL164, WL165, WL166, WL167, WL168, WL169, WL170, WL171, WL172, WL173, WL174, WL176, WL192, WL193, WL194, WL196, WL197, WL198, WL200, WL201, WL202, WL203, WL204, WL210, WL221, WL222, WL223, WL225, WL226, WL227, WL228, WL229, WL230, WL232, WL233, WL234, WL233, WL236, WL237, WL238, WL239, WL230, WL232, WL233, WL234, WL236, WL237, WL238, WL239, WL230, WL232, WL233, WL234, WL234, WL235, WL236, WL237, WL238, WL239, WL230, WL230, WL233, WL234, WL235, WL236, WL237, WL238, WL239, WL240	30496 35703	Jan-90 Dec-91
SOG	ENC002, ENC003, ENC004, ENC005, ENC006, ENC007, ENC008, ENC009, ENC011, ENC012, ENC013	51454	May-97
	ENC286, ENC289	55360	Nov-97
	ENC457, ENC458, ENC459, ENC460, ENC461, ENC462	62396	Feb-01
POLLO	WV018, WV019, WV020, WV021, WV023, WV024, WV025, WV026, WV027	68420	Apr-04
CRESCENT	WV029, WV030, WV031, WV032, WV033, WV034, WV035, WV036, WV037, WV038, WV039, WV040, WV041, WV044, WV045, WV046, WV047	74767	Mar-07
	WLDD001, WLDD002, WLDD003, WLRC200, WLRC201, WLRC202, WLRC203, WLRC204, WLRC205, WLRC207	81229	Feb-09
	WLRC209, WLRC210, WLRC211, WLRC212, WLRC213, WLRC215, WLRC217, WLRC218, WLRC219, WLRC220, WLRC222, WLRC223, WLRC224, WLRC225, WLRC226, WLRC227, WLRC228, WLRC229, WLRC230, WLRC231, WLRC232, WLRC233, WLRC234, WLRC235, WLRC244, WLRC242, WLRC243, WLRC244, WLRC245, WLRC246	86387	Feb-10
	WLRC247, WLRC248, WLRC249, WLRC250, WLRC251, WLRC252, WLRC253,	90143	Apr-11

Criteria	Commonton						
Criteria	Commentary	WLRC254, WLRC255, WLRC2	256 WLRC257 WLRC258	WI RC259 WI RC260	<u> </u>		
		WLRC261, WLRC262, WLRC2	63, WLRC264, WLRC265,	WLRC266, WLRC267,			
		WLRC268	4 14# D0070 14# D0070 1	W D0074 14# D0075	20101		
		WLRC269, WLRC270, WLRC27 WLRC276, WLRC277	1, WLRC2/2, WLRC2/3, V	VLRC2/4, WLRC2/5,	98404	Jun-13	
	FOCUS	WLDD004, WLDD005, WLDD00	6		102282	Jun-14	
	West Laverton collar of	etails of holes not previously externa	Illy reported:	MOA 04 7ana 51			
		DII 7/05		MGA 94 Zone 51			DEPTH
	COMPANY I	RILL TYPE HOLE ID RGAC1 4:	EAST NORTH 38822.98 6833124.1	RL 453.06	AZIMUTH 255	-60	(m) 20
	WMC RC		38845.71 6833090.8	453.81	255	-60	30
	WMC RC		38864.36 6833096.6	453.94	255	-60	40
	WMC RC		38892.72 6833194.8 39007.55 6833208.9	452.85 453.99	255 255	-60 -60	40 20
	WMC RC	RGAC23 4	39026.74 6833215.8	454.28	255	-60	30
	WMC RC	RGAC24 4 RGAC25 4	39010.96 6833168.3 39029.13 6833176.5	454.61 454.81	255 255	-60 -60	20 30
	WMC RC	RGAC26 4	39044.89 6833222.4	454.58	255	-60	40
	WMC RC		138994.8 6833184.4	454.19	255	-60	20
	WMC RC		39013.41 6833190.5 39032.79 6833196.9	454.32 454.64	255 255	-60 -60	30 40
	WMC RC	RGAC31 4	39007.94 6833146.2	454.9	255	-60	30
	WMC RC	RGAC32 4 RGAC33 4	39027.06 6833152.3 39045.81 6833158.9	455.06 455.38	255 255	-60 -60	40 40
	SOG RC	WLRC023 4	39043.34 6832760.2	454.1	0	-90	65
	SOG RC	WLRC025 4	38882.88 6832784.5 38903.53 6832784.7	453.18	0	-90	25 35
	SOG RC		38903.53 6832784.7 38923.04 6832784.4	453.31 453.48	0 0	-90 -90	35 45
	SOG RC	WLRC028 4	38942.93 6832785.9	453.45	0	-90	50
	SOG RC	WLRC030 4 WLRC032 4	38887.98 6832809.2 38879.23 6832834.3	453.05 453.02	0 0	-90 -90	25 30
	SOG RC	WLRC033 4	38899.81 6832834.8	453.17	0	-90	40
	SOG RC	WLRC034 4	38917.8 6832835	453.47	0	-90	50
	SOG RC	WLRC035 4: WLRC036 4:	38938.08 6832834.8 38958.76 6832834.7	453.74 453.98	0 0	-90 -90	60 70
	SOG RC	WLRC037 4	38977.87 6832834.6	453.97	0	-90	85
	SOG RC		38998.21 6832834.9 38858.82 6832859.8	454 453.51	0 0	-90 -90	40 15
	SOG RC	WLRC040 4	38902.92 6832859.5	453.55	0	-90	40
	SOG RC	WLRC041 4 WLRC042 4	38925.04 6832859.5 138943.5 6832859.1	453.63 453.65	0 0	-90 -90	50 60
	SOG RC	WLRC042 4	38963.12 6832859.9	453.58	0	-90 -90	70
	SOG RC		38988.07 6832859.6	453.38	0	-90	85
	SOG RC		39033.54 6832860.1 38858.12 6832884.9	453.52 453.11	0 0	-90 -90	100 25
	SOG RC	WLRC047 4	38877.91 6832884.7	453.42	0	-90	25 35
	SOG RC	WLRC048 4 WLRC049 4		453.44 453.17	0	-90 -90	35 45
	SOG RC		38938.73 6832884.7	452.91	0	-90	55
	SOG RC	WLRC051 4	38958.4 6832885.1	452.99	0	-90	65 75
	SOG RC		38979.52 6832887.2 38997.35 6832883.2	453.05 453.16	0	-90 -90	75 85
	SOG RC	WLRC054 4	39019.66 6832884.9	453.27	0	-90	100
	SOG RC	WLRC055 4 WLRC056 4	139037.9 6832884.8 38867.66 6832908.7	453.36 452.77	0	-90 -90	40 30
	SOG RC	WLRC058 4	38912.23 6832909.4	452.82	0	-90	40
	SOG RC	WLRC060 4 WLRC061 4	38988.39 6832909.9 38867.93 6832934.8	452.89 452.24	0	-90	40
	SOG RC	WLRC061 4	38888.32 6832934.7	452.24 452.38	0	-90 -90	30 40
	SOG RC	WLRC063 4	38908.54 6832934.9	452.47	0	-90	40
	SOG RC		138928.1 6832934.5 38947.95 6832934.4	452.58 452.73	0	-90 -90	55 65
	SOG RC	WLRC066	438968 6832935.2	452.83	0	-90	80
	SOG RC		38988.23 6832934.5 39007.92 6832935.5	452.86 453	0	-90 -90	60 70
	SOG RC	WLRC069 4	38873.1 6832960.3	451.91	0	-90	30
	SOG RC	WLRC070 4	38893.63 6832959.6	452.17	0	-90	35
	SOG RC	WLRC071 4 WLRC072 4	38912.73 6832959.9 38948.24 6832959.5	452.44 452.65	0	-90 -90	45 65
	SOG RC	WLRC073 4	38967.95 6832959.5	452.93	0	-90	75
	SOG RC	WLRC076 4 WLRC077 4	138957.9 6832984.4 38978.24 6832984.7	452.51 452.74	0 0	-90 -90	35 50
	SOG RC	WLRC078 4	38998.02 6832985	452.74 452.85	0	-90 -90	60
	SOG RC	WLRC079 4	39017.8 6832988.4	452.88	0	-90	70
	SOG RC	WLRC084 4 WLRC086 4	38962.64 6833036.5 38871.56 6832990.6	452.32 451.97	0	-90 -90	35 35
	SOG RC	WLRC087 4	38877.12 6833040.5	451.78	0	-90	35
	SOG RC	WLRC100 4	39062.68 6832985.9	453.01	0	-60 60	102 90
	SOG RC	WLRC101 4 WLRC102 4	39041.76 6832992.7 39019.58 6833006.8	452.93 452.74	0 0	-60 -60	90 74
	SOG RC	WLRC103 4	39001.03 6833007.5	452.77	0	-60	66
<u></u>	SOG RC	WLRC104 4	38967.14 6833011.1	452.34	0	-60	54

Criteria	Commentary	,								
Oritoria		RC	WLRC	106 438877.48	6832861.5	453.22)	0	-90	25
	SOG	RC	WLRC			454.26		Ō	-90	100
	SOG	RC	WLRC	108 439050.07		452.94		0	-90	90
	West Laverton sha		RC grade control h	oles not externally	reported					
		DRILL					MGA 94 Zone 51			DEPTH
	COMPANY	TYPE	HOLE ID	EAST	NORTH		RL 450.00	AZIMUTH	DIP	(m)
	WMC WMC	RC RC	RGAC1 RGAC12	438822.98 438845.71	6833124.1 6833090.8		453.06 453.81	255 255	-60 -60	20 30
	WMC	RC	RGAC12 RGAC13	438864.36	6833096.6		453.61	255 255	-60 -60	40
	WMC	RC	RGAC13	438892.72	6833194.8		452.85	255	-60	40
	WMC	RC	RGAC22	439007.55	6833208.9		453.99	255	-60	20
	WMC	RC	RGAC23	439026.74	6833215.8		454.28	255	-60	30
	WMC	RC	RGAC24	439010.96	6833168.3		454.61	255	-60	20
	WMC	RC	RGAC25	439029.13	6833176.5		454.81	255	-60	30
	WMC	RC	RGAC26	439044.89	6833222.4		454.58	255	-60	40
	WMC	RC	RGAC27	438994.8	6833184.4		454.19	255	-60	20
	WMC	RC	RGAC28	439013.41	6833190.5		454.32	255	-60	30
	WMC	RC	RGAC29	439032.79	6833196.9		454.64	255	-60	40
	WMC	RC	RGAC31	439007.94	6833146.2		454.9	255	-60	30
	WMC	RC	RGAC32	439027.06	6833152.3		455.06	255	-60	40
	WMC	RC	RGAC33	439045.81	6833158.9		455.38	255	-60	40
	SOG	RC	WLRC023	439043.34	6832760.2		454.1	0	-90	65
	SOG	RC	WLRC025	438882.88	6832784.5		453.18	0	-90	25
	SOG	RC	WLRC026	438903.53	6832784.7		453.31	0	-90	35
	SOG	RC DC	WLRC027	438923.04	6832784.4		453.48	0	-90 00	45 50
	SOG SOG	RC RC	WLRC028 WLRC030	438942.93 438887.98	6832785.9 6832809.2		453.45 453.05	0 0	-90 -90	50 25
	SOG	RC	WLRC030 WLRC032	438879.23	6832834.3		453.05 453.02	0	-90 -90	30
	SOG	RC	WLRC032 WLRC033	438899.81	6832834.8		453.02 453.17	0	-90 -90	40
	SOG	RC	WLRC033 WLRC034	438917.8	6832835		453.47	0	-90 -90	50
	SOG	RC	WLRC035	438938.08	6832834.8		453.74	0	-90 -90	60
	SOG	RC	WLRC036	438958.76	6832834.7		453.98	0	-90	70
	SOG	RC	WLRC037	438977.87	6832834.6		453.97	0	-90	85
	SOG	RC	WLRC038	438998.21	6832834.9		454	0	-90	40
	SOG	RC	WLRC039	438858.82	6832859.8		453.51	0	-90	15
	SOG	RC	WLRC040	438902.92	6832859.5		453.55	0	-90	40
	SOG	RC	WLRC041	438925.04	6832859.5		453.63	0	-90	50
	SOG	RC	WLRC042	438943.5	6832859.1		453.65	0	-90	60
	SOG	RC	WLRC043	438963.12	6832859.9		453.58	0	-90	70
	SOG	RC	WLRC044	438988.07	6832859.6		453.38	0	-90	85
	SOG	RC	WLRC045	439033.54	6832860.1		453.52	0 0	-90 00	100
	SOG SOG	RC RC	WLRC046 WLRC047	438858.12 438877.91	6832884.9 6832884.7		453.11 453.42	0	-90 -90	25 35
	SOG	RC	WLRC047 WLRC048	438898.49	6832885		453.44 453.44	0	-90 -90	35
	SOG	RC	WLRC049	438918.35	6832885		453.17	0	-90	45
	SOG	RC	WLRC050	438938.73	6832884.7		452.91	0	-90	55
	SOG	RC	WLRC051	438958.4	6832885.1		452.99	0	-90	65
	SOG	RC	WLRC052	438979.52	6832887.2		453.05	0	-90	75
	SOG	RC	WLRC053	438997.35	6832883.2		453.16	0	-90	85
	SOG	RC	WLRC054	439019.66	6832884.9		453.27	0	-90	100
	SOG	RC	WLRC055	439037.9	6832884.8		453.36	0	-90	40
	SOG	RC	WLRC056	438867.66	6832908.7		452.77	0	-90	30
	SOG	RC	WLRC058	438912.23	6832909.4		452.82	0	-90	40
	SOG	RC	WLRC060	438988.39	6832909.9		452.89	0	-90	40
	SOG	RC	WLRC061	438867.93	6832934.8		452.24	0	-90 00	30
	SOG	RC BC	WLRC062	438888.32	6832934.7		452.38	0	-90 00	40
	SOG SOG	RC RC	WLRC063 WLRC064	438908.54 438928.1	6832934.9 6832934.5		452.47 452.58	0 0	-90 -90	40 55
	SOG	RC RC	WLRC064 WLRC065	438928.1	6832934.5		452.58 452.73	0	-90 -90	55 65
	SOG	RC	WLRC065 WLRC066	438968	6832935.2		452.73 452.83	0	-90 -90	80
	SOG	RC	WLRC067	438988.23	6832934.5		452.86	0	-90 -90	60
	SOG	RC	WLRC068	439007.92	6832935.5		453	0	-90	70
	SOG	RC	WLRC069	438873.1	6832960.3		451.91	0	-90	30
	SOG	RC	WLRC070	438893.63	6832959.6		452.17	0	-90	35
	SOG	RC	WLRC071	438912.73	6832959.9		452.44	0	-90	45
	SOG	RC	WLRC072	438948.24	6832959.5		452.65	0	-90	65
	SOG	RC	WLRC073	438967.95	6832959.5		452.93	0	-90	75
	SOG	RC	WLRC076	438957.9	6832984.4		452.51	0	-90	35
	SOG	RC	WLRC077	438978.24	6832984.7		452.74	0	-90	50
	SOG	RC	WLRC078	438998.02	6832985		452.85	0	-90	60
	SOG	RC	WLRC079	439017.8	6832988.4		452.88	0	-90	70
	SOG	RC	WLRC084	438962.64	6833036.5		452.32	0	-90	35
	SOG	RC	WLRC086	438871.56	6832990.6		451.97	0	-90	35
	SOG	RC	WLRC087	438877.12	6833040.5		451.78	0	-90	35
	SOG	RC	WLRC100	439062.68	6832985.9		453.01	0	-60	102

Criteria	Commentary	/							
	SOG	RC	WLRC101	439041.76	6832992.7	452.93	0	-60	90
	SOG	RC	WLRC102	439019.58	6833006.8	452.74	0	-60	74
	SOG	RC	WLRC103	439001.03	6833007.5	452.77	0	-60	66
	SOG	RC	WLRC104	438967.14	6833011.1	452.34	0	-60	54
	SOG	RC	WLRC106	438877.48	6832861.5	453.22	0	-90	25
	SOG	RC	WLRC107	439012.66	6832834.9	454.26	0	-90	100
	SOG	RC	WLRC108	439050.07	6832985.9	452.94	0	-90	90

Gladiator
Historic drilling information has been validated against publicly available WAMEX reports. Not all drill holes can be found referenced in the WAMEX reports.
However, cross-checking of original drill surveys was verified against the database.

Company	Drill Hole Number	WAMEX Report A- Number	WAMEX Report Date
Teck Exploration Ltd	GP1, GP2	11969	01-Jan-83
Technominerals	GRC1, GRC3	20213	01-Mar-87
Hill Minerals	GP7, GP8	15071	01-Mar-85
	GP11, GP12, GP14, GP16,GP17,GP18	17467	01-Feb-86
	GP25, GP26, GP27, GP28, GP29, GP30	27702	01-Dec-88
	GP31, GP32, GP33, GP34, GP35, GP36, GP37, GP38, GP39, GP40	27703	01-Feb-89
WMC	TWP107, TWP108, TWP109, TWP110, TWP111, TWP112, TWP113, TWP114	22647	31-Jan-88
	TWP175, TWP176, TWP177, TWP178, TWP179, TWP180	35126	01-Feb-92
Ashton	GP100, GP101, GP116, GP119, GP120, GP123, GP126, GP127, GP130, GP131, GP132, GP135, GP142, GP143, GP144, GP145, GP146, GP147, GP148, GP149, GP151, GP152, GP161, GP162, GP163, GP164, GP168, GP41, GP42, GP43, GP44, GP45, GP46, GP47, GP48, GP49, GP50, GP51, GP52, GP53, GP54, GP55, GP56, GP57, GP58, GP59, GP60, GP61, GP62, GP63, GP64, GP65, GP66, GP67, GP68, GP69, GP70, GP71, GP72, GP73, GP74, GP82, GP83, GP84, GP85, GP86, GP87, GP88, GP89, GP90, GP93, GP94, GP95, GP96, GP97, GP98, GP99, GSD1, GSD2, GSD4, GSD5	17957	01-Jan-90
	GP102, GP103, GP105, GP106, GP107, GP109, GP110, GP111, GP112, GP113, GP114, GP115, GP118, GP121, GP122, GP124, GP125, GP128, GP129, GP133, GP134, GP136, GP137, GP138, GP139, GP141, GP150, GP153, GP155, GP156, GP157, GP158, GP159, GP160, GP165, GP169, GP76, GP77, GP78, GP79, GP80, GP81, GSD3, GSD6, GSD7	30488	01-Jan-90
	GP174, GP175, GP176	34630	01-Sep-91
	GP171, GP177, GP178, GP179, GP180, GP181, GP182, GP183, GP184, GP185, GP186, GP187, GP188, GP189, GP190, GP191, GP192, GP193, GP194, GP195, GP196, GP197, GP199, GP200, GP201, GP202, GP203, GP204, GP205, GP206, GP207, GP208, GP209, GP210, GP211, GP212, GP213, GP214, GP215, GP216, GP217, GP218, GP219, GP220, GP221, GP222, GP224, GP225, GP226, GP227, GP228, GP229, GP230, GP231, GP232, GP234, GP235, GP236, GP237, GP238, GP239, GP240, GP241, GP242	34657	01-Sep-91
	GP244, GP245, GP249, GP251, GP252, GP255, GP256, GP257, GP258, GP259, GP265, GP266, GP270, GP271, GP277	35680	01-Jan-92
SOG	ENC298, ENC299, ENC301, ENC302, ENC303, ENC306, ENC307, ENC308, ENC310, ENC311, ENC312	55360	30-Nov-97
	ENC465	62396	28-Feb-01
Metex Resources	GMDH001, GMDH002, GMDH003, GMRC001, GMRC004, GMRC005, GMRC006, GMRC007, GMRC008, GMRC009, GMRC010, GMRC012, GMRC013, GMRC014, GMRC015, GMRC016, GMRC018, GMRC019, GMRC020, GMRC021, GMRC023, GMRC024, GMRC026, GMRC029, GMRC030	69813	01-Feb-05
	GMDH004, GMRC049, GMRC031, GMRC032	72705	01-Mar-06
Metex/Barrick (Granny Smith)	LJC0014, LJC0015, LJC0016, LJC0017, LJC0018, LJC0032	72705	01-Mar-06
Pty Ltd	LJC0033, LJC0034, LJC0035, LJC0036	75073	01-Mar-07

