

CHALLENGER GOLD COMPLETES TOLL PROCESSING AGREEMENT AND STRATEGIC PLACEMENT

- Final binding Toll Processing Agreement executed.
- Guaranteed annual processing capacity of 150,000 tonnes for three years.
- Working capital support for mining, ore transport, and processing¹ of Hualilan ore.
- Final binding agreements for \$6.6M strategic placement completed.

Challenger Gold (ASX: CEL) ("CEL" or the "Company") is pleased to announce the execution of the definitive Binding Toll Processing Agreement (the "Agreement") with Casposo Argentina Mining Limited the operator of the Casposo treatment plant located in San Juan Argentina (the "Toll Mill Operator"), for toll processing ore from the Hualilan Gold Project.

The agreement guarantees processing capacity of 150,000 tonnes per annum over three years, with a total secured capacity of 450,000 tonnes. It also provides working capital support¹ to cover mining, trucking and processing costs until CEL begins receiving cash flow from the toll milling operations.

Additionally, final binding Agreements for the \$6.6 million strategic placement to an entity controlled by Eduardo Elsztain, have been completed with the placement expected to settle on 2 January 2025. The placement will be completed at a price of 4.5 cents per share² as previously announced. The placement proceeds will be used to fund the US\$2m upfront access payment under the Toll Processing Agreement, the costs associated with preparation for Toll Milling, and general working capital purposes.

¹defined in the Binding Agreement as "the Mill Operator has undertaken to use best commercial efforts to finance working capital - directly or through third parties"

² refer to page 3 below for more detail on placement terms



FINALISED TOLL PROCESSING TERMS

The definitive Toll Processing Agreement establishes favorable toll processing terms designed to align with the Hualilan Project's economics, including processing at cost, a base toll processing fee, and an incentive structure tied to gold recovery. Highlights include:

- Base Toll Processing Fee: Processing at cost including processing costs uplift for additional reagent consumption which is typical of an agreement of this type.
- Monthly Access Fee: US\$8.80/t, with a monthly minimum payment of US\$110,000 from the start of Toll Milling.
- Upfront Payment: US\$2M, with US\$1M deferred until the second year, equivalent to US\$6.67/t
- **Performance Fee:** Incentive-based fee of 20-30% of milling costs depending on recovery rates, expected to range between US\$12-18/t.

Table 1 - Recovery Based Performance Fee

Gold Recovery (%)	70%-80%	+ 80%-85%	+85%	