Gladiator FML previously reported drilling information not yet available on WAMEX reports:

Gladiator TWE previously reported drilling information flot yet available	on with the ports.	
Drill Hole Number	ASX Release Title	ASX Release Date
19GLRC001, 19GLRC002, 19GLRC003, 9GLRC004,19GLRC005,	Strong gold hits from Layerton regional drilling campaign	29-Jan-20
19GLRC006, 19GLRC007,	Carong gold files from Edverton regional drilling campaign	
21GLRC001, 21GLRC002, 21GLRC003,21GLRC004,		
21GLRC005, 21GLRC006, 21GLRC007,21GLRC008,	Exploration Update - Laverton Gold Project	28-Apr-21
21GLRC009, 21GLRC010, 21GLRC011	,	,

Gladiator Ashton dhiled	RC noies not externally reported					
		MGA 94 Z	Zone 51			
HOLE ID	EAST	NORTH	RL	AZIMUTH	DIP	DEPTH (m)
GP304	437617.25	6832650.1	457.72	0	-90	100
GP306	437679.17	6832638.7	456.53	0	-90	100
GP307	437676.68	6832816.9	457.76	270	-60	65
GP308	437701.83	6832867.9	463.12	270	-60	65
GP309	437743.55	6832840.4	464.29	270	-60	60
GP310	437727.69	6832791.5	461.85	270	-60	60
GP311	437724.41	6832816.9	462.2	270	-60	60
GP312	437698.71	6832742.9	460.19	270	-60	60
GP313	437668.3	6832793.9	457.33	270	-60	68
GP314	437699.83	6832892.8	466.25	270	-60	60
GP315	437675.8	6832868.7	460.29	270	-60	80

Criteria	Commentary						
	GP316	437670.48	6832844	459.37	270	-60	70
	GP317	437584.32	6832999.1	459.19	0	-90	94
	GP318	437613.39	6833000.1	458.24	0	-90	118
	GP320	437722.97	6832817.6	462.11	270	-60	105
	GP321	437707.93	6832793.5	460.55	270	-60	93
	GP322	437686.76	6832835.1	458.37	225	-60	81
	GP323	437699.04	6832769.6	460.72	270	-60	60
	GP324	437749.16	6832766.8	460.07	270	-60	80
	GP325	437749.66	6832791.6	461.16	270	-60	100
	GP326	437750.8	6832816.6	462.37	270	-60	150
	GP329	437730.26	6832792.3	461.75	270	-60	90
	GP5	437184.81	6832777.8	461.74	272	-60	42

Craigiemore/Mary Mac

Historic drilling information has been validated against publicly available WAMEX reports. Not all drill holes can be found referenced in the WAMEX reports. However, cross-checking of original drill surveys was verified against the database.

		WAMEX Report A-	WAMEX Repo
Company	Drill Hole Number	Number	Date
Hillmin	SL16, SL17, SL18, SL19, SL20, SL21, SL22, SL23, SL26, SL27, SL28, SL29 WESTLAV, SL31,	14966	Apr-85
1 1111111111	SL10, SL17, SL10, SL13, SL20, SL21, SL22, SL23, SL20, SL21, SL20, SL21, SL20, SL21, SL32 WESTLAV, SL33, SL34, SL35, SL36, SL37, SL38, SL39, SL40, SL41, SL42, SL43, SL44, SL46	14300	Αρι-03
	SL80. SLD1. SLD2. SLD4. SLD6. SLD7. SLD8	17424	Feb-86
	SL106, SL107, SL108, SL109	20572	Dec-86
	SL117, SL118, SL119, SL120, SL121, SL122, SL123, SL126, SL127, SL128, SL129, SL130, SL131,	23452	Jan-88
	SL132, SL133, SL134, SL135, SL136, SL137, SL138, SL140, SL141, SL142, SL143, SL144, SL146,	25452	Jan-00
	SL147, SL148, SL149, SL150, SL151, SL152, SL153, SL154, SL155, SL156, SL157, SL158, SL159,		
	SL160, SL161, SL162, SL164, SL166, SL167, SL168, SL169, SL171, SL173, SL174, SL175, SL176,		
	SL180, SL181, SL182, SL183, SL184, SL185, SL186, SL187, SL188, SL190, SL191, SL192, SL193,		
	SL194, SL195, SL196, SL197, SL198, SL199		
	SL206, SL207, SL208, SL209, SL210, SL212, SL213, SL214, SL215, SL216, SL217, SL218, SL219,	27853	Feb-89
	SL220, SL221, SL222, SL223, SL224, SL227, SL228, SL229, SL230, SL231, SL232, SL233, SL234,		
	SL235, SL236, SL237, SL238, SL239, SL240, SL241, SL242, SL247, SL248, SL249, SL250, SL251,		
	SL252, SL253, SL254, SL256, SL257, SL258, SL259, SL261, SL262, SL263, SL265, SL266, SL267,		
	SL268, SL270, SL271,		
	SL272, SL273, SL274, SL275, SL276, SL278, SL279, SL281, SL282, SL283, SL284, SL285, SL286,		
	SL287, SL288, SL289, SL291, SL292, SL297, SL298, SL299, SL301, SL302, SL314, SL315, SL316,		
	SL317, SL319, SL320, SL324, SL325, SL326, SL327, SL328, SL329, SL332, SL334, SL358, SL359,		
	SL360, SL368, SL369, SL371, SL372, SL374, SL375, SL383, SL386, SL388, SL389, SL390, SL391,		
	SL393, SL394, SL395, SL396, SL397, SL398, SL399, SL400, SL401, SL402, SL403, SL405, SL406,		
	SL407, SL408, SL409, SL410, SL411, SL419, SL420, SL421, SL422, SL423, SL424, SLD11, SLD12,		
	SLD13, SLD14, SLD16, SLD17, SLD19, SLD21, SLD22, SLD23, SLD24, SLD25, SLD27		
Ashton	SL547, SL548, SL549, SL550	35678	Jan-92
Sons of	ENC316, ENC317, ENC318, ENC319, ENC320, ENC321, ENC323	59191	Nov-99
Gwalia			
Apollo	WV001, WV002, WV004, WV008, WV009, WV012, WV013, WV014, WV015, WV016, WV028	68420	Apr-04
Crescent	CMRC001, CMRC002A, CMRC004, CMRC005, CMRC006, CMRC007	74767	Mar-07
	CMD001, CMRC008, CMRC009, CMRC010, CMRC011, CMRC012, CMRC013, CMRC014,	77949	Apr-08
	CMRC015, CMRC016, CMRC017, CMRC018, CMRC019, CMRC020, CMRC021, CMRC022,		
	CMRC023, CMRC024, CMRC025, CMRC026, CMRC027, CMRC028, CMRC029, CMRC030,		
	CMRC031, CMRC032, CMRC033, CMRC034, CMRC035, CMRC036, CMRC037, CMRC038,		
	CMRC039, CMRC040, CMRC041, CMRC046, CMRC047, CMRC049, CMRC050, CMRC051,		
	CMRC052, CMRC053, CMRC054, CMRC055, CMRC056, CMRC057, CMRC058, CMRC059,		
	CMRC060, CMRC061, CMRC062, CMRC063, CMRC064, CMRC065, CMRC067, CMRC068,		
	CMRC069, CMRC070, CMRC071, CMRC072, CMRC073, CMRC076, CMRC077,		
	CMRC078, CMRC079	04000	F 1 00
	CMRC082, CMRC083, CMRC084, CMRC085, CRNRC001	81229	Feb-09
	CMRC201, CMRC202, CMRC203, CMRC204, CMRC205, CMRC206, CMRC207, CMRC208,	86387	Feb-10
		00001	
	CMRC209, CMRC210, CMRC211, CMRC212, CMRC213, CMRC214, CMRC215, CMRC216,	00001	
	CMRC209, CMRC210, CMRC211, CMRC212, CMRC213, CMRC214, CMRC215, CMRC216, CMRC217, CMRC219, CMRC221, CMRC222, CMRC223, CMRC224, CMRC225, CMRC226,	00007	
	CMRC209, CMRC210, CMRC211, CMRC212, CMRC213, CMRC214, CMRC215, CMRC216, CMRC217, CMRC219, CMRC221, CMRC222, CMRC223, CMRC224, CMRC225, CMRC226, CMRC227, CMRC228		Apr 11
	CMRC209, CMRC210, CMRC211, CMRC212, CMRC213, CMRC214, CMRC215, CMRC216, CMRC217, CMRC219, CMRC221, CMRC222, CMRC223, CMRC224, CMRC225, CMRC226, CMRC227, CMRC228 CMRC229, CMRC230, CMRC231, CMRC232, CMRC233, CMRC234, CMRC235, CMRC236,	90143	Apr-11
	CMRC209, CMRC210, CMRC211, CMRC212, CMRC213, CMRC214, CMRC215, CMRC216, CMRC217, CMRC219, CMRC221, CMRC222, CMRC223, CMRC224, CMRC225, CMRC226, CMRC227, CMRC228 CMRC229, CMRC230, CMRC231, CMRC232, CMRC233, CMRC234, CMRC235, CMRC236, CMRC237, CMRC238, CMRC239, CMRC240, CMRC241, CMRC242, CMRC243, CMRC245,		Apr-11
	CMRC209, CMRC210, CMRC211, CMRC212, CMRC213, CMRC214, CMRC215, CMRC216, CMRC217, CMRC219, CMRC221, CMRC222, CMRC223, CMRC224, CMRC225, CMRC226, CMRC227, CMRC228 CMRC227, CMRC230, CMRC231, CMRC232, CMRC233, CMRC234, CMRC235, CMRC236, CMRC237, CMRC238, CMRC239, CMRC240, CMRC241, CMRC242, CMRC243, CMRC245, CMRC246, CMRC247, CMRC248, CMRC249, CMRC251, CMRC255, CMRC255,		Apr-11
	CMRC209, CMRC210, CMRC211, CMRC212, CMRC213, CMRC214, CMRC216, CMRC216, CMRC217, CMRC219, CMRC221, CMRC222, CMRC223, CMRC224, CMRC225, CMRC226, CMRC227, CMRC228, CMRC227, CMRC230, CMRC231, CMRC231, CMRC233, CMRC234, CMRC235, CMRC236, CMRC237, CMRC238, CMRC239, CMRC240, CMRC241, CMRC242, CMRC243, CMRC245, CMRC246, CMRC247, CMRC248, CMRC249, CMRC252, CMRC253, CMRC255, CMRC256, CMRC257, CMRC258, CMRC259, CMRC250, CMRC261, CMRC262, CMRC263, CMRC261, CMRC262, CMRC263, CMRC263, CMRC263, CMRC263, CMRC263, CMRC263, CMRC263, CMRC263, CMRC263, CMRC261, CMRC262, CMRC263, CMRC263, CMRC261, CMRC262, CMRC263, CMRC261, CMR		Apr-11
	CMRC209, CMRC210, CMRC211, CMRC212, CMRC213, CMRC214, CMRC216, CMRC216, CMRC217, CMRC219, CMRC221, CMRC222, CMRC223, CMRC224, CMRC225, CMRC226, CMRC227, CMRC228, CMRC221, CMRC223, CMRC230, CMRC230, CMRC231, CMRC231, CMRC233, CMRC234, CMRC235, CMRC236, CMRC237, CMRC238, CMRC239, CMRC240, CMRC241, CMRC242, CMRC243, CMRC245, CMRC246, CMRC247, CMRC248, CMRC249, CMRC251, CMRC252, CMRC253, CMRC255, CMRC256, CMRC257, CMRC258, CMRC259, CMRC260, CMRC261, CMRC262, CMRC263, CMRC264, CMRC265, CMRC266, CMRC266, CMRC266, CMRC266, CMRC271,		Apr-11
	CMRC209, CMRC210, CMRC211, CMRC212, CMRC213, CMRC214, CMRC216, CMRC216, CMRC217, CMRC219, CMRC221, CMRC222, CMRC223, CMRC224, CMRC225, CMRC226, CMRC227, CMRC220, CMRC221, CMRC223, CMRC223, CMRC224, CMRC235, CMRC236, CMRC237, CMRC230, CMRC231, CMRC231, CMRC233, CMRC234, CMRC235, CMRC236, CMRC237, CMRC238, CMRC239, CMRC240, CMRC241, CMRC242, CMRC243, CMRC245, CMRC246, CMRC247, CMRC248, CMRC249, CMRC251, CMRC252, CMRC253, CMRC255, CMRC256, CMRC257, CMRC258, CMRC259, CMRC260, CMRC261, CMRC262, CMRC263, CMRC264, CMRC265, CMRC266, CMRC266, CMRC267, CMRC268, CMRC269, CMRC270, CMRC271, CMRC272, CMRC273, CMRC274, CMRC276, CMRC277, CMRC278, CMRC279, CMRC280,		Apr-11
	CMRC209, CMRC210, CMRC211, CMRC212, CMRC213, CMRC214, CMRC215, CMRC216, CMRC217, CMRC219, CMRC221, CMRC222, CMRC223, CMRC224, CMRC225, CMRC226, CMRC227, CMRC228 CMRC227, CMRC230, CMRC231, CMRC231, CMRC233, CMRC234, CMRC235, CMRC236, CMRC237, CMRC238, CMRC239, CMRC240, CMRC241, CMRC242, CMRC245, CMRC246, CMRC247, CMRC248, CMRC249, CMRC251, CMRC252, CMRC253, CMRC255, CMRC256, CMRC257, CMRC258, CMRC259, CMRC260, CMRC261, CMRC262, CMRC263, CMRC264, CMRC265, CMRC265, CMRC266, CMRC267, CMRC267, CMRC268, CMRC270, CMRC271, CMRC272, CMRC273, CMRC274, CMRC276, CMRC277, CMRC278, CMRC279, CMRC280, CMRC281, CMRC281, CMRC284, CMRC284, CMRC284, CMRC286, CMRC286, CMRC286, CMRC269, CMRC269, CMRC260, CMRC260, CMRC276, CMRC276, CMRC276, CMRC277, CMRC276, CMRC276, CMRC276, CMRC276, CMRC276, CMRC276, CMRC281, CMRC281, CMRC281, CMRC281, CMRC281, CMRC281, CMRC281, CMRC286, CMRC286, CMRC286, CMRC296,		Apr-11
	CMRC209, CMRC210, CMRC211, CMRC212, CMRC213, CMRC214, CMRC215, CMRC216, CMRC217, CMRC219, CMRC221, CMRC222, CMRC223, CMRC224, CMRC225, CMRC226, CMRC227, CMRC228, CMRC221, CMRC221, CMRC222, CMRC223, CMRC234, CMRC235, CMRC236, CMRC237, CMRC230, CMRC231, CMRC231, CMRC231, CMRC231, CMRC231, CMRC231, CMRC231, CMRC231, CMRC231, CMRC245, CMRC246, CMRC247, CMRC248, CMRC249, CMRC251, CMRC252, CMRC253, CMRC255, CMRC256, CMRC257, CMRC258, CMRC259, CMRC260, CMRC261, CMRC262, CMRC263, CMRC264, CMRC265, CMRC266, CMRC267, CMRC268, CMRC269, CMRC270, CMRC271, CMRC272, CMRC273, CMRC274, CMRC276, CMRC277, CMRC278, CMRC279, CMRC280, CMRC281, CMRC282, CMRC284, CMRC286, CMRC287, CMRC281, CMRC311,		Apr-11
	CMRC209, CMRC210, CMRC211, CMRC212, CMRC213, CMRC214, CMRC215, CMRC216, CMRC217, CMRC219, CMRC221, CMRC222, CMRC223, CMRC224, CMRC225, CMRC226, CMRC227, CMRC228 CMRC227, CMRC230, CMRC231, CMRC231, CMRC233, CMRC234, CMRC235, CMRC236, CMRC237, CMRC238, CMRC239, CMRC240, CMRC241, CMRC242, CMRC245, CMRC246, CMRC247, CMRC248, CMRC249, CMRC251, CMRC252, CMRC253, CMRC255, CMRC256, CMRC257, CMRC258, CMRC259, CMRC260, CMRC261, CMRC262, CMRC263, CMRC264, CMRC265, CMRC265, CMRC266, CMRC267, CMRC267, CMRC268, CMRC270, CMRC271, CMRC272, CMRC273, CMRC274, CMRC276, CMRC277, CMRC278, CMRC279, CMRC280, CMRC281, CMRC281, CMRC284, CMRC284, CMRC284, CMRC286, CMRC286, CMRC286, CMRC269, CMRC269, CMRC260, CMRC260, CMRC276, CMRC276, CMRC276, CMRC277, CMRC276, CMRC276, CMRC276, CMRC276, CMRC276, CMRC276, CMRC281, CMRC281, CMRC281, CMRC281, CMRC281, CMRC281, CMRC281, CMRC286, CMRC286, CMRC286, CMRC296,		Apr-11

CM/MM FML holes ASX announcements:

Drill Hole Number	ASX Release Title	ASX Release Date
CMDD348, CMRC341, CMRC342	Focus Confirms Strong Results from Coolgardie and Laverton Exploration Campaigns	30-Jul-14

Criteria	Commentary

CM/MM historic Collar details of holes not previously externally reported:

		MGA 94	Zone 51				
Hole ID	Easting	Northing	RL	Azimuth	Dip	Depth (m)	Comp
SL10	440357.68	6829668.2	478.33	277	-60	93	Hillmin
SL14	440317.25	6829956.1	474.58	106	-56	50	Hillmin
SL5	440404.13	6829942.9	478.08	287	-60	63	Hillmin
SL6	440371.95	6829937.7	477.86	280	-60	69	Hillmin
SL7	440365.44	6829870.9	477.98	288	-60	33	Hillmin
SL8	440351.28	6829874.9	477.6	288	-60	63	Hillmin
CM002	440311.56	6829547.3	471.33	256.7	-60	42	SOG
CM003	440307.18	6829566.8	472.4	256.7	-60	30	SOG
CM004	440311.3	6829582.9	469.62	256.7	-60	40	SOG
CM005	440308.45	6829590.4	469.52	256.7	-60	50	SOG
CM007	440318.17	6829613.8	467.79	256.7	-60	42	SOG
CM008	440322.5	6829625.4	467	256.7	-60	54	SOG
CM010	440407.48	6829541.8	469.99	285.7	-60	50	SOG
CM012	440329.14	6829616.2	467	256.7	-60	60	SOG
CM013	440334.25	6829627.5	467	256.7	-60	60	SOG
CM015	440404.77	6829532.4	469.66	285.7	-60	46	SOG
CM016	440401.59	6829515.9	468.58	285.7	-60	46	SOG
CM017	440309.71	6829612.3	468.24	256.7	-60	54	SOG
CM018	440310.7	6829623	467	256.7	-60	54	SOG
CM019	440390.54	6829504.2	468.79	285.7	-60	50	SOG
CM020	440321.34	6829583.7	469	256.7	-60	60	SOG
CM021	440306.41	6829583.4	470.21	256.7	-60	36	SOG
CM022	440318.56	6829593.3	469.19	261.7	-60	54	SOG
CM023	440301.3	6829589.1	470.27	256.7	-60	30	SOG
CM024	440308.1	6829566.4	472.27	286.7	-60	60	SOG
CM025	440307.14	6829548.1	471.59	256.7	-60	34	SOG
CM026	440319.25	6829546.6	471	256.7	-60	72	SOG
CM027	440312.91	6829597.5	469.64	286.7	-60	39	SOG
CM028	440324	6829593.8	468.52	291.7	-60	72	SOG
CM029	440322.1	6829627.9	467	286.7	-60	40	SOG
CM030	440311.34	6829626.7	467	286.7	-60	75	SOG
CM032	440309.49	6829547.7	471.48	256.7	-60	39	SOG
CM017	440309.71	6829612.3	468.24	256.7	-60	54	SOG
CM018	440310.7	6829623	467	256.7	-60	54	SOG
CM019	440390.54	6829504.2	468.79	285.7	-60	50	SOG
CM020	440321.34	6829583.7	469	256.7	-60	60	SOG
CM021	440306.41	6829583.4	470.21	256.7	-60	36	SOG
CM022	440318.56	6829593.3	469.19	261.7	-60	54	SOG
CM023	440301.3	6829589.1	470.27	256.7	-60	30	SOG
CM024	440308.1	6829566.4	472.27	286.7	-60	60	SOG
CM025	440307.14	6829548.1	471.59	256.7	-60	34	SOG
CM026	440319.25	6829546.6	471.33	256.7	-60	72	SOG
CM027	440312.91	6829597.5	469.64	286.7	-60	39	SOG
CM028	440312.91			291.7	-60	72	SOG
	440322.1	6829593.8	468.52				SOG
CM029	440322.1 440311.34	6829627.9	467	286.7	-60	40	
CM030 CM032	440311.34	6829626.7 6829547.7	467	286.7	-60	75	SOG

Grade Control holes not previously reported have been reduced to those still remaining in unmined areas or beneath current base of pit.

riteria	Commentary						
	Craigiemore:						
	Hole ID			MGA 94 Zone 51			
		Easting	Northin	•	Azimuth	Dip	Depth (m)
	CR410002	440344.34	6829679		285	60	36
	CR410003 CR410004	440348.93 440342.37	6829678 6829690		285 285	60 60	36 36
	CR410004 CR410006	440356.26	6829696		285	60	42
	CR410008	440339.92	6829711		285	50	36
	CR410009	440356.08	6829706		285	60	42
	CR410011	440345.71	6829720	1.1 409.16	285	50	40
	CR410012	440354.46	6829717		285	60	42
	CR410014	440347.43	6829730		285	60	42
	CR410021	440344.45	6829749		285	65	40
	CR410023 CR410026	440312.52 440314.5	6829771 6829767		285 105	60 60	28 36
	CR410029	440313.16	6829781		105	65	36
	CR410029 CR410032	440317.92	682979		105	80	33
	CR410035	440355.27	682978		285	60	24
	CR410040	440354.46	6829790		285	60	36
	CR410042	440317.57	6829810	1.8 410.01	285	85	33
	CR410045	440350.97	6829802		105	55	36
	CR410048	440318.87	6829822		105	85	33
	CR410055	440321.73	6829841		0	90	33
	CR410056	440324.37	6829841		0	50	24
	CR410057 CR420005	440350.41 440302.55	682983 6829660		285 105	60 60	24 24
	CR420005 CR420016	440302.55	6829600		105	60	12
	CR420016 CR420017	440300.76	6829598		105	60	12
	CR425005	440361.47	6829685		285	55	54
	CR425006	440360.21	6829695		285	60	54
	CR425008	440306.18	6829711	.2 425.06	105	60	54
	CR425010	440361.07	682970	6 424.95	285	60	54
	CR425012	440307.06	6829721		105	60	54
	CR425013	440304.97	6829732	-	105	60	54
	CR425017	440307.57	6829741		105	60	54
	CR425018	440305.71	6829752		105	55	54
	CR425027	440375.36 440378.94	6829764		285	60	54
	CR425028 CR425031	440376.94	6829773 6829783		285 285	60 60	24 54
	CR425031		6829794	-			
	CR425035	440377.76 440381.39	6829804		285 285	60 60	24 24
	CR425036	440376.16	6829815		285	50	55
	CR425037	440374.52	6829827		285	60	24
	CR425038	440374.42	6829837		285	60	24
	CR425045	440366.74	6829870		285	60	24
	CR425048	440333.05	6829900		285	60	24
	CR425051	440328.76	6829901		285	60	24
	CR470001	440378.67	6829463		285	60	15
	CR470002	440383.34	6829461		285	60	25
	CR470004	440380.57	6829472	2.1 469.29	285	60	20
	CR470005	440384.8	6829470	0.8 468.61	285	60	30
	CR470006	440385.61	682948		285	60	22
	CR470009	440402.57	6829486		285	60	40
	CR470012	440401.85	6829497		285	60	40
	CR470026	440318.92	6829551		285	60	54
	CR470090	440390.43	6829947		285	60	38
	CR470093	440375.86	6829940		285	60	15
	CR470094	440382.08	6829928		285	60	34
	CR470104	440427.7	6829812		285	60	46
	CR470106	440419.59	6829803		285	60	39
	CR470110	440420.32	6829782		285	60	48
	CR470113	440423.46	6829771		285	60	48
	CR470116	440423.83	6829761		285	60	47
	CR470121	440426.9	6829749	0.8 467.86	285	60	47
	Mary Mac:						
	Hole ID			MGA 94 Zone 51			Depth (m)
		Easting	Northing	RL	Azimuth	Dip	
	MMHFPR14	440753.26	6831356.5	494.86	285	87	65
	MMHFPR19	440767.15	6831353.6	494.27	285	50	54
	MMHGC0028	440648.2	6831204.3	487.32	286	60	54

Criteria	Commonton						
riteria	Commentary						
	MMHGC0122	440769.7	6831377.9	497.31	286	60	50
	MMHGC0123	440774.52	6831376.5	495.56	286	60	40
	MMHGC0124	440779.33	6831375.1	495	286	60	40
	MMHGC0129	440766.4	6831347.6	492.98	286	60	40
	MMHGC0130	440771.21	6831346.3	492	286	60	40
	MMHGC0131	440776.03	6831344.9	491.5	286	60	40
					286		
	MMHGC0134	440764.29	6831358.6	495.05		60	40
	MMHGC0135	440759.48	6831359.9	496.12	286	60	41
	MMHGC0140	440758.88	6831339.3	493.1	286	60	50
	MMHGC0157	440744.9	6831405.3	507.44	234	60	35
	MMHGC0159	440742.71	6831400.7	507.28	236	60	35
	MMHGC0218	440745.08	6831384.8	497.78	287	60	18
	MMHGC0219	440745.08	6831384.8	497.78	0	90	18
	MMHGC0221	440773.52	6831384.2	495	0	90	30
	MMHGC0222	440773.52	6831384.2	495	105	63	30
	MMHGC0224	440730.13	6831368.3	500	0	90	48
	MMHGC0225	440728.13	6831358.5	500.16	285	60	54
	MMHGC0228	440715.86	6831351.6	500	285	70	48
	MMHGC0259	440713.19	6831248.7	487.72	285	60	42
	MMHGC0260	440721.12	6831246.3	486.49	285	60	36
	MMHGC0283	440647.18	6831187.8	470.02	285	60	30
	MMHGC0284	440639.6	6831179.6	475	285	80	36
	MMHGC45001	440726.03	6831318.3	449.79	285	60	30
	MMHGC45002	440696.84	6831326.3	450.09	285	60	12
	MMHGC45005	440699.18	6831303.1	450.13	285	60	24
	MMHGC45008	440676.35	6831310.9	450.54	285	60	12
	MMHGC45010	440698.27	6831284.4	450.29	285	60	30
	MMHGC45011	440690.63	6831285.8	450.31	285	60	24
	MMHGC45012	440676.09	6831287.2	450.03	285	60	18
	MMHGC45015	440689.93	6831276.3	450	285	60	30
	MMHGC45016	440683.17	6831278.1	450	285	60	24
	MMHGC45017	440675.36	6831281.1	450.03	285	60	18
	MMHGC45019	440658.17	6831284.9	450	285	60	18
	MMHGC45020	440678.73	6831269.3	450.12	285	60	30
	MMHGC45021	440673.73	6831270.2	450.13	285	60	24
	MMHGC45022	440666.88	6831272.1	450.02	285	60	24
	MMHGC45023	440660.55	6831273.9	449.96	285	60	24
	MMHGC45024	440653.36	6831275.8	450.05	285	60	24
	MMHGC45025	440646.68	6831277.5	449.92	285	60	24
	MMHGC45032	440650.41	6831255.5	450.06	285	60	26
	MMHGC45035	440628.05	6831261.4	449.96	285	60	15
					285	60	24
	MMHGC45037	440639.23	6831248.6	449.84			
	MMHGC45038	440632.02	6831249.1	449.86	285	60	24
	MMHGC45040	440650.69	6831235.2	450.01	285	60	30
	MMHGC45041	440640.01	6831238	449.86	285	60	24
	MMHGC45042	440629.41	6831240.7	449.81	285	60	24
	MMHGC45044	440636.33	6831228.7	449.95	285	60	24
	MMHGC45046	440649.33	6831214.2	450.28	0	90	15
	MMHGC45047	440641.74	6831216.4	450.12	285	60	18
	MMHGC45048	440635	6831218.5	450.02	285	60	18
	MMHGC45051	440633.85		449.89	285	80	30
			6831213.6				
	MMHGC45054	440618.44	6831208.5	450.13	285	80	30
	MMHGC45056	440622	6831191	452.52	285	60	30
	MMHGC45058	440618.22	6831182.7	452.66	285	60	30
	MMHGC45059	440602.46	6831186.5	452.52	285	60	30
	MMHGC45060	440612.99	6831174	452.52	285	80	30
	MMHGC465009	440650.69	6831222	464.64	285	60	45
	MMHGC465013	440632.36	6831228.8	465	285	60	36
	MMHGC465014	440657.86	6831221.4	464.59	285	60	53
	MMHGC465016	440647.93	6831235.3	464.8	285	60	46
	MMHGC465026	440621.33	6831263.5	464.68	0	90	30
	MMHGC465057	440681.11	6831329.8	464.95	105	85	24
	MMHGC470005	440741.69	6831333.5	469.63	285	70	30
	MMHGC470008	440738.19	6831324.4	469.46	285	60	36
	MMHGC470009	440730.55	6831326.4	469.65	285	60	36
	MMHGC470010	440722.93	6831328.6	470.1	285	60	36
	MMHGC470015	440736.19	6831304	469.61	285	60	36
	MMHGC470017	440728.89	6831295.8	469.64	285	60	45
	MMHGC470018a	440712.1	6831300	469.78	285	60	48
	MMHGC470022	440720.1	6831277.9	469.88	285	60	36
	MMHGC470034	440706.95	6831260.5	469.93	285	60	36
	MMHGC470039	440687.67	6831245.1	469.95	285	60	24
	MMHSPR02	440632.01	6831195.8	469.92	285	50	36
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Criteria	Commentary						
Officia	MMHSPR08	440624	6831176	470	285	50	30
	MMHSPR09	440594	6831175	470	105	70	36
	MMSGC001	440474.35	6830411.8	466.39	286	60	45
	MMSGC002	440467.53	6830413.6	466.42	286	60	45
	MMSGC003	440460.82	6830415.5	466.62	286	60	20
	MMSGC004	440454.04	6830417.4	466.81	286	60	10
	MMSGC005	440477.08	6830421.4	466.13	286	60	45
	MMSGC006	440470.29	6830423.3	466.08	286	60	45
	MMSGC007 MMSGC009	440456.83 440479.66	6830427 6830430.9	466.27 465.65	286 286	60 60	20 45
	MMSGC010	440473.04	6830432.9	465.85	286	60	45
	MMSGC011	440466.24	6830434.8	465.79	286	60	30
	MMSGC012	440459.49	6830436.6	465.74	286	60	20
	MMSGC014	440489.15	6830438.6	466.23	286	60	45
	MMSGC015	440482.42	6830440.5	466.42	286	60	45
	MMSGC016	440475.35	6830442.5	466.08	286	60	45
	MMSGC017	440468.51	6830444.9	466.03	286	60	20
	MMSGC019	440455.46	6830448.2	465.69	286	60	10
	MMSGC020	440485.13	6830450.3	466.27	286	60	45
	MMSGC021	440478.44	6830451.3	469.08	286	60	49
	MMSGC022	440471.38	6830454.1	469.37	286	60	24
	MMSGC023	440465.07	6830455.9	469.19	286	60	14
	MMSGC024	440457.41	6830457.8	465.9	286	60	10
	MMSGC025	440487.95	6830459.9	466.2	286	60	45
	MMSGC026	440481.26	6830461.6	466.06	286	60	45
	MMSGC027	440474.16	6830463.7	466.12	286	60	45
	MMSGC028	440467.58	6830465.5	466.04	286	60	20
	MMSGC029 MMSGC030	440460.78 440490.51	6830467.1 6830469.5	465.9 466.41	286 286	60 60	10 45
	MMSGC030	440490.31	6830475.2	466.3	286	60	45 45
	MMSGC032	440463.54	6830477.1	466.2	286	60	45
	MMSGC033	440450.15	6830480.8	466.12	286	60	20
	MMSGC034	440443.39	6830482.8	466.02	286	60	10
	MMSGC035	440486.52	6830481	466.56	286	60	45
	MMSGC036	440479.7	6830482.9	466.54	286	60	45
	MMSGC037	440473.01	6830484.8	466.5	286	60	45
	MMSGC038	440466.26	6830486.8	466.38	286	60	45
	MMSGC039	440459.51	6830488.6	466.36	286	60	45
	MMSGC040	440452.86	6830490.5	466.39	286	60	20
	MMSGC041	440446.13	6830492.4	466.35	286	60	10
	MMSGC042	440489.16	6830490.6	466.85	286	60	45
	MMSGC043	440475.71	6830494.4	466.61	286	60	45
	MMSGC044 MMSGC045	440469 440455.55	6830496.3 6830500	466.63 466.64	286 286	60 60	45 20
	MMSGC046	440448.79	6830502	466.57	286	60	10
	MMSGC047	440491.93	6830500.3	467.13	286	60	45
	MMSGC048	440485.16	6830502.2	466.93	286	60	45
	MMSGC049	440478.46	6830504.1	466.88	286	60	45
	MMSGC050	440471.67	6830506	466.89	286	60	45
	MMSGC0500	440540.91	6830855.5	475.7	285	60	24
	MMSGC0501	440548.13	6830854.3	475.06	285	60	24
	MMSGC0502	440554.24	6830851.6	474.79	285	60	24
	MMSGC051	440464.92	6830507.9	466.81	286	60	45
	MMSGC0512	440585.12	6830962.3	474.85	285	60	24
	MMSGC0514	440570.11	6830997.7	475.45	285	60	24
	MMSGC0515 MMSGC0516	440577.31 440585.47	6830995.8 6830993.5	475.51 475.5	285 285	60	24 24
	MMSGC0517	440593.14	6830991.3	475.37	285	60	24
	MMSGC0517	440600.94	6830989.2	475.34	285	60	24
	MMSGC0519	440572.82	6831007.6	475.78	285	60	24
	MMSGC052	440458.31	6830509.8	466.75	286	60	20
	MMSGC0520	440580.42	6831005.1	475.7	285	60	24
	MMSGC0521	440588.24	6831003	475.7	285	60	24
	MMSGC0522	440595.97	6831000.9	475.52	285	60	24
	MMSGC0523	440603.6	6830998.8	475.49	285	60	24
	MMSGC0524	440583.12	6831014.8	475.91	285	60	24
	MMSGC0525	440585.78	6831024.5	476.16	285	60	24
	MMSGC0526	440593.66	6831022.2	476.17	285	60	24
	MMSGC0527	440601.27	6831020.1	476.15	285	60	24

Criteria	Commentary						
OTILOTIA	MMSGC0528	440608.78	6831017.8	476.01	285	60	24
	MMSGC0529	440616.64	6831015.7	475.83	285	60	24
	MMSGC053	440451.49	6830511.6	466.68	286	60	10
	MMSGC0530	440588.69	6831034.1	476.41	285	60	24
	MMSGC0531	440602.02	6831030.5	476.33	285	60	24
	MMSGC0532	440595.9	6831052.8	477.1	286	60	24
	MMSGC0533	440603.41	6831050.7	477.06	286	60	24
	MMSGC0534	440611.22	6831048.5	476.97	286	60	24
	MMSGC0535	440619.06	6831046.3	476.94	286	60	24
	MMSGC054	440494.64	6830510	467.33	286	60	45
	MMSGC055	440481.11	6830513.7	467.28	286	60	45
	MMSGC056	440474.47	6830515.5	467.18	286	60	45
	MMSGC057	440461.01	6830519.4	467.14	286	60	30
	MMSGC058	440454.16	6830521.2	467.01	286	60	10
	MMSGC059	440497.4	6830519.5	467.9	286	60	30
	MMSGC060	440490.65	6830521.4	467.72	286	60	45
	MMSGC061	440483.84	6830523.4	467.69	286	60	45
	MMSGC062	440477.08	6830525.2	467.6	286	60	45
	MMSGC063	440470.37	6830527.1	467.76	286	60	45 45
	MMSGC064 MMSGC065	440463.63 440456.89	6830529 6830530.9	467.64 467.57	286 286	60 60	20
	MMSGC066	440456.89	6830530.9	467.41	286	60	10
	MMSGC067	440493.25	6830531.1	468.23	286	60	45
	MMSGC068	440486.49	6830533	468.2	286	60	45
	MMSGC069	440479.79	6830534.9	468.15	286	60	45
	MMSGC070	440473.05	6830536.8	468.05	286	60	45
	MMSGC071	440466.31	6830538.6	468.08	286	60	45
	MMSGC072	440459.68	6830540.5	468.05	286	60	45
	MMSGC073	440452.77	6830542.5	467.96	286	60	20
	MMSGC074	440446.1	6830544.3	467.79	286	60	10
	MMSGC075	440482.44	6830544.6	468.64	286	60	45
	MMSGC076	440475.73	6830546.3	468.71	286	60	35
	MMSGC077	440462.24	6830550.1	468.65	286	60	45
	MMSGC078	440455.6	6830552.1	468.68	286	60	20
	MMSGC079	440498.68	6830550.3	469.07	286	60	45
	MMSGC080	440492.04	6830552.2	469.17	286	60	45
	MMSGC081	440485.26	6830554.1	469.18	286	60	25
	MMSGC084	440465.06	6830559.8	469.24	286	60	25
	MMSGC085	440458.27	6830561.7	469.17	286	60	25
	MMSGC091	440474.42	6830567.5	469.76	286	60	45
	MMSGC092	440460.98	6830571.3	469.71	286	60	45
	MMSGC094	440544.55	6830558.2	468.63	286	60	45
	MMSGC098 MMSGC099	440490.63 440483.92	6830573.3	470.17 470.21	286	60 60	45 45
	MMSGC100		6830575.2 6830577.2	470.27	286 286	60	45
	MMSGC100	440477.16 440470.38	6830579	470.27	286	60	45 45
	MMSGC101	440470.38 440547.25	6830567.8	468.37	286	60	45 45
	MMSGC106	440540.51	6830569.7	468.81	286	60	45
	MMSGC129	440552.66	6830587.1	468.8	286	60	45
	MMSGC331	440558.29	6830865.9	474.58	286	60	25
	MMSGC332	440551.53	6830867.8	474.97	286	60	25
	MMSGC333	440544.76	6830869.6	475.26	286	60	20
	MMSGC334	440538.09	6830871.7	475.76	286	60	10
	MMSGC335	440561.02	6830875.5	474.78	286	60	25
	MMSGC336	440554.32	6830877.5	474.96	286	60	25
	MMSGC337	440547.46	6830879.4	475.32	286	60	20
	MMSGC338	440540.81	6830881.2	477.08	286	60	25
	MMSGC340	440557	6830887.1	474.94	286	60	25
	MMSGC341	440543.52	6830890.9	475.67	286	60	25
	MMSGC342	440536.81	6830892.9	476.19	286	60	20
	MMSGC343	440530.09	6830894.7	477.52	286	60	10
	MMSGC344	440566.41	6830894.9	474.58	286	60	25
	MMSGC345	440559.53	6830896.6	474.72	286	60	45
	MMSGC346	440552.85	6830898.6	475.04	286	60	45
	MMSGC347	440546.13	6830900.5	475.52	286	60	45
	MMSGC348	440539.45	6830902.4	475.83	286	60	45
	MMSGC427	440478.88	6830410.4	466.34	286	60	50
	MMSGC428	440483.55	6830419.6	466.13	286	60	55

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	MMSGC429	440495.74	6830436.8	466.31	286	60	70	
	MMSGC430	440492.58	6830448.2	466.21	286	60	65	
	MMSGC431	440495.37	6830457.9	466.04	286	60	65	
	MMSGC432	440497.95	6830477.8	466.49	286	60	65	
	MMSGC433	440507.33	6830485.6	467.05	286	60	75	
	MMSGC434	440506.16	6830496.4	467.09	286	60	70	
	MMSGC435	440499.72	6830498.3	467.06	286	60	55	
	MMSGC436	440501.45	6830508	467.76	286	60	65	
	MMSGC437	440506.95	6830516.8	467.54	286	60	70	
	MMSGC438	440505	6830548.3	468.86	286	60	65	
	MMSGC439	440468.84	6830558.7	469.1	286	60	40	
	MMSGC440	440504.99	6830558.9	469.33	286	60	70	
	MMSGC441	440498.39	6830560.8	469.43	286	60	50	
	MMSGC442	440490.43	6830562.8	469.55	286	60	60	
	MMSGC443	440490.98	6830386.2	467.14	286	60	60	
	MMSGC444	440497.75	6830384.4	467.15	286	60	70	
	MMSGC446	440556.6	6830575.6	468.61	286	60	70	
	MMSGC447	440482.19	6830399.1	466.73	286	60	55	
	MMSGC448	440488.86	6830397.3	466.76	286	60	65	
	MMSGC449	440495.64	6830395.4	466.77	286	60	70	
	MMSGC450	440484.98	6830408.8	466.53	286	60	65	
	MMSGC451	440490.52	6830417.6	466.12	286	60	70	
	MMSGC452	440497.09	6830426.1	466.07	286	60	70	
	MMSGC453	440504.26	6830434.9	466.3	286	60	70	
	MMSGC454	440502.35	6830455.8	466.83	286	60	65	
	MMSGC458	440553.99	6830565.9	468.44	286	60	60	
	MMSGC459	440560.73	6830564	468	286	60	70	
	MMSGC461	440585.19	6830858.4	473.6	286	60	35	
	MMSGC462	440571.75	6830862.1	473.96	286	60	50	
	MMSGC463	440578.44	6830860.3	473.69	286	60	60	
	MMSGC464	440565.11	6830864.1	474.33	286	60	70	
	MMSGC465	440583.18	6830869.3	473.59	286	60	70	
	MMSGC466	440570.37	6830883.3	474.31	286	60	65	
	MMSGC467	440573.19	6830893	474.42	286	60	10	
	MMSGC468	440532.82	6830904.4	476.4	286	60	20	
	MMSGC470	440530.63	6830915.3	476.08	286	60	15	
	MMSGC471	440537.32	6830913.4	475.76	286	60	25 55	
	MMSGC472	440554.67	6830908.6	475.1 475.2	286 286	60 60	20	
	MMSGC473 MMSGC474	440538.21	6830923.6				40	
	MMSGC474	440544.89 440551.65	6830921.6 6830919.8	475.2 475.06	286 286	60 60	45	
	MMSGC476	440566.07	6830915.8	474.78	286	60	60	
	MMSGC477	440572.73	6830913.9	474.62	286	60	70	
	MMSGC479	440540.74	6830943.6	475.25	286	60	20	
	MMSGC480	440547.33	6830941.7	475.19	286	60	30	
	MMSGC481	440554.13	6830939.9	475.15	286	60	40	
	MMSGC482	440560.84	6830937.9	474.96	286	60	55	
	MMSGC483	440575.23	6830933.8	474.75	286	60	70	
	MMSGC484	440547.28	6830952.2	475.29	286	60	35	
	MMSGC485	440558.78	6830948.9	475.16	286	60	40	
	MMSGC486	440567.47	6830946.5	474.98	286	60	50	
	MMSGC487	440574.15	6830944.6	474.82	286	60	55	
	MMSGC488	440551.85	6830961.2	475.45	286	60	15	
	MMSGC489	440580.68	6830953.1	474.78	286	60	70	
	MMSGC490	440573.62	6830975.9	475.28	286	60	40	
	MMSGC491	440589.98	6830971.3	475.03	286	60	70	
	MMSGC492	440596.79	6830969.4	474.96	286	60	50	
	MMSGC493	440603.51	6830967.6	474.9	286	60	50	
	MMSGC494	440610.14	6830965.7	474.76	286	60	40	
	MMSGC495	440582.07	6830984	475.33	286	60	60	
	MMSGC496	440600.39	6830978.7	475.18	286	60	70	
	MMSGC497	440591.22	6831012.5	475.9	286	60	50	
	MMSGC498	440596.91	6831011	475.89	286	60	70	
	MMSGC499	440613.37	6831006.2	475.74	286	60	70	
Data				it-off with a minimum			-	0 2m for dia
aggregation methods	holes, composite		ou at a v.og/t Au ot	a on mara millimulli	. oporting what	01 11111	ioi ito iioios aiiu	J.ZIII IOI UIA
Relationship	Holes were drille	d orthogonal to min	eralisation as much	as possible, however	the exact rela	tionshin h	netween intercent	width and tru
wanonsinp		ated exactly in all ca		ao possibie, nowevel	uio oxaotitid	riorioriih r	,οιννοοπ πποισορι	muui ailu lit

Criteria	Commentary
between mineralisation widths and intercept lengths	
Diagrams	Refer to Figures and Tables in body of the release.
Balanced reporting	 All drill assay results used in this estimation are published in previous news releases. Historic drill hole results available on WAMEX.
Other substantive exploration data	There is no other material exploration data to report at this time.
Further work	The company is designing drill programs to follow up on the results of these re-models.

Criteria	on 1, and where relevant in section 2, also apply to this section) Commentary
Database integrity	Data was geologically logged electronically; collar and downhole surveys were also received electronically as was the laboratory analysis
,	results. These electronic files were loaded into an acQuire database by the company in-house Database Administrator. Data was routinely
	extracted to Microsoft Access during the drilling program for validation by the geologist in charge of the project.
	FML's database is a Microsoft SQL Server database (acQuire), which is case sensitive, relational and normalised to the Third Normal
	Form. Because of normalisation, the following data integrity categories exist:
	Entity Integrity: no duplicate rows in a table, eliminated redundancy and chance of error.
	 Domain Integrity: Enforces valid entries for a given column by restricting the type, the format or a range of values.
	 Referential Integrity: Rows cannot be deleted which are used by other records.
	 User-Defined Integrity: business rules enforced by acQuire and validation codes set up by FML.
	Additionally, in-house validation scripts are routinely run in acQuire on FML's database and they include the following checks:
	Missing collar information
	Missing logging, sampling, downhole survey data and hole diameter
	Overlapping intervals in geological logging, sampling, down hole surveys
	Checks for character data in numeric fields
	· Data extracted from the database were validated visually in GEOVIA Surpac software and ARANZ Geo Leapfrog software. Also, when
	loading the data any errors regarding missing values and overlaps are highlighted.
	Historic data has been validated against WAMEX reports where possible.
Site visits	• Alex Aaltonen, the Competent Person for Sections 1, 2 and 3 of Table 1 is FML's General Manager - Exploration and conducts regular
	site visits.
Geological	All available drill hole and historic mining data was used to guide the geological interpretation of the mineralisation.
interpretation	• The mineralised geological interpretation was completed using Seequent Leapfrog software on a section-by-section basis. All available
	drill hole and pit mapping data was used with an approximate 0.5g/t Au value cut-off to guide the interpretation.
	· Mineralisation along the Chatterbox Trend has previously been modelled as discrete and discontinuous individual deposits. Not with a
	view of the structural controls over the whole trend. The aim of this remodel was to look at the shear zone as one large deposit that is
	later sub-divided into the historically defined individual deposits.
	A total of 56 individual lodes have been modelled along the Chatterbox trend.
	• 10 closely spaced lodes were modelled within the Rumor deposit that strike NNE curving towards the NE to the north, dipping 60° to the
	East. The three most southern lodes have a flatter 45° easterly dip.
	• Apollo is the longest of the three deposits and had 32 individual lodes modelled along its strike. The mineralisation has been interpreted
	from the end of what is considered the Rumor footprint striking NE before
	• curving to a Northerly trend at the northern extents. Two core pervasive lodes have been modelled dipping 65° to the east, with numerous
	closely spaced smaller lodes in the footwall and hanging wall dipping from 30° to 70° to the east along the entire Apollo strike. To the
	west of the main Apollo trend two NNE trending lodes dipping 45° to the east have been modelled.
	• Slightly offset to the East from Apollo in the north lies the Eclipse (Garden Well) deposit that has one N-S, 60° dipping lode modelled
	through the historic pit area and to the east, two N-NNW trending closely spaced lodes dipping 55° to the east.
	• Innuendo to the north has ten lodes modelled, in the south five stacked lodes trend NNE and dipping 60° to the east. The mineralisation
	then swings to the NNW where four stacked lodes dip ~ 55° to the NE. Approx 150m to the North of this a further two NNW trending 60°
	dipping lodes have been interpreted.
	Along the West Laverton Trend, a total of 13 individual lodes were modelled.
	• At Rega two lodes were modelled, one main lode within the existing pit and a second lode ~ 150m to the west. Both strike NNW with a
	gentle ~25° dip to the ESE and gently plunging to the east. Within the main in- pit lode, two higher grade internal "core" lodes were
	modelled. These HG core lodes have hard boundaries between themselves and the surrounding main lode. There is no defined northing
	or cut-off for the Rega / West Laverton mineralisation extents and the two deposits interlace each other at the southern edge of the Rega
	pit wall.
	• Six lodes were modelled at West Laverton, two hanging wall and two footwall lodes adjacent to two splay lodes. All lodes are closely
	spaced and sub-parallel to each other, striking NNE, gently dipping ~ 25° to the ESE and gently plunging to the east, similarly to Rega.
	Bulldog sits some 900m along strike to the south of West Laverton and consists of five lodes. All are closely spaced, sub-parallel with
	similar orientations to West Laverton and Rega.
	 Bulldog sits some 900m along strike to the south of West Laverton and consists of five lodes. All are closely spaced, sub- parallel with
	similar orientations to West Laverton and Rega.
	Gladiator Trend had a total of 54 individual lodes have been modelled.

Criteria Commentary Gladiator West is closely associated with Gladiator Underground footwall, 21 closely spaced lodes were modelled within the Gladiator West deposit that strike NNE, dipping 45° to the East. The three most southern lodes have a flatter 27° easterly dip. Murrays deposit was modelled in 4 lodes striking NNE, dipping 51° to the East. Cousin Murray was modelled in 7 lodes striking NW, dipping 75° to the SW. Gladiator south was modelled in 19 lodes overall striking North, dipping 70° to the East with northeastern lodes striking NNW. Central BIF was modelled in 1 lode striking NNW dipping 53° to the East. NE BIF was modelled in 1 lode striking north, dipping 79° to the East. Mineralisation at Craigiemore and Mary Mac is commonly associated with quartz veining and disseminated pyrite within the silica altered chert horizons of the BIF. It can also be found in the mafic host rocks near the BIF contact. Understanding the relationship between gold mineralisation and the BIF was used to guide the interpretation. Lithological logging of BIF and mafic host rocks was consistent across drill holes and allowed for an accurate lithological model to be constructed. This model highlighted the contacts and thickened zones of BIF where gold mineralisation is prevalent. Geophysics was also used to guide the interpretation with the BIF highlighted by regional and project scale surveys. A site visit and pit analysis conf lodes were identified striking north-south with a moderate plunge to the north. Minor northwest striking sub vertical splays were also identified. Increased thrust faulting and drag folding enhances the structural complexity at Mary Mac Hill. Two orientations were identified, north south striking BIF hosted lodes that dip steeply to the east and more moderate east dipping quartz and shear hosted lodes. A total of 23 individual lodes were modelled at Craigiemore. Two sets of closely spaced north trending sub-vertically dipping lodes that gently plunge to the north. In the "gap" between Craigiemore and Mary Mac, two NNW trending, sub-vertical lodes have been interpreted. A remnant ROM Pad stockpile exists and has been modelled to the immediate East of the current open pit berm. The triangular shaped stockpile extends 275m at its longest, 190m at its widest and averages 5m thick. Covering the entire CM-MM-GP resource area a thin continuous Supergene layer of mineralised enrichment has been modelled as a surface. However, lack of RC/DD drilling away from the pits has affected its estimation. A total of 59 individual lodes have been modelled along the MM/GP trend. Mary Mac South consists of 12 stacked, NNW trending, sub-vertically dipping lodes that extend over 470m from near surface to approx. 210m below ground. Mary Mac North consists of 5 stacked NW trending sub-vertical lodes intersected by 5 flatter dipping lodes modelled over 270m from near surface to 180m below. Mary Mac Hill consists of 11 stacked flat dipping lodes intersected by 20 steep NNW trending lodes that has been modelled over 385m from near surface to 150m below surface. Golden Pinnacles consists of 6 lodes of variable orientations and limited strike that has been modelled from near surface to approx. 100m below.irmed the vein orientation. At Craigiemore and Mary Mac numerous tightly spaced sub vertical Dimensions The entire Chatterbox Trend was remodelled as one project in Leapfrog, over a Estimation and 5.7 km NNE - North - NNW strike length. modelling Rumor has been interpreted over a 1.5km NE - NNE trending strike from near surface to 230m below ground, however the average depth techniques modelled is 120m. Widths vary from 1m to 17m. Apollo trends NNE over a 1.5km strike before swinging to the North for a further 900m. Lodes average 150m depth but have been interpreted to 330m at the deepest point of drill penetration. Lode widths vary from 1m to 20m wide. Eclipse is the smallest of the deposits – inferred over a 600m N to NNW strike from near surface to 150m below ground with 1m – 10m wide lodes. Innuendo extends over 1.4km, 300m striking to the NNE before swinging to the NNW for 1.1km. Mineralisation has been modelled from near surface to 230m below surface and vary from 1m 10 16m wide. Overall, the West Laverton Trend has been modelled over a 2.2km strike length. West Laverton/Rega has been modelled over 850m with a gap of 900m before Bulldog has been interpreted over 500m strike. The lodes are interpreted from near surface to approximately 150m below surface, limited by depth of drilling. Lodes have been interpreted from 1m to 25m thick. The Gladiator deposits were remodelled as one project in Leapfrog, over a 2 km NNE - North - NNW strike length. Gladiator West has been interpreted over 570m NNW trending strike from near surface to 300m below ground level, however the average depth approximates 170m. Width vary from 2.5m to 15m. Murrays has been interpreted over 950m NNE trending strike from near surface to 130m below ground level. Widths vary from 2.5m to Cousin Murray has been interpreted over 275m NNW trending strike from near surface to 167m below ground level. Widths vary from 1m to 8m. Gladiator South has been interpreted over 1080m NNW trending strike from near surface to 160m below ground level. Widths vary from Central BIF and NE BIF have been interpreted over 800m and 500m respectively North trending strike from near surface to 150m and 50m below ground level Width vary from 1m to 4m. The entire Craigiemore - Mary Mac - Golden Pinnacles Trend was remodelled as one project in Leapfrog, over a 2.8km N to NNW strike length. The lodes are interpreted from near surface to approximately 230m below surface, limited by depth of drilling. Lodes have been interpreted from 1m to 15m thick. Craigiemore has been modelled over 960m with a gap of 75m before a further two lodes have been interpreted over 185m strike. Mary Mac South through to Golden Pinnacles (MM/GP) extends over 1.6km NNW trending strike. All deposits used an Ordinary Kriging (OK) estimate run in Datamine software. Snowden Supervisor software was used for geostatistical analysis and variography of the individual lode composited samples.

Criteria Commentary

Chatterbox:

- Drill hole data was selected within mineralised lodes. Boundaries between lodes were considered hard boundaries and no data is shared between lodes. All drill hole data was composited to 1m downhole intervals 1m is the dominant raw sampling interval.
- Samples flagged as unreliable or wet were set to absent and ignored in the estimation process.
- A review of histograms, probability plots and mean/variance plots for the individual lodes revealed outlier sample values. A maximum topcut of 20ppm Au and an average of 7ppm Au was used for Apollo, only two lodes were top- capped at Eclipse – 25ppm and 8ppm Au, a max of 15ppm Au and average of 9ppm Au for Rumor and a max of 8ppm and average of 4ppm Au for Innuendo. Assays above the topcut are set to the top-cut value.
- Variography was modelled on data transformed to normal scores, the variogram models were back transformed to original units before exporting.
- Variography was performed on the individual lodes with larger sample numbers, a total of 26 variograms were modelled along the Chatterbox trend.
- · These models were shared with the other lodes of similar orientation and proximity.
- The back-transformed variogram models had moderate nugget effects (12% to 48% of total sill), with a range from 28m for the smaller lodes through to 170m for the largest lodes at Apollo.
- Estimation (via Ordinary Kriging) was into a non-rotated block model in MGA94 grid, with a parent block size about the average drill spacing within the deposit areas, Apollo and Eclipse was 5 mE x 10 mN x 5 mRL with the infill RC grade control holes; Rumor and Innuendo was 10 mE x 25 mN x 5 mRL. Sub-blocking was used to best fill the wireframes and inherit the grade of the parent block.
- The ellipsoid search parameters used the variogram ranges, with a minimum of 6 and maximum of 16-18 samples per block estimate was used. After the first pass for un-estimated blocks, the search distance was expanded by a factor of two and the minimum number of samples dropped to 4. A third pass was then run with an increased search distance by a factor of four and the same minimum number of samples. A few lodes across all deposits had blocks not fill after the third pass, this was in areas at the extents of strike or in outside the main strike in the FW and HW lodes. For a small number of missing blocks in a lode, the average of the surrounding blocks was used. This was below cut-off grade and the blocks assigned sub-inferred, unclassified. For large numbers of absent blocks, a 0.01ppm Au was assigned.
- Along the main strike, lodes had high numbers of blocks filling in the first pass, 99% in a few lodes at Apollo. Smaller lodes along FW or
 HW had lower first pass estimation ~ 50% of blocks had estimated. In the second pass an average of 46% of blocks estimated an average
 of 11% of blocks estimated in the third search pass.
- The estimate was validated by visually stepping through the estimated blocks and sample data in Datamine. Comparing the estimated block statistics with composited sample data and generate trend (Swath) plots to ensure the estimate was honouring the trends of the data. Also, a review of the output parameters from the estimation process like kriging variance, negative weights, search distances and sample numbers.
- Following a review of estimated lode grades vs composite lode grades, a "distance limited search" was applied to lodes that had high
 grades being spread into areas of fewer lower grade samples artificially influencing the grades of these blocks.
- The process is to apply a distance limit, 10m for Eclipse, 20m for Apollo and 25m for Rumor and Innuendo, to samples above a cut- off grade. Outside the 10m 25m search ellipse, assays above cut-off are removed from the estimation, resulting in blocks better honouring the low grades in areas of less drilling. Different grades were used for different deposits, Apollo = 10ppm Au, Rumor = 8ppm Au, Eclipse = 6ppm and Innuendo = 4.5ppm Au top-cut applied.

West Laverton:

- Drill hole data was selected within mineralised domains and then within the internal vein set lodes. Boundaries between veins and the
 surrounding domain were considered hard boundaries and no data is shared between lodes or between domains and lodes. All drill hole
 data was composited to 1m downhole intervals 1m is the dominant raw sampling interval.
- outlier sample values. A maximum top-cut of 15ppm Au and an average of 10ppm Au was used for the different lodes, with assays above
 the top-cut set to the top-cut value.
- Normal scores variography was performed on the individual lodes with larger sample numbers, in total 9 variograms were modelled and shared with the other lodes of similar orientation and proximity.
- The back-transformed variogram models had moderate to high nugget effects (26 to 36% of total sill), with a range from 20m to 100m for the lodes. In general, the ranges for the variograms were quite short, averaging 40m.
- Estimation (via Ordinary Kriging) was into a non-rotated block model in MGA94 grid, with a parent block size of 10 mE x 10 mN x 5 mRL
 – this is about the average drill spacing in the deposit. Sub-blocking was used to best fill the wireframes and inherit the grade of the parent block.
- The ellipsoid search parameters used the variogram ranges, with a minimum of 8 and maximum of 14 samples per block estimate was used. After the first pass 57% of blocks had estimated, primarily due to the short search range used. For un-estimated blocks after this first pass, the search distance was expanded by a factor of two and the minimum number of samples dropped to 4. In the second pass 30% of blocks estimated. A third pass was then run with an increased search distance by a factor of four and the same minimum number of samples, 13% of blocks estimated in the third search pass.
- The estimate was validated by the same process described above.
- Following the review, a "distance limited search" described in Chatterbox was applied to 5 lodes that had high grades being spread into areas of fewer lower grade samples artificially influencing the grades of these blocks. A distance limit of 20m was selected based on a visual review of sample spacing in affected lodes and a grade cut-off above 5ppm.

Gladiator

- Drill hole data was selected within mineralised lodes. Boundaries between lodes were considered hard boundaries and no data is shared between lodes. All drill hole data was composited to 1m downhole intervals – 1m is the dominant raw sampling interval.
- A review of histograms, probability plots and mean/variance plots for the individual lodes revealed outlier sample values. A maximum topcut of 12ppm Au was applied to a Gladiator lode, an average of 5ppm Au was used for Gladiator and Murrays with an average of 4ppm

As all Cousen Murray. Top- capping was lower at Gladalotz West with an avarage of 275ppm Au used. Assays above the top- cut are set to the Rep- cut are set to the Author of the Rep- cut and the Author of the Rep- cut are set to furray, one at Cousin Murray and two variograms monotive due incidents to high rugge effects (18% to 48% of total sill, with a range from 31 in for the enable rodge through to 100m for the largest lodes at Gladalotz. Estimation (via Circinary Kriging) was interest and a convolated both conde in MCAP4 and with a parent block size of 5 miz x 10 mix x 5 mix 1 mix 1 mix 1 mix 2 mix 1 mix 2	Criteria	Commentary
to the top cut value. • Variographs upperformed on the individual lodes with large sample numbers. The skewed data sets were bransformed to normal scores, the variogram models were back transformed to original units before exporting. Nine variograms were modelled at Cladation, three at Murrays, one at Cousin Murray and two variograms modelled at Cladation Vest. The variogram modes were shared with the other lodes of similar orientation and proximity. The back-transformed variogram models had moderate to high mugget effects (16% to 45% of total sill), with a range from 3 lim for the smaller lodes through 100 flor for the larges total explicit on the country of the country	Officeria	
Variography was performed on the individual locks with larger sample numbers. The skewed data sets were transformed to normal scores, the variogram models were but scharformed to original units before exporting. Nine variograms were modelled at Gladiator, three at Murray, one at Cousin Murray and two variograms models and Gladiator West. The variogram models were started with the other locks of similar orientation and poxinity. The back-harsofront variogram models and moderate but high ungel effects (16%) 46% of total sill, with a range from 3 fm for the smaller locks through to 100m for the largest bodes at Gladiator. Estimation (viol. Gridnary Kriging) was into a non-rotated block model in MCARP (qri), with a parent block size of 5 mE x 10 mN x 5 mRL—this is about the average and illigatory and the variogram ranges, with a minimum of 12 -18 samples per block estimate bode. The eligibad search parameters used the variogram ranges, with a minimum of 12 -18 samples per block estimate was used. The variable maximum amplier numbers used depended on size of this lock and dill specing. After the first pass of 50° of blocks had estimated. For un-estimated blocks after this first pass, the search distance was expanded by a factor of four and the same minimum number of samples, 50° of blocks estimated. After places was withen the with an increased search distance by a factor of four and the same minimum number of samples, 50° of blocks estimated in the third search pass. The estimate was validated by the same process described in the Chatebook estimate. Following a review of estimated lode grades vs. composite lode grades, the "distance limit of four was selected and a grade cut-off above 50 ppm. Craigemore/May Minimum of the same process described in the Chatebook estimate. Following a review of estimated lode grades vs. composite lode grades, the "distance limit of four was selected and a grade cut-off above 50 ppm. As was used for Craigemore and an average of 70° pm. 40° pm. 40° pm. 40° pm. 40° pm. 40° pm. 40		
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Craigiemore May Mac: Drill hole data was selected within mineralised lodes. Boundaries between lodes were considered hard boundaries and no data is shared between lodes. All drill hole data was composited to 1m downhole intervals – 1m is the dominant raw sampling interval. A review of histograms, probability plots and mean/variance plots for the individual lodes revealed outlier sample values. A maximum top-out of 25pm Au and an average of 25pm Au was used for Craigiemore and an everage of 7pm Amy Mac. Assays above the top-out are set to the top-out value. Normal Scores variography was performed on the individual lodes with larger sample numbers, 17 variograms were modelled at Craigiemore and 37 variograms at Mary Mac. These models were shared with the other lodes of similar orientation and proximity. The back-transformed variogram models had low to moderate rugget effects (6% to 35% of total still), with a range from 25m for the smaller lodes through to 127m for the largest lodes. Estimation (via Ordinary Kriging) was into a non-rotated block model in MGA94 grid, with a parent block size of 5 mE x 10 mN x 5 mRL—this is about the average drill spacing in the deposit. Sub-blocking was used to best fill the wireframes and inherit the grade of the parent block. The ellipsoid search parameters used the variogram ranges, with a minimum of 6 and maximum of 14 samples per block estimate was used. After the first pass 68% of blocks had estimated, for un-estimated blocks after this first pass, the search distance was expanded by a factor of two and the minimum number of samples dropped to 4. In the second pass 11% of blocks estimated in the third search pass. The estimate was validated by methods described above at Chatterbox. Following the review, a "distance limited search" was applied to 5 lodes that thad high grades being spread into areas of fewer lower grade samples afficially influencing the grades of these blocks. A distance limit of 10m and a grade out-off above 5ppm was used. Moisture The mineral res		had high grades being spread into areas of fewer lower grade samples artificially influencing the grades of these blocks. A distance limit
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i		recoveries over 94%. A mineralogical analysis was conducted in June 2010 through AMMTEC. A total of 4 samples from Craigiemore,
representing the main lode and SE lode's oxide layer and fresh rock were submitted and showed high gold liberation.		
Environmental • All deposits have been historically mined by either open pit or underground methods and existing ground disturbances including haul	Environmental	• All deposits have been historically mined by either open pit or underground methods and existing ground disturbances including haul

Criteria	Commentary
factors or	roads and waste dumps exist in the area. There are no unforeseen environmental considerations that would prevent open pit mining from
assumptions	re-commencing in the area.
Bulk density	 Density values used across the deposits comes from a mix of diamond core testing to rock sample test work conducted during the mining process, all test work using the water immersion SG test. Density values were assigned based on weathering profile and/or mineralisation or waste classification. Along the Chatterbox within the mineralised lodes an average t/m3 of 1.82 was used in oxide, 2.54 Transitional and 2.75 Fresh. In the waste an average t/m3 of 1.8 was used in oxide, 2.34 Transitional and 2.75 Fresh. At West Laverton density values were assigned based on weathering profile using the average results from 2008 test work. The water immersion SG test work was conducted on diamond core samples collected from two Crescent diamond holes. However, no fresh rock was encountered and an average SG for fresh basalt was applied. An average SG of 2.1t/m3 for oxidised, 2.24 t/m3 for transitional material and 2.70 t/m3 for Fresh rock were applied. Gladiator used average densities applicable to the region after a review of the figures used by Ashton during mining to be too high given the more recent mining by Crescent at nearby deposits. An average SG of 1.80t/m3 for oxidised, 2.40 t/m3 for transitional material and
	 2.75 t/m3 for Fresh rock were applied. CM/MM density values were assigned based on weathering profile and rock type using water immersion SG test work on rock samples collected during Crescent mining operations. In the Mafic units an average t/m3 of 1.84 was used in oxide, 2.49 Transitional and 2.78 Fresh. In the BIF formation an average t/m3 of 2.2 was used in oxide, 2.5 Transitional and 3.1 Fresh.
Classification	 Material has been classified Indicated and Inferred based on a number of criteria such as geological continuity, drill hole spacing, estimation pass and proximity to the existing open pits.
	 Chatterbox blocks within the 12.5m x 12.5m to 25m x 25m close spaced drill pattern that estimated in the first pass was classified as Indicated. Blocks where drill spacing increased to 25m x 50m to 50m x 50m and estimated in the second or third pass were classified as Inferred. Inferred resources are predominantly at the extents of the deposits and at depth. Rumor was classified as Inferred given the wet sampling issues and lack of more recent Crescent/FML drilling. Blocks in areas where the drilling extends to 100m spacing blocks have not been classified and are used for target generation for future drill programs. Resources along the West Laverton Trend have been classified as Indicated and Inferred. West Laverton/Rega resources that are within the tight drill spacing of 10m x 10m to 20m x 15m that primarily filled in the first search pass have been classified as Indicated.
	 These blocks are mostly in the transitional weathering zone between and beneath the existing open pits. Inferred resources at West Laverton/Rega are the fringe blocks where the drill spacing has increased to 30m x 30m to 30m x 60m and blocks at depth (~ below 85m from surface) with less drill penetration. Bulldog has been classified as Inferred and requires follow up drilling by FML to confirm the historic Ashton drilling.
	 Gladiator resources within the 25m grid drill spacing that predominantly estimated in the first pass were assigned as Indicated. Blocks at depth where drill coverage was patchy and smaller hanging wall lodes supported by less drill holes were assigned Inferred category. Cousin Murray was assigned Inferred given the lack of recent drilling and mining activity. A shape was created at Gladiator West to classify blocks in the predominantly 25m grid drill pattern.
	 Along the entire CM/MM trend Indicated Resources were those unmined blocks within the close drill spacing of 10m x 10m and 20m x 10m, these blocks primarily filled in the first search pass. Blocks that were either in areas of less drill density or at depth along the fringes of the lodes where follow up deeper holes are required were classified as Inferred. The two "gap" lodes north of Craigiemore were primarily assigned Inferred status. Golden Pinnacles resource area has been classified as Inferred and warrants further drilling.
Audits or reviews	No external Audits of the mineral resource have been conducted.
Discussion of relative accuracy/ confidence	 This is addressed in the relevant paragraph on Classification above. The Mineral Resource relates to global tonnage and grade estimates.

JORC Code, 2012 Edition – Table 1 Telegraph Section 1 Sampling Techniques and Data

Criteria	Commentary
Sampling techniques	 FML RC Sampling RC percussion drill chips were collected through a cone splitter from the drill rig. The bulk sample from drilling was placed in neat rows directly on the ground (not bagged) with the nominal 2-3kg calico split sub-sample placed on top of the corresponding pile. RC chips were passed through a cone splitter to achieve a nominal sample weight of approximately 3kg. The splitter was levelled at the beginning of each hole. Geological logging defined whether a sample was to be submitted as a 1m cone split sample or a 4m spear composite sample. Split samples (1m) were transferred to sample numbered calico bags for submission to the laboratory. Composite samples were spear sampled using a scoop to obtain a small representative sample and deposited into numbered sample bags. FML Diamond Sampling Diamond core was sampled across geologically identified zones of mineralisation, the sample widths varied between a minimum of 0.2m and a maximum of 1.2m with material sampled into waste hanging wall and footwall to capture the entire mineralised zone. The diamond core was marked up for sampling by the supervising geologist during the core logging process, with sample intervals determined by the presence of lithology, alteration and where applicable core loss. No sample included core loss. The core was cut in half using a core saw and the same half of the core (RHS looking downhole) was routinely sent to the laboratory for analysis. Some soft core was sampled half by using a bolster, and some fractured quartz core were cut in half by using manual diamond core saw to ensure half core was sampled. A small number of whole core samples where routinely collected for bulk density analysis. These samples were submitted to the same lab for gold analysis after bulk density measurement. Historic Sampling

Criteria	Commentary
Ontona	WMC RC samples were collected in plastic bags in 1m intervals, while diamond core was sampled to at 1m intervals or on geological
	contacts.
	Metex RC samples were either 1m riffle splits or 4m composite spear samples.
	The single Ashton Mines RC drill hole reference is unknown; however, this single
	hole is also of low-grade and not considered to have a large influence in the estimate.
Drilling	FML Drilling
techniques	RC drilling was conducted using a 5 3/8inch face sampling hammer for RC drilling. At help a small time down halo assess for RO halo was a small shall be a 40 minute size. Thus North Cooling Company for RO.
	 At hole completion, downhole surveys for RC holes were completed at a 10m interval by using True North Seeking Gyro tool. At hole completion diamond holes were survey using a single shot tool at a range of intervals between 20m and 50m, averaging 30m
	 At note completion diamond notes were survey using a single shot tool at a range of intervals between 20th and 50th, averaging 50th Diamond drill holes with dips less than 50 degrees were collared from surface to a predetermined depth using a rock roller bit.
	 Where possible on holes with dips more than 50 degrees an RC pre-collar was completed to improve drilling efficiency.
	All pre-collars where cased off and the diamond component of the drill hole completed using HQ3 equipment producing 63mm diameter.
	core.
	Wherever core conditions and hole orientation would allow, drill core was oriented by the drilling contractor using the electronic ACT III
	Tool.
	Historic Drilling
	RC drill methods were not recorded in WAMEX reports. WMC diamond holes had a RC pre-collars from existing RC holes.
	Metex RC holes were surveyed by Eastman single shot camera at hole completion.
Drill sample	FML Drilling
recovery	RC sample recovery was recorded by a visual estimate during the logging process.
	DD sample recovery was measured and calculated (core loss) during the logging process. DD core had generally reasonable recovery 100/ page loss in and around mineralization. Where this earn loss was exercised around HC and VHC gold assert it likely had a
	<10% core loss in and around mineralisation. Where this core loss was experienced around HG and VHG gold assays it likely had a material impact on the calculated intersection grade as all core loss was fully diluted and assigned a grade of 0.0g/t Au.
	Historic Drilling
	WMC did not document core loss in their annual report.
	Metex didn't note any sample quality issues in their drill logs.
Logging	FML Drilling
	All RC samples were geologically logged to record weathering, regolith, rock type, alteration, mineralisation, structure, texture and any
	other notable features that are present. All data is entered directly into validating digital software.
	All core samples were oriented where possible, marked at metre intervals and compared to the depth measurements on the core
	blocks. Any loss of core was noted and recorded in the drilling database.
	All diamond core was logged for structure, geology and geotechnical data using the same system as that for RC.
	Logging was qualitative, however the geologists often recorded quantitative mineral percentage ranges for the sulphide minerals
	 present. The logging information was transferred into the company's drilling database once the log was complete.
	 Diamond core was photographed one core tray at a time using a standardised photography jig. RC chip trays are routinely
	photographed.
	The entire length of all holes is geologically logged, except for rock roller diamond pre-collars which produce no sample.
	Historic Drilling
	WMC RC samples were logged to record colour, weathering, rock type and texture.
	Diamond core was logged to lithological boundaries; recording rock type, structure, texture, alteration and veining.
	Metex Drill logs captured colour, weathering, fabric, grainsize, rock type, alteration, veining.
Sub-sampling	FML Drilling
techniques and	All samples were collected in a pre-numbered calico bag bearing a unique sample ID. At the control of the second se
sample	 At the assay laboratory, all samples were oven dried, crushed to a nominal 10mm using a jaw crusher (core samples only) and weighed. Samples in excess of 3kg in weight were riffle split to achieve a maximum 3kg sample weight. All samples were pulverized to 90%
preparation	passing 75µm.
	Gold analysis was by 40g Fire Assay with an AAS Finish.
	Jinning Testing & Inspection completed the assay testing, with sample preparation completed in Kalgoorlie or Perth and analysis
	completed in Perth.
	The assay laboratories' sample preparation procedures follow industry best practice, with techniques and practices that are appropriate
	for this style of mineralisation. Pulp duplicates were taken at the pulverising stage and selective repeats conducted at the laboratories'
	discretion.
	QAQC checks involved inserting standards 1:20 samples (with minimum 3 standards every submission). Duplicate samples for RC
	were achieved by producing 2 samples for each metre one hole every 20th hole drilled and submitting all produced samples. The
	remaining bulk sample was also bagged to plastic bags for retention and further checks. Diamond core field duplicates were not taken.
	 Regular reviews of the sampling were carried out by the supervising geologist and senior field staff, to ensure all procedures were followed and best industry practice carried out.
	 The sample sizes were appropriate for the type, style and consistency of mineralisation encountered during this phase of exploration.
	Historic Drilling
	WMC RC samples were collected as 1m samples and submitted to the Windarra mine laboratory for Au analysis by fire assay.
	Diamond core was submitted as 1m samples or to geological contact to the Windarra mine laboratory for fire assay.
	Metex 1m RC samples were submitted to Genalysis for a Fire Assay with a 25g charge to a 0.01ppm detection limit. The 4m composite
	samples were analysed by aqua regia with a 10g charge

Criteria	Commentary
assay data and	The assay method and laboratory procedures were appropriate for this style of mineralisation. The fire assay technique was designed
laboratory tests	to measure total gold in the sample.
	No geophysical tools, spectrometers or handheld XRF instruments were used for assay determination.
	The QA/QC process described above was sufficient to establish acceptable levels of accuracy and precision. All results from assay
	standards and duplicates were scrutinised to ensure they fell within acceptable tolerances and where they didn't further analysis was
	conducted as appropriate.
	Umpire samples are collected on a routine basis will be submitted to independent ISO certified labs in 2019
	 Additional bulk mineralised RC samples have also been collected and retained for follow up QAQC, metallurgical and sample characterisation purposes.
	Historic Drilling
	 Notwithstanding the lack of information on WMC laboratory techniques, the assay method and laboratory procedures were appropriate for this style of mineralisation. The fire assay technique was designed to measure total gold in the sample. WMC successfully mined the Telegraph OP and Lancefield main lode for a number of years with documented reconciliation numbers. This is taken as an indication that WMC's drill hole sampling and analytical methods were adequate for resource / reserve calculation. Metex utilised standards and duplicates in the field samples and laboratory duplicates to monitor sample quality.
Verification of	Significant intervals were visually inspected by company geologists to correlate assay results to logged mineralisation. Consultants
sampling and	were not used for this process.
assaying	Primary logging data is sent in digital format to the company's Database Administrator (DBA) as often as was practicable. The DBA
	imports the data into an acQuire database, with assay results merged into the database upon receipt from the laboratory. Once loaded,
	data was extracted for verification by the geologist in charge of the project.
	Historic data has been validated against WAMEX logs were possible and validated when imported into the FML database.
Location of data	FMI Dulling
points	FML Drilling • Drill collars are surveyed after completion using a DGPS instrument. Where possible, all drill core was oriented by the drilling contractor
	using an ACT III electronic system.
	A True North Seeking Gyro for RC end of holes surveys or a Reflex single shot camera for diamond drilling was used for "single shot"
	surveys whilst advancing drilling.
	All coordinates and bearings use the MGA94 Zone 51 grid system.
	FML utilises Landgate sourced regional topographic maps and contours as well as internally produced survey pick-ups produced by
	the mining survey teams utilising DGPS base station instruments.
	After completion the drill hole locations were picked up by DGPS with accuracy of +/- 20cm.
	Historic Drilling
	WMC holes were surveyed by WMC survey staff in local mine grid
D ()	Metex holes were surveyed by a consultant survey company. RC holes were downhole surveyed by an Eastman Single Shot camera. The surveyed by a Consultant survey company. RC holes were downhole surveyed by an Eastman Single Shot camera.
Data spacing and distribution	Telegraph drill spacing approximates 25m x 20m along the open pit. Recent FML drilling targeted remaining resources beneath the
นเรนามนแบบ	current pit to an average 100m below surface.
Orientation of	 Spacing is deemed to be appropriate for the type of mineralisation Drilling was designed based on known/developing geological models, field mapping, verified historical data, cross-sectional and long-
Orientation of data in relation	
to geological	sectional interpretation.
structure	Where achievable, drill holes were oriented at right angles to strike of deposit, with dip optimised for drill capabilities and the dip of the are holds.
	ore body. • True widths have not been calculated for reported intersections. However, drill orientation was wherever possible consistently optimised
	to approximate true width of
	mineralisation.
Comple consults	• mineralisation. FML Drilling
Sample security	All samples were reconciled against the sample submission with any omissions or variations reported to FML.
	 All samples were bagged in a tied numbered calico bag. The bags were placed into plastic green bags with a sample submission sheet
	secured by cable ties and delivered directly from site to the Kalgoorlie laboratories by FML personnel at completion of each hole.
	WMC and Metex sample security is not recorded.
L	Time and meter satisfies accurate to not recorded.

Section 2 Reporting of Exploration Results (Criteria listed in the preceding section also apply to this section.)

Criteria	Commentary
Mineral tenement	The drilling was conducted on tenements 100% owned by Focus Minerals (Laverton) Pty Ltd.
and land tenure	All tenements are in good standing.
status	Various royalties may be in place as documented in the FML Annual Report 2018
	There are currently no registered Native Title claims over the Laverton project areas.
Exploration done by	Telegraph (Formerly Crown Jewels) was last mined as an open pit to about 70m depth by WMC between1984 and 1986 with
other parties	production of 20Koz Au.
	Later exploration has been performed by Metex/Delta Gold 1996/1997 and then Crescent Gold in 2010.
Geology	Telegraph geological sequence falls to the north of the Lancefield Mine and forms part of the well documented Lancefield Mine
	Sequence, comprising of footwall ultramafics overlain by a series of Ultramafic lavas, dolerites and basalt units with interflows of
	carbonaceous sediments. The ultramafic/mafic mine sequence is overlain by pelitic and arenaceous sediments.
	The Telegraph sequence strikes N-S dipping moderately to the East.
	Mineralisation primarily occurs within the east dipping W7 sheared interflow sediments near the base of a Komatiitic lava

Criteria	Commentary										
- Critoria	sequence. The Mineralisation is characterised by strong to intense silica-carbonate- sulphide alteration and replacement. In										
	historical logs usually described the altered shears as cherts.										
	• The altered shears range in width from 1-+6m the main host of the mineralisation is locally termed the W7 lode and lesser mineralisation is located on the footwall W6 lode (footwall to the W7 lode) and on the main lode position (Hanging wall to the W7										
	located on the u			•		,	on the n	idiii iodo position	i (i langing waii t	O the Wi	
						,	ain lode	position. Historic	production at L	ancefield	
		The W series lodes stratigraphically are footwall and west of the Lancefield Main lode position. Historic production at Lancefield for the W series lodes resulted in free milling mineralisation.									
	• In the plane of the W7 the core of Telegraph mineralisation has 280m strike and in general plunges moderately SE. Thicker and										
Drill hole information	high grade shoots within the core of Telegraph mineralisation plunge moderately to the NE. WAMEX References										
Driii noie imormation	WAMEX Report										
	Company			Drill Hol	e Number			A- Number	Report Date		
	Ashton Gold Mines Pty Ltd	LNP09	9					Unkn	nown		
	Western Mining	LFP00	16, LFP0017,	LFP004	3, LFP0	044, LFP0045	5.	Unkn	nown		
	Corporation Ltd	LFP01	55, LFP0181,	LFP018	32 LFP0	183, LFP0184					
	Н		98, LFP0199, LI								
			6, LFD058, LFD 44, LFP0345, LI	,		,	357.	14832	1985		
		LFP03	58, LFP0359, LI	FP0360, L	.FP0361, L	FP0362, LFP0	363,				
			88, LFP0390, LI 95, LFP0396, LI								
			95, LFP0390, L1 04, LFP0405, L1								
		LFP04	10, LFP0411, LI	FP0412, L	FP0413, L	FP0415, LFP04	416,				
			17, LFP0418, LI 23, LFP0424, LI								
			35, LFP0436, LI								
		LFP04	41, LFP0442, LI								
			47, LFP0448, LI	FP0449, L	.FP0450, I	_FP0451, LFP04	481,				
			LFP0482, LFP0483 LFD065, LFD066, LFD067, LFD068 16961 January								
			93, LFP0495,			497, LFP0498	3.				
		LFP04	99, LFP0500,	LFP050)1, LFP0	503, LFP0504	4,				
	LFP0505, LFP0506, LFP0507, LFP0508, LFP0509, LFP0510, LFP0511, LFP0581 LFP0605, LFP0606, LFP0607, LFP0608, LFP0609,										
	LFP0610, LFP0611, LFP0612, LFP0613, LFP0614,								June 1986		
		LFP0615, LFP0616, LFP0617 LFD100, LFD101, LFD102, LFD103, LFD104, LFD105									
	Metex Resources	td			42284 72705	1994 2006					
	Wickey Hosodroos	LRC018, LRC019, LRC020, LRC021, LRC022]	
	FML Drilled holes av	ailable on V	VAMEX								
						WAMEX					
	Company	Drill Hole Number Report A- F						port Date			
						Number					
		19LNRC0	02, 19LNRC003	3,							
			04, 19LNRC005								
	Focus		19LNRC006, 19LNRC007,								
	Minerals Ltd	19LNRC008, 19LNRC009, 19LNRC010, 19LNRC011,						2019			
		19LNRC012, 19LNRC013, 19LNRC014,									
		19LNRC0	15, 19LNRC016	i							
	Collar details of FML	. holes drille	d during 2019 a	re given b	elow						
	Hole ID	Easting	Northing	RL	Dip	Azimuth	Depth	Tenement			
			(MGA 94 Zone 5	1)		(MGA94)	(m)				
			Telegr	aph 2019	Drill Collars	3					
	19LNDD001	439604.7	6842055.7	450.9	-46.9	241.9	173.7	M38/37	_		
	19LNDD002	439585.8	6841921.6	451.1	-32.9	266.4	170.3	M38/37	_		
	19LNDD004	439602.0	6842057.8	451.1	-36.7	266.3	176.34	4 M38/37	_		
	19LNDD005	439603.2	6842056.8	451.0	-51.1	271.9	170.9	M38/37	_		
	19LNDD006	439597.7	6841956.8	451.9	-40.7	275.9	179.6	M38/37	_		
	19LNDD007	439586.3	6841918.3	450.8	-45.4	261.6	167.6	M38/37	_		
	19LNDD008	439578.1	6841876.5	450.9	-42.4	262.7	176.9	M38/37	_		
	19LNDD009										

Criteria	Commentary								
	19LNRC002	439572.8	6841876.9	451.4	-55.06	269.47	174	M38/37	
	19LNRC003	439581.5	6841921.3	451.1	-56.41	265.79	170	M38/37	
	19LNRC006	439449.2	6841699.0	451.6	-49.66	273.44	132	M38/37	
	19LNRC008	439491.0	6841782.4	451.3	-60.39	274.93	150	M38/37	
	19LNRC009	439514.9	6841787.4	451.1	-58.03	292.67	168	M38/37	
	19LNRC010	439586.4	6841934.2	451.4	-50.73	276.64	162	M38/37	
	19LNRC013	439513.1	6842164.9	451.1	-71.41	271.47	120	M38/37	
	19LNRC014	439494.7	6842186.6	451.4	-59.97	273.45	102	M38/37	
	19LNRC015	439563.9	6842143.0	451.3	-61.45	268.57	150	M38/37	
	19LNRC016	439600.2	6841978.7	452.3	-52.71	270.36	168	M38/37	
	19LNRC056	439456.7	6842313.2	451.3	-50.22	278.98	90	M38/37	
	19LNDD003	439585.8	6841892.4	451.3	-32.6	269.2	144.1	M38/37	
	19LNRC004	439446.8	6841755.1	451.4	-65.55	289.42	126	M38/37	
	19LNRC005	439433.7	6841730.8	450.8	-56.36	282.8	108	M38/37	
	19LNRC007	439478.1	6841706.8	451.2	-55.74	290.6	144	M38/37	
	19LNRC011	439461.3	6842225.4	450.9	-55.59	274.23	78	M38/37	
	19LNRC012	439488.2	6842221.1	451.2	-60.88	273.66	102	M38/37	
Data aggregation methods	The length we	eighted averag	e grades from o	liamond c	ore can inc	lude measure	d intervals o	f core loss.	to 3m internal dilution.
Relationship between					ch as poss	ible, however	the exact rel	ationship betw	een intercept width and
mineralization widths and intercept lengths			ted exactly in al						
Diagrams	·		I in this annound	ement. 3	D perspecti	ve views and	schematic c	ross-sections a	re included to illustrate
	the distributio	•	- - \A/AN4	· · · · · · · · · · · · · · · · · · ·					
Balanced reporting	Historic drill results are available on WAMEX Drilling results are spectral in a halogood constitute that The ASX appropriate FMI halogophaya actual locations of halogophaya.							ctual locations of holos	
	Drilling results are reported in a balanced reporting style. The ASX announcement for FML holes shows actual locations of hod drilled, and representative sections as appropriate.						ictual locations of floles		
Other substantive exploration data	There is no other material exploration data to report at this time.								
Further work	FML anticipates additional drilling to follow up on encouraging results in Laverton.								

Criteria	Commentary
Onteria Database integrity	·
Site visits	 Alex Aaltonen, the Competent Person for Sections 1 and 2 of Table 1 is FML's General Manager - Exploration and conducts regular site visits. Hannah Kosovich, the Competent Person for Section 3 of Table 1 is FML's Resource Geologist and last visited site in September 2019.
Geological interpretation	 All Focus drill holes and historic drill holes, mining data and pit mapping / observations were used to guide the geological interpretation of the mineralisation. The mineralised geological interpretation was digitized in Micromine software on a section by section basis. An approximate 0.5g/t cutoff was used, however sub 0.5g/t samples were included for continuity. Minor deviation only of the lode geometry was noticed between drill holes along strike and down-dip.
Dimensions	 The Telegraph deposit has been modelled over a total strike length of 800m. Multiple lodes were modelled however the W7 lode carries most of the gold; two hanging wall lodes (including Main Lode) were modelled and one footwall lode (W6). All lodes have been modelled

Criteria	Commentary
	from surface to approximately 300m below surface. Mineralisation has an average width of 3m for the W7 lode and 2m for the minor
	lodes.
Estimation and	A total of 374 drill holes were used in the Estimation; 97 diamond holes, most with an RC pre-collar and 277 RC holes.
modelling	The drill hole samples were composited to 1m within each domain. This is the dominant sampling interval.
techniques	• All domain boundaries were considered "hard" boundaries and no drill hole information were used by another domain in the estimation.
	Composited assay values of each domain were exported to a text file (.csv) and imported into Snowden Supervisor for geostatistical
	analysis.
	A review of histograms, probability plots and mean/variance plots for the main lode domain revealed outlier sample values. A top-cut
	of 22g/t Au was used for the different lodes, with assays above the top-cut were set to the top-cut value.
	Variograms were modelled in Supervisor.
	GEOVIA Surpac Software was used for the estimation and modelling process. The model was created in GDA 94 grid co-ordinates.
	Block sizes for the model were
	• 12.5m in Y, 3m in X and 3m in Z direction. Sub celling of the parent blocks was permitted to 3.125m in the Y direction, 0.75m in the X
	direction and 1.5m in the Z direction. Sub-blocking was used to best fill the wireframes and inherit the grade of the parent block. No
	rotation was applied to the orientation of the blocks.
	• Block size is approximately ½ of the average drill hole spacing along strike and across strike was selected to best fill the narrow lode
	wireframe volumes.
	An Ordinary Kriging (OK) estimation technique was selected and used the variograms modelled in Supervisor.
	• The main lode was estimated using a minimum (6) and maximum (16) samples were selected based on a Kriging Neighborhood
	analysis in Supervisor.
	• The minor lodes were estimated using a minimum (4) and maximum (8) samples to generate a local estimate that reflected the nearby
	samples.
	An elliptical search was used based on range/ratio of the Variograms.
	Three search passes were run in order to fill the block model with estimated Au values. After each search pass the search range was
	approximately doubled and in the second search pass minimum number of samples was decreased.
	The estimate was validated by a number of methods. An initial visual review was done by comparing estimated blocks and raw drill
	holes.
	Tonnage weighted mean grades were compared for the main lode with no major differences.
	Swath plots of drill hole values and estimated Au grades by northing and RL were run for the main domain and showed that the
	estimated grades honored the trend of the drilling data.
Moisture	Tonnages are estimated on a dry basis.
Cut-off	The mineral resource for Telegraph has been reported above a 0.8g/t Au cut-off.
parameters Mining factors or	The Telegraph deposit would be mined by a cut-back on the existing open pit.
assumptions	The Telegraph deposit would be mined by a cut-back on the existing open pit.
Metallurgical	FML have submitted samples for metallurgical test work and as at time of reporting the results are outstanding.
factors or	Telegraph West Lodes have been modelled and historical WMC production indicates mineralisation was non-refractory.
assumptions	
Environmental	Telegraph has been historically mined by open pit methods in the mid-1980's by WMC.
factors or	
assumptions	
Bulk density	Density values were assigned based on weathering profile and SG test work on FML diamond core samples from different weathering Assessment SG of 4.2 for a side weathering profile and SG test work on FML diamond core samples from different weathering Assessment SG of 4.2 for a side weathering profile and SG test work on FML diamond core samples from different weathering The second sec
	zones. An average SG of 1.8 for oxide weathering profile, 2.5 for transitional material and 2.86 for Fresh rock were applied.
01 '5 '	The water immersion technique was used for these determinations.
Classification	Material has been classified Indicated and Inferred based on a number of criteria such as geological continuity, drill hole spacing, activation area provinity to spirity to a principle.
	estimation pass, proximity to existing open pit.
Audits or reviews	The Telegraph October 2019 Mineral Resource was modelled in house by the exploration group. The resulting wireframes were invariant for the Company for a virial facility to the company of the com
	imported into Surpac for review/validation by Hannah Kosovich (FML Resource Geologist).
D:	The resource model has been reviewed in house for consistency with the database. This is a black of the second of the seco
Discussion of	This is addressed in the relevant paragraph on Classification above. The Ministry of the Control of the C
relative accuracy/ confidence	The Mineral Resource relates to global tonnage and grade estimates
COMMUNICE	

JORC Code, 2012 Edition – Table 1 Lancefield Far North Deposit Section 1 Sampling Techniques and Data (Criteria in this section apply to all succeeding sections.)

Contena in this section apply to an succeeding sections.)							
Criteria	Commentary						
Sampling	FML RC Sampling						
techniques	Focus Minerals Ltd (FML) RC percussion drill chips were collected through a cyclone and riffle splitter. Samples were collected as 4m						
	composites or as 1m samples through mineralised ground or interesting geology. Where the 4m composite samples returned greater						
	than 0.20g/t Au, 1m samples were submitted. The spoils were either bagged per metre in appropriately sized plastic bags or placed on						
	the ground and left in neat rows at 1m intervals with an accompanying cone split 1m calico sample.						
	• At the assay laboratory all samples were oven dried, crushed to a nominal 10mm using a jaw crusher (core samples only) and weighed.						
	Samples in excess of 3kg in weight were riffle split to achieve a maximum 3kg sample weight before being pulverized to 90% passing						

Criteria	Commentary
	75μm.
Drilling	Years 2019 onward FML RC drilling was conducted using a 5 3/8inch face sampling hammer for RC drilling.
techniques	• At hole completion, downhole surveys for RC holes were completed at a 10m interval by using True North Seeking Gyro tool. Otherwise,
	a single shot Eastman camera downhole survey was used either "in-rod" or "open hole".
	Earlier drilling by FML was completed using an RC face sampling hammer. Most holes were surveyed upon completion of drilling using
	an EMS camera open hole.
Drill sample	FML sample recovery was recorded by a visual estimate during the logging process.
recovery	• All RC samples were drilled dry whenever possible to maximize recovery, with water injection on the outside return to minimise dust.
Logging	The information of logging techniques below applies to the drill holes drilled by FML only.
	 All RC samples were geologically logged to record weathering, regolith, rock type, alteration, mineralisation, veining, structure and texture and any other notable features that are present.
	The logging information was transferred into the company's drilling database once the log was complete.
	 Logging was qualitative, however the geologists often recorded quantitative mineral percentage ranges for the sulphide minerals
	present.
	RC chip trays are wet photographed.
	The entire length of all holes is logged.
Sub-sampling	• FML RC samples were riffle split to a nominal 2.5kg to 3kg sample weight. The drilling method was designed to maximise sample
techniques and	recovery and delivery of a clean, representative sample into the calico bag.
sample	Prior to 2019 - samples were submitted to ALS or Kal Assay for analysis.
preparation	 2019 onward FML samples were submitted to Jinning lab in Kalgoorlie with gold analysed by fire assay.
	Where possible all RC samples were drilled dry to maximise recovery. Sample condition was recorded (wet, dry, or damp) at the time
	of sampling and recorded in the database.
	• The samples were collected in a pre-numbered calico bag bearing a unique sample ID. Samples were crushed to 75µm at the laboratory
	and riffle split (if required) to a maximum 3kg sample weight. Gold analysis was primarily a 40g Fire Assay for individual samples with
	an ICP-OES or AAS Finish.
	 The assay laboratories' sample preparation procedures follow industry best practice, with techniques and practices that are appropriate for this style of mineralisation. Pulp duplicates were taken at the pulverising stage and selective repeats conducted at the laboratories'
	discretion.
	 FML QAQC checks involved inserting a certified standard or blank alternating every 20 samples. A minimum of 3 standards was
	inserted for every sample batch submitted.
	 The sample sizes are considered to be appropriate for the type, style and consistency of mineralisation encountered during this phase
	of exploration.
	Laboratory repeat checks were also run on the assay data.
Quality of assay	The assay method and laboratory procedures were appropriate for this style of mineralisation. The fire assay technique was designed
data and	to measure total gold in the sample.
laboratory tests	No geophysical tools, spectrometers or handheld XRF instruments were used.
	• The QA/QC process described above was sufficient to establish acceptable levels of accuracy and precision. All results from assay
	standards and duplicates were scrutinised to ensure they fell within acceptable tolerances.
Verification of	, , , , , , , , , , , , , , , , , , , ,
sampling and assaying	were not used for this process.
assayiiig	Primary data is sent in digital format to the company's Database Administrator (DBA) as often as was practicable. The DBA imports the data into an applying database with applying the database was practicable. The DBA imports The DBA impo
	the data into an acQuire database, with assay results merged into the database upon receipt from the laboratory. Once loaded, data was extracted for verification by the geologist in charge of the project.
	 No adjustments were made to any current or historic data. If data could not be validated to a reasonable level of certainty it was not
	used in any resource estimations.
Location of data	All 2019 onwards FML RC holes were down hole surveyed using a north seeking gyro.
points	All pre 2019 FML holes were surveyed using an EMS system.
	After completion, the drill hole locations were picked up by DGPS with accuracy of +/-20cm.
	All coordinates and bearings use the MGA94 Zone 51 grid system.
	FML utilises Landgate sourced regional topographic maps and contours as well as internally produced survey pick-ups produced by
	the mining survey teams utilising DGPS base station instruments.
	• Detailed drone topography and imagery has also been acquired over the project area to provide additional topographic detail and
	spatial accuracy.
Data spacing and	Drill spacing at Lancefield Far North within resource area is approximately 50m x 40m. The average vertical depth of the RC drilling is
distribution	90m, with a maximum depth of 102m.
Orientation of	Drilling was designed based on known geological models, field mapping, verified historical data and cross-sectional interpretation.
data in relation to	The vast majority of holes are oriented at right angles to the strike of historic mineralization, with dip optimised for drill capabilities and
geological structure	the dip of the ore body.
Sample security	All samples were reconciled against the sample submission with any omissions or variations reported to FML.
	All samples were bagged in a tied numbered calico bag, grouped into tied green plastic bags.
	 The bags were placed into bulk bags or pods with a sample submission sheet and delivered directly from site to the Kalgoorlie
	laboratories by FML personnel.
	Historic sample security is not recorded.
	· · · · · · · · · · · · · · · · · · ·

Section 2 Reporting of Exploration Results

(Criteria listed in the n											
(Criteria listed in the p		oling section also mmentary	αμμι y τ Ο trilS	oʊuull.)							
Mineral tenement			r North is Id	cated within	Explora	tion I e	ase F38/3	3186 regis	stered to Focus Mine	erals I to and Focus Operation	s Ptv I td of
and land tenure	ľ	 Lancefield Far North is located within Exploration Lease E38/3186, registered to Focus Minerals Ltd. and Focus Operations Pty Ltd of Perth, Western Australia and which is current until 3 May 2022. 									
status	١.						•		o no Lavorton claim	e have progressed to determin	od status
Exploration done		 The Nyalpa Pirniku claim cover the Laverton Project tenure. At this stage no Laverton claims have progressed to determined status. Lancefield Far North and adjacent prospects have been explored in the past with geological, geophysical and drilling techniques. 									
by other parties	ľ	Geological mapping, ground magnetics, aeromagnetics and soil sampling have been routinely carried out by other parties since the									
by other parties		_		-		-		-	-		s since the
Geology	 mid 1980's. Drilling included rotary air blast, reverse circulation, aircore, vacuum drilling and auger. The Lancefield Far North deposit mineralisation is hosted by the East dipping Lancefield Shear. Mineralisation is a 									inted with a	
Geology	ľ	silicified horizons of interflow black shale-chert dipping at 20 degrees to the east.									ialeu Willi a
	١.									ootwall of high magnesium ba	ealte to the
	ľ	west.	iu onean is	Sanuwichieu	with a fi	anging	wali oi pii	ilow basai	t to the east and a h	ootwall of flight magnesium ba	isalis lo lile
	١.		on of the ore	hody is tah	ular and	ctacko	Ч				
		 The orientation of the ore body is tabular and stacked. The Lancefield Far North deposit averages varies from 2m to +7m width over 300m strike and open along strike. 									
					-					-	aintant and
	•		ensional drii	ing conduct	ea in 20	19 and	2021 nas	s snown th	ie mineralisation at i	Lancefield Far North to be con	isistent and
D ::: 1		predictable.	2.11.0	1 14/4 14T1/							
Drill hole Information	•	Holes not ava						ACV	D-1 T:41-	ACV Dalassa Data	Ì
imormauon		Cor	mpany		rill Hole RC047, 1			ASX	Release Title	ASX Release Date	
					RC047, RC049, 1						
					RC051, 1		2052				
		FC	CUS		RC053, 1				Update - Laverton	28-Apr-21	
				19LN	RC055, 2	21LNR	2001,	G	iold Project	r	
				21LN	RC002, 2	21LNR	C003,				
				21LN	IRC004,	21LNR	C005				
	١.	Lancefield Fa	ar North Sig	nificant Inter	roante nro	aviouel	v renorted	ļ.			
	Ĭ	Lancelleid i d	Easting	Northing	RL	I VIOUSI	Azimuth			Intersection	
		Hole ID		GA 94 Zone 51		Dip					
									0.5g/t Au cut off an up		
		19LNRC047 19LNRC048	442554 442614	6849459 6849467	446 446	-51 -50	273 272	174.0 180.0	1.00m @ 0.81g/t from 2 1.00m @ 0.51g/t from 2		
		13214110040	772017	0040401	710	-50	212	100.0	1.00m @ 1g/t from 34m		
		19LNRC049	442672	6849453	469	-52	275	180.0	1.00m @ 1.72g/t from 6		
									2.00m @ 0.87g/t from 7 2.00m @ 1.02g/t from 1		
									1.00m @ 0.54g/t from 1	50m for (GxM 1)	
		19LNRC050	442550	6849362	468	-51	269	54.0	1.00m @ 0.74g/t from 3 1.00m @ 0.55g/t from 4		
		19LNRC051	442611	6849361	468	-51	269	174.0	1.00m @ 1.32g/t from 2		
		19LNRC052	442670	6849358	469	-50	266	174.0	9.00m @ 2.31g/t from 3		
		19LNRC053 19LNRC054	442594 442640	6849254 6849257	468 453	-51 -51	268 266	174.0 174.0	3.00m @ 1.11g/t from 1 2.00m @ 1.87g/t from 1		
									2.00m @ 1.36g/t from 3	37m for (GxM 3)	
		19LNRC055	442653	6849258	469	-51	268	168.0	13.00m @ 1.22g/t from 6		
		21LNRC001	442752	6849259	470	-61	272	90.0	1.00m @ 0.92g/t from 4	0 1 (0 11.1)	
									15.00m @ 0.77g/t from	49m for (GxM 12)	
		21LNRC002	442792	6849260	470	-60	271	102.0	2.00m @ 1.59g/t from 7 2.00m @ 0.56g/t from 5		
		21211110002	112102	0010200		00	2,,	102.0	4.00m @ 0.66g/t from 6	64m for (GxM 3)	
		21LNRC003	442709	6849359	470	60	274	90.0	3.00m @ 1.43g/t from 7 6.00m @ 3.08g/t from 6		
		Z ILINROUS	442709	0049309	4/0	-60	2/4	90.0	4.00m @ 0.5g/t from 76	Sm for (GxM 2)	
		21LNRC004	442750	6849358	470	-59	272	96.0	3.00m @ 1.01g/t from 8	30m for (GxM 3)	
		21LNRC005	442712	6849453	470	-60	271	132.0	4.00m @ 0.6g/t from 92 4.00m @ 0.77g/t from 6		
		ZILINIOUUS	772112	0040400	470	-00	271	102.0	6.00m @ 2.1g/t from 76		
				. 0: :::	<u> </u>	<u> </u>	<u> </u>		1.00m @ 0.7g/t from 91	m for (GxM 1)	
	•	New Lancefie	eld Far Nort	h Significant	t Intercep	ts not p	previously	reported:			
			Easting	Northing	RL	Dip	Azimuth				
		Hole ID		MGA 94 Zon			(MGA94			ersection	
		CN)	X Drill Collar	s. Significant	t Intersec	tions ca	alculated a	it 0.5g/t Au	cut off an up to 3m in		
		21LNRC006	442689	6849409	468	-60	270	90		Ag/t from 61m for (GxM 8) Gg/t from 74m for (GxM 3)	
	 -		-	-	+			-		g/t from 73m for (GxM 12)	
		21LNRC007	442730	6849409	468	-60	270	96		4g/t from 93m for (GxM 1)	
	-	21LNRC008	442679	6849308	467	-60	270	66		g/t from 43m for (GxM 5)	
	-									Ig/t from 58m for (GxM 5)	
		21LNRC009	442717	6849310	467	-60	270	78		5g/t from 71m for (GxM 2)	
				-	1 1			<u> </u>	1		
Data a		Minarelle	:-t		ad at - ^	E a lk A		4h a ==!=!		of the fee DO halos and OO C	an allama a a d
Data aggregation methods	•				ea at a 0.	og/t Au	cut-off wi	ıın a mınım	ium reporting width o	of 1m for RC holes and 0.3m for	alamond
		holes, comp			min a == !!	otic =	o mercele	00 pos-11 1	lo houses #	not rolationable between int	ont u.i.dil-
Relationship	•	noies were	urillea ortr	iogonal to f	nineralis	alion a	is much a	as possibl	ie, nowever the exa	act relationship between interc	ept wiath

Criteria	Commentary
between mineralisation widths and intercept lengths	and true width cannot be estimated exactly in all cases.
Diagrams	Refer to Figures and Tables in body of the release.
Balanced reporting	Drilling results are reported in a balanced reporting style. The ASX announcement for Focus Minerals holes shows actual locations of holes drilled, and representative sections as appropriate
Other substantive exploration data	There is no other material exploration data to report at this time.
Further work	Metallurgical testwork and geotechnical study will be initiated in the next 24 months

Criteria	Commentary					
Database integrity	 FML data was geologically logged electronically, collar and downhole surveys were also received electronically as was the laboratory analysis results. These electronic files were loaded into an acQuire database by either consultants rOREdata or the company in-house Database Administrator. Data was routinely extracted to Microsoft Access during the drilling program for validation by the geologist in charge of the project. FML's database is a Microsoft SQL Server database (acQuire), which is case sensitive, relational, and normalised to the Third Normal Form. As a result of normalisation, the following data integrity categories exist: Entity Integrity: no duplicate rows in a table, eliminated redundancy and chance of error. Domain Integrity: Enforces valid entries for a given column by restricting the type, the format, or a range of values. Referential Integrity: Rows cannot be deleted which are used by other records. User-Defined Integrity: business rules enforced by acQuire and validation codes set up by FML. Additionally, in-house validation scripts are routinely run in acQuire on FML's database and they include the following checks: Missing collar information Missing logging, sampling, downhole survey data and hole diameter Overlapping intervals in geological logging, sampling, down hole surveys Checks for character data in numeric fields. Data extracted from the database were validated visually in GEOVIA Surpac software and Seequent Leapfrog software. Also, when loading the data any errors regarding missing values and overlaps are highlighted. 					
	Historic data has been validated against WAMEX reports where possible.					
Site visits	 Alex Aaltonen, the Competent Person for Sections 1 and 2 of Table 1 is FML's General Manager - Exploration and conducts regular site visits. Hannah Kosovich, the Competent Person for Section 3 of Table 1 is FML's Resource Geologist and last visited site in September 2019. 					
Geological	All available drill hole data was used to guide the geological interpretation of the mineralisation.					
interpretation	 Further drilling by FML in 2021 confirmed the mineralisation interpretation from the 2019. Four stacked lodes striking NNE and dipping gently (~ 20°) to the east have been interpreted. Minor deviation only of the lode geometry was noticed between drill holes along strike and down-dip within each lode. 					
Dimensions	 The Lancefield Far North – Lancefield-Telegraph-Wedge strikes SSW – NNE over 9km Lancefield Far North mineralisation has been modelled over 300m, the lodes have been interpreted from near surface t approximately 110m below surface to the 360mRL. 					
Estimation	 The average thickness of the lodes is 3m. The drill hole samples were composited to 1m within each domain. This is the dominant sampling interval. 					
and modelling techniques	 Composited assay values of each domain were imported into Snowden Supervisor for geostatistical analysis. A review of histograms, probability plots and mean/variance plots for each domain revealed some outlier sample values. Top capping of higher Au values within each domain was carried out with Au values above the cut-off grade reset to the cut-off 					
	grade. Only 1 grade was capped to 8ppm Au. Due to the small data set meaningful Variograms could not be generated. Datamine Software was used for the estimation and modelling process. The model was created in GDA 94 grid co-ordinates. Block sizes for the model were 25m in Y, 10m in X and 5m in Z direction. Sub-celling of the parent blocks was permitted to 3.125m in the Y direction, 1.25m in the X direction and 2.5m in the Z direction. Sub-blocking was used to best fill the wireframes and inherit the grade of the parent block. Block size is approximately ½ of the average drill hole spacing. An Inverse Distance Squared estimation technique was selected given the lack of variography. Minimum (6) and maximum (14) sample numbers were selected, this was dropped to a minimum (4) samples on the second and third search pass. An elliptical search was used based on the orientation of the modelled lodes. Three search passes were run in order to fill the block model with estimated Au values. The search distance was doubled between each estimation run. The estimate was validated by a number of methods. An initial visual review was done by comparing estimated blocks and raw drill holes. Tonnage weighted mean grades were compared for all lodes with the raw and top-capped drill hole values. There were no major differences.					

Criteria	Commentary					
	Swath plots of drill hole values and estimated Au grades by northing and RL were reviewed and showed that the estimated grades honoured the trend of the drilling data.					
Moisture	Tonnages are estimated on a dry basis.					
Cut-off parameters	 The Resources for Lancefield Far North have been reported above a 0.5g/t cut-off for open pit above 360mRL ~ 110m below surface. 					
Mining factors or assumptions	The Lancefield Far North deposit would be mined by open-cut methods.					
Metallurgical factors or	 Metallurgical test work is yet to be carried out at Lancefield Far North, however nearby Wedge and Lancefield North had tests performed. 					
assumptions	 Metallurgical test work was carried out by AMMTEC on behalf of Hill Minerals NL in August and September 1988. An end of mine report by Ashton Gold states mill recoveries were typically in the range of 94% - 95% A single sample of fresh rock from Wedge was submitted for gravity and leach recovery metallurgical test work. The grecovery was 14.2%. The leach returned 74.8% recovery after 8 hrs. A single sample of fresh rock from Lancefield North was tested for gravity and leach recovery. The gravity recovery was 11.5° the leach returned 94.9% recovery after 8hrs. 					
Environmental factors or assumptions	The tenement is within the Laverton Water Reserve.					
Bulk density	Density values were assigned based on weathering profile. A value of 2.0 t/m³ was applied to oxide blocks, 2.49 t/m³ was applied to transitional material blocks and a value of 2.80 t/m³ applied to Fresh Rock.					
Classification	 Resources have been classified as Inferred based on geological confidence in the geometry and continuity of the lodes and the use of only recent FML RC drillholes when estimating the resource. Sub-Inferred blocks exist at the northern and southern extension of the mineralisation where it has been inferred beyond reasonable distance past the last line of FML drilling. 					
Audits or reviews	No external audit or review has been carried out.					
Discussion of relative accuracy/ confidence	 This is addressed in the relevant paragraph on Classification above. The Mineral Resource relates to global tonnage and grade estimates. 					

JORC Code, 2012 Edition – Table 1 Lancefield

Section 1 Sampling Techniques and Data (Criteria in this section apply to all succeeding sections.)

	ion apply to all succeeding sections.)
Criteria	Commentary
Sampling	This report relates to results from Reverse Circulation (RC) and diamond core (DDH) drilling.
techniques	Lancefield has been drilled by various companies over the years, this report contains information on holes drilled by Western Mining Corporation Ltd (WMC), Golden Plateau N.L (GPNL), Metex Resources N.L (Metex) and Focus Minerals Ltd (FML).
	WMC drilled pre-collars on their surface diamond holes that were not sampled. Diamond core was sampled at 1m intervals or on geological contacts.
	GPNL stated diamond core was sampled at 0.5 to 1m intervals or geological contacts.
	Metex sampled and assayed for gold over the entire drill hole. Pre-collar drill chips were spear sampled in 5m composites using a 50mm PVC pipe tube. Unaltered or unmineralised core intervals were filleted and composited up to 5m. Zones of sulphide mineralisation and/or alteration were half core sampled up to 1m or geological contact.
	The information of sampling techniques below applies to the drill holes drilled by Focus Minerals (FML) only.
	• RC percussion drill chips were collected through a cyclone and in-line cone splitter under driller control. RC samples were collected on a 1m basis. Diamond core was sampled across identified zones of mineralisation by site geologists, the sample widths varied between a nominal minimum of 0.2m and a nominal maximum of 1m.
	The diamond core was marked up for sampling by the supervising geologist during the core logging process, with sample intervals determined by the presence of mineralisation and/or alteration. The core was cut in half using an automatic core saw. Samples for assay were put into pre-numbered calico bags.
	• RC chips were passed through a cone splitter to achieve a sample weight of approximately 3kg. The splitter was levelled at the beginning of each hole using a bullseye level. The spoils were collected in green bags at 1m intervals. Samples for assay were collected in pre-numbered calico bags.
	 At the assay laboratory all calico bagged assay samples were oven dried, core samples (only) crushed to a nominal 10mm using a jaw crusher and weighed. Samples in excess of 3kg in weight were riffle split to achieve a maximum 3kg sample weight before being pulverized to 90% passing 75µm.
	Duplicate samples were collected from RC pre-collars at the rate of 5 per 100m (every 20m). The duplicates were collected directly from the cone splitter at the same time as the primary sample. The duplicates were of similar weight to the primary sample and were treated identically to the primary sample. No duplicates were collected from the diamond core material.
	• Standards of appropriate grade were inserted into the RC sample runs at a rate of 3 per 100m (1 per 25m – excepting where it clashed with a duplicate position).
	No blanks were used as many of primary samples on the project recorded assays below or close to the detection limit making the role of the blank superfluous. Instead gold geochemical standards with low expected values were utilised regularly.

Criteria	Commentary
Drilling	WMC diamond holes drilled from surface usually had an RC pre-collar from surface to approx. 70m. Underground diamond drilling was
techniques	also conducted from available drive cuddies.
	GPNL diamond drilling was carried out by tricone drill bit from surface to approximately 100m, switching to HQ and finally NQ as the
	drill hole progressed.
	Metex drilled RC pre-collars to a maximum depth of 96.3m, diamond drilling was then used to complete the holes using HQ and NQ And the state of the state o
	core barrels. The drilling was directional and Navi drilling used to make directional corrections or cut wedges when drilling the secondary "daughter" hole off the first completed drill hole.
	Downhole surveys were conducted by either Eastman single shot camera or gyroscopic data in areas of extreme magnetic deviation.
	Drill core was oriented using a spear tip method which was successful 50% of the time.
	All FML drilling was completed using RC gear with face sampling hammer for the pre-collar, followed by HQ (if required by ground)
	conditions) and then NQ2 size diamond core equipment. As the holes were collared vertical, the core in the upper part of the hole was
	not oriented due to limitations of the core orientation system available. Deeper parts of the holes were oriented by the drilling contractor
	using an EzyMark system. Holes were surveyed upon completion of drilling initially using a north-seeking gyroscope tool within the rod
Drill sample	string. • WMC did not document drill recoveries in their annual reports.
recovery	GPNL did not document drill recoveries in their annual reports.
	Metex states no significant core loss was encountered with all recoveries averaging 99% or better.
	FML RC sample recovery was recorded by a visual estimate during the logging process. Diamond core recovery was calculated by
	measuring the drill core against drill rod length (as annotated on core blocks). Recoveries for FML drilling were good.
Logging	WMC logged the diamond core to lithological boundaries; recording rock type, structure, texture, alteration and veining. The pre-collar
	drill cuttings do not appear to have been logged.
	GPNL logged the diamond core to lithological boundaries; recording weathering, rock type, structure, texture, alteration, veining and solars. The Triespe drill suffice were not learned.
	 colour. The Tricone drill cuttings were not logged. Metex logged the entire drill hole including the RC pre-collar chips for weathering, rock type, structure, texture, alteration, veining,
	mineralisation and colour. Drill core was photographed wet and dry prior to cutting.
	The information of logging techniques below applies to the drill holes drilled by FML only. Core samples were oriented where possible,
	marked into metre intervals and compared to the depth measurements on the core blocks. Any loss of core was noted and recorded
	in the drilling database.
	All RC samples were geologically logged to record weathering, regolith, rock type, colour, alteration, mineralisation, structure and to the samples were geologically logged to record weathering, regolith, rock type, colour, alteration, mineralisation, structure and to the samples were geologically logged to record weathering, regolith, rock type, colour, alteration, mineralisation, structure and the samples were geologically logged to record weathering, regolith, rock type, colour, alteration, mineralisation, structure and the samples were geologically logged to record weathering, regolith, rock type, colour, alteration, mineralisation, structure and the samples were geologically logged to record weathering.
	texture and any other notable features that are present. In addition to parameters logged over RC chips, all diamond core was also logged for structure. If an orientation line was available,
	structure orientation was recorded.
	The logging information was transferred into the company's drilling database once the log was complete.
	Logging was qualitative, however the geologists often recorded quantitative mineral percentage ranges for the sulphide minerals
	present.
	Diamond core was photographed one core tray at a time using a standardised photography jig.
	Samples from RC holes were archived in standard 20m plastic chip trays. The standard 20m plastic chip trays.
Sub-sampling	• The entire length of all holes was logged. The bulk of the WMC complementation and problem was conducted at the poorty. Windows Nickel Brainet laboratory and problem.
techniques and	The bulk of the WMC sample preparation and analysis were conducted at the nearby Windarra Nickel Project laboratory and records of the methods used to analyse the samples have not been found.
sample	GPNL submitted drill core as 2.5-3kg samples in pre-numbered bags for analysis to either Analabs or Genalysis where it was crushed,
preparation	single stage mixed and ground. The crushed core was sampled in triplicate for gold by a fire assay on a 50g charge to a lower detection
	limit of 0.01 ppm gold. As, Ag, Cu and Ni were also analysed on the original sample only.
	Metex samples were submitted to Amdel Laboratories in Kalgoorlie for analysis by 50g fire assay to a lower detection limit of 0.01ppm
	Au.
	 The information of sub-sampling and sample preparation below applies to the drill holes drilled by FML only. Core samples were taken from half core, cut using an automatic core saw. The remainder of the core was retained in core trays tagged
	Core samples were taken from half core, cut using an automatic core saw. The remainder of the core was retained in core trays tagged with a hole number and metre mark.
	RC samples were cone split to a nominal 2.5kg to 3kg sample weight. The drilling method was designed to maximise sample recovery
	and delivery of a clean, representative sample into the calico bag.
	• The samples were collected in a pre-numbered calico bag bearing a unique sample ID. Samples were crushed to 75µm at the laboratory
	and riffle split (if required) to a maximum 3kg sample weight. Gold analysis was a 40g Fire Assay for individual samples with an ICP-
	OES or AAS Finish.
	• The assay laboratories' sample preparation procedures follow industry best practice, with techniques and practices that are appropriate for this style of mineralisation. Pulp duplicates were taken at the pulversing stage and selective repeats conducted at the laboratories'
	for this style of mineralisation. Pulp duplicates were taken at the pulverising stage and selective repeats conducted at the laboratories' discretion.
	For RC sampling, duplicates were collected directly from the cone splitter every 20th sample number (5 duplicates per 100 samples).
	Diamond core field duplicates were not taken. Standards were inserted every 25th sample number with the exception of numbers
	ending in "00" (reserved for duplicate in RC sampling). All sample despatches had multiple standards inserted.
	Regular reviews of the sampling were carried out by the supervising geologist and senior field staff, to ensure all procedures were
	followed and best industry practice carried out.

Criteria	Commentary
	The sample sizes were considered to be appropriate for the type, style and consistency of mineralisation encountered during this phase of exploration.
Quality of assay data and laboratory tests	 Notwithstanding the lack of information on WMC laboratory techniques, the assay method and laboratory procedures were appropriate for this style of mineralisation. The fire assay technique was designed to measure total gold in the sample. No geophysical tools, spectrometers or handheld XRF instruments were used. The QA/QC process described above was sufficient to establish acceptable levels of accuracy and precision. All results from assay standards and duplicates were scrutinised to ensure they fell within acceptable tolerances. WMC successfully mined Lancefield main lode for a number of years with documented reconciliation numbers. This is taken as an indication that WMC's drill hole sampling and analytical methods were adequate for resource / reserve calculation.
Verification of sampling and assaying	 Significant intervals were visually inspected by company geologists to correlate assay results to logged mineralisation. Primary data is sent in digital format to the company's Database Administrator (DBA) as often as was practicable. The DBA imports the data into an acQuire database, with assay results merged into the database upon receipt from the laboratory. Once loaded, data was extracted for verification by the geologist in charge of the project. No adjustments were made to any current or historic data. If data could not be validated to a reasonable level of certainty it was not used in any resource estimations.
Location of data points	 WMC drill collars would have been surveyed by the site mine surveyors in a local mine grid. Down hole surveys were by Eastman single and multi-shot camera. GPNL collar survey methods are unknown, down hole surveys were by Eastman single shot camera. Metex used Spectrum Surveys of Kalgoorlie to layout the collar locations and survey the collar position once completed using established control points around the old mine site. Drill core was orientated using a spear system and either an Eastman single shot camera or down hole gyroscope tool. FML drill collars were surveyed after completion, using a DGPS instrument. Drill core was oriented by the drilling contractor using an Ezy-mark system. A north- seeking gyroscope tool was used to survey down hole. Holes were surveyed open- hole. Otherwise a single shot Eastman camera downhole survey was used. All coordinates and bearings use the MGA94 Zone 51 grid system. FML utilises Landgate sourced regional topographic maps and contours as well as internally produced survey pick-ups produced by the mining survey teams utilising DGPS base station instruments.
Data spacing and distribution	Drill spacing along the Lancefield trend is irregular, varying from 25m x 50m in the upper middle section to more than 150m x 250m to the south. Numerous "fans" have been drilled from underground drive shafts.
Orientation of data in relation to geological structure	 Drilling was designed based on known geological models, field mapping, verified historical data and cross-sectional interpretation. Drill holes were oriented at right angles to strike of deposit, with dip optimised for drill capabilities and the dip of the ore body.
Sample security	 All samples were reconciled against the sample submission with any omissions or variations reported to FML. All samples were bagged in a tied numbered calico bag, grouped into green plastic bags. The bags were placed into bulka bags with a sample submission sheet and kept within the Laverton yard until ready for transport to Kalgoorlie by transport courier. Historic sample security is not recorded.
Audits or reviews	 After Metex Resources acquired the WMC data, a thorough data validation of the WMC GEOVIA SurpacTM database against raw data hard copy information and Eastman photographic survey shots was conducted in the mid 1990's. Focus Minerals has purchased the Metex validated database and associated hard copies as part of the Lancefield project acquisition.

Section 2 Reporting of Exploration Results (Criteria listed in the preceding section also apply to this section.)

tomena instituti in the preceding section also apply to ans section.							
Criteria	Commentary						
Mineral tenement and land tenure status	 All exploration was conducted on tenements 100% owned by FML or its subsidiary companies Focus Operations Pty Ltd. All tenements are in good standing. Various royalties may be in place as documented in the FML Annual Report 2016 FML holds Native Title agreements with traditional Landowners. 						
Exploration done by other parties	 The Lancefield Project has been historically mined with the bulk of the development by WMC from 1980 to 1994 (when the mine closed) by both underground and surface mining of the Main Lode and West Lode Horizons. The area consists of numerous open pits and underground workings, including Lancefield open pit, South Lancefield open pit and Telegraph open pit. Underground activities focused on the Main Lode and Main Lode Deeps which were mined using both shaft and decline access. The maximum vertical depth of development is 830m. Production figures quoted in the Metex Resources NL, Annual Technical Report 1996-1997, state "WMC produced 3.72Mt @ 6.59g/t, with 3.23Mt @ 6.77g/t produced from the underground mining operation" (Johnson, 1997). GPNL held the prospecting tenements and later combined Mining Lease for the ground immediately south of WMC's operating Lancefield mine. From mid-1987 to mid-1988 they successfully completed 3 diamond holes designed to test the down dip extension of the Lancefield Deeps. 						

Criteria C	Commentary							
Onteria		tonomonts from WMC in November 1005 and drilled 2	doop diamond halas /	with 2 "dayabtar" balas				
	Metex acquired the Lancefield tenements from WMC in November 1995 and drilled 3 deep diamond holes (with 2 "daughter" holes wedged off the main hole traces).							
	 The ground was subsequently acquired by Crescent Gold NL in June 2010 before being taken over by Focus Mineral October 2012. 							
Geology	 The geological setting at Lance 	field is that of a basal komatiite overlain by tholeiitic bas	alt and gabbro units w	ith carbonaceous shale				
.,	interflow sediments. The ultramafic / mafic package is overlain by a sedimentary pile, commencing with a basal conglomeratic unit that is overlain by pelitic and arenaceous sediments.							
		eurs within stacked interflow sediments within the mafic	unite. The codiments o	nnoar to have localised				
,				• •				
		pecoming silicified and sulphidic. Grade and alteration						
	_	terflow sediments being increasingly carbonaceous to		ade shoots are spatially				
		in turn relate to syenite intrusives in the ultramafic footw						
١٠		by silica - carbonate - sulphide replacement of carbona						
	=	arsenopyrite – pyrrhotite – pyrite – quartz – carbonate –		-				
		strong As - Ag correlation with gold (also Cu - Zn in the						
	=	de occluded elemental grains within arsenopyrite. To t	he north, the lode style	e has less arsenopyrite				
	and is more banded. The high-g	grade shoots becoming more localised.						
Drill hole	 Historic Lancefield drilling inforr 	mation has been validated against publicly available Wi	AMEX reports. Not all	drill holes can be found				
information	referenced in the WAMEX repo	rts. However, cross-checking of original drill surveys an	d paper geology logs	was verified against the				
		are in the sub-inferred or mined out part of the resource						
		istent with surrounding drill hole information.						
	,	3 · · · · · · · · · · · · · · · · · · ·						
			WAMEX Report A-					
	Company	Drill Hole Number	Number	Report Date				
	Metex Resources NL	MLD01, MLD01W1, MLD02, NMLD01, NMLD01W1	48547	January 1996				
	Golden Plateau NL	GLD1	23426	1989				
		GLD2, GLD3	28728	1989				
	Western Mining	LFD069	16961	January 1986				
	Corporation Ltd	LFD072, 074, 074W1, 074W2, 083	19483	June 1986				
		LFD075, 076, 081, 082, 084AW1, 084AW2, 085A, 086, 087, 088, 088W1, 088W2, 089A, 090A, 092, 092W1, 093, 094, 096, 096W1	22649	January 1988				
		LFD097, 098; LFU050-02, 960-01, 960-02, 960-03, 960-04, 960-05, 960-06, 960-07, 960-08	32929	March 1991				
		ASSAY ONLY: LFU050-01, 056- 05, 056-06, 056- 07, 233-01, 233- 02, 233-03, 248-01, 248-02,						
		LFU941-01, 941-02, 941-03, 941-04, 942-01, 942-	42284	September 1994				
		02, 942-03, 942-04, 942-05, 942-06						
		ASSAY ONLY: LFU9801-01, 9801-02, 9801-03,						
		102-01, 102-02, 110-01, 110-02, 170-03, 233-04,						
		233-05, 876-03, 876-04, 876-05						
	 FML drilled 5 holes at Lance 	field in mid-2017, 2 RC holes and 3 RC/DD holes	of these 2 RC/DD					
	holes (LFRD012, 014) were							
	Drill Hole Number	ASX Release Title	ASX Release Date					
	LFRC015, 026	0 5 111 11	25-Jul-17					
	LFRD012, 013, 014	Operational Update						
Data aggregation	 Mineralised intersections are re 	ported at a 2g/t Au cut-off, composited to 1m for diamor	nd holes					
methods								
Relationship	 Holes were drilled orthogonal to 	mineralisation as much as possible, however the exact	relationship between	intercept width and true				
between	width cannot be estimated exac	tly in all cases.						
mineralisation								
widths and								
intercept lengths								
Diagrams	 Refer to Figures and Tables in b 	body of the release.						
Balanced	Historic drill hole results availab	le on WAMEX.						
reporting		in the previous drill hole information table.						
	There is no other material explo	<u>'</u>						
substantive	- mere is no otner material explo	ναιιοπ ααια το τ ο ροττ.						
Jungallare								
exploration data								
exploration data Further work	The company is further reviewir	ag the exploration results						

Criteria	Commentary
Database	FML data was geologically logged electronically, collar and downhole surveys were also received electronically as was the laboratory
integrity	analysis results. These electronic files were loaded into an acQuire database by the company in-house Database Administrator.
	FML's database is a Microsoft SQL Server database (acQuire), which is case sensitive, relational and normalised to the Third Normal
	Form. As a result of normalisation, the following data integrity categories exist:
	Entity Integrity: no duplicate rows in a table, eliminated redundancy and chance of error.
	Domain Integrity: Enforces valid entries for a given column by restricting the type, the format or a range of values. Poferential laterative Power connect by deleted which are used by other records.
	 Referential Integrity: Rows cannot be deleted which are used by other records. User-Defined Integrity: business rules enforced by acQuire and validation codes set up by FML.
	Additionally, in-house validation scripts are routinely run in acQuire on FML's database and they include the following checks:
	Missing collar information
	Missing logging, sampling, downhole survey data and hole diameter
	 Overlapping intervals in geological logging, sampling, down hole surveys Checks for character data in numeric fields
	Data extracted from the database were validated visually in GEOVIA SurpacTM software and ARANZ Geo Leapfrog software. Also,
	when loading the data any errors regarding missing values and overlaps are highlighted.
	Historic data has been validated against WAMEX reports where possible.
Site visits	Jeff Ion, the Competent Person for Sections 1 and 2 of Table 1 is FML's Principal Geologist via his contracting company Jeffrey Geo
	Pty Ltd, conducts regular site visits.
	Hannah Kosovich is FML's Resource Geologist and has visited Lancefield in 2014.
	Michael Job, the Competent Person for Section 3 of Table 1 is Principal Consultant with Cube Consulting, an independent mineral
	industry consulting group.
Geological	All available drill hole and historic mining data was used to guide the geological interpretation of the mineralisation.
interpretation	The mineralised geological interpretation was digitized in GEOVIA SurpacTM software on a section by section basis. An approximate
1	2g/t cut-off was used, infrequently sub 2g/t samples were included for continuity. The logging of sediments and sulphides also guided
	the interpretation.
	Minor deviation only of the lode geometry was noticed between drill holes along strike and down-dip. This is evident by the old WMC underground development.
Dimensions	The entire Lancefield deposit strikes NS with a total strike length of over 1.5km. The main lode of mineralisation has been modelled to
	approximately 1.5km below surface, the bulk of the main lode sits approx. 300m beneath surface. Mineralisation has an average width of 3-5m.
Estimation and	Diamond holes were used in the estimation. In total 108 holes were used in the estimate; 50 diamond holes, most with an RC pre-
modelling techniques	collar (RCDD) and 58 Underground diamond holes. The two "daughter" holes drilled by Metex have been counted as separate drill
loomiquoo	holes as they have unique hole id's. This includes 2 of the 5 holes discussed in section 2 of this table.
	The drill hole samples were composited to 1m within each domain. This is the dominant sampling interval. All density be added to a drill be
	 All domain boundaries were considered "hard" boundaries and no drill hole information were used by another domain in the estimation. Composited assay values of each domain were exported to a text file (.csv) and imported into Snowden Supervisor for geostatistical
	analysis.
	A review of histograms, probability plots and mean/variance plots for the main lode domain revealed no significant outlier sample
	values. Therefore, no top-capping of the gold values was undertaken in the estimation.
	The data was declustered in Supervisor using a cell weighted approach.
	Variograms were modelled in Supervisor. CEOVA Compact TM Coff was a variety and facility and and alliance and alliance and alliance and and alliance and and alliance and and alliance and alliance and alliance and and alliance and al
	GEOVIA SurpacTM Software was used for the estimation and modelling process. The model was created in GDA 94 grid co-ordinates. Block sizes for the model were
	• 12.5m in Y, 6m in X and 6m in Z direction. Sub celling of the parent blocks was permitted to 3.125m in the Y direction, 1.5m in the X
	direction and 1.5m in the Z direction. Sub-blocking was used to best fill the wireframes and inherit the grade of the parent block. No
	rotation was applied to the orientation of the blocks.
	Block size is approximately ½ of the average drill hole spacing along strike and across strike to best fill the wireframe volume.
	An Ordinary Kriging (OK) estimation technique was selected and used the variograms modelled in Supervisor.
	Minimum (8) and maximum (20) samples were selected based on a Kriging Neighbourhood analysis in Supervisor.
	An elliptical search was used based on range/ratio of the Variograms.
	Three search passes were run in order to fill the block model with estimated Au values. After each search pass the search range was doubted and in the third search pass were run in order to fill the block model with estimated Au values. After each search pass the search range was doubted and in the third search pass the search range was
	doubled and in the third search pass minimum number of samples was decreased.
	The estimate was validated by a number of methods. An initial visual review was done by comparing estimated blocks and raw drill holes.
	holes. Tonnage weighted mean grades were compared for the Main Lode with no major differences.
	Swath plots of drill hole values and estimated Au grades by northing and RL were run for the main domain and showed that the
	estimated grades honoured the trend of the drilling data.
Moisture	Tonnages are estimated on a dry basis.
Cut-off	The Mineral Resources for Lancefield have been reported above a 4g/t Au cut-off.
-u. 011	- The Milheral Nesources for Lancelleid have been reported above a 49/t Au cut-off.

Criteria	Commentary
parameters	
Mining factors or assumptions	The majority of the Lancefield deposit would most likely be mined by underground mine methods.
Metallurgical factors or assumptions	Metex commissioned metallurgical studies and the production records of WMC document plant recoveries. The Main Lode of Lancefield is known to be sulphide refractory.
Environmental factors or assumptions	Lancefield deposit occurs in a historic mining centre with both open cut and underground workings in the area.
Bulk density	Specific gravity measurements were taken on select core samples during the Metex deep diamond drilling program of 1995, (Little, 1996). Based on the test work an average SG for the Main Lode of 2.86 has been applied to the block model.
Classification	Mineral Resources have been classified as Inferred.
Audits or reviews	Cube Consulting worked with and reviewed/critiqued FML's work on the geological interpretation, estimation methodology and parameters, and estimate validation. Michael Job from Cube Consulting is satisfied to act as one of the Competent Persons for the Mineral Resource estimate.
Discussion of relative accuracy/ confidence	This is addressed in the relevant paragraph on Classification above. The Mineral Resource relates to global tonnage and grade estimates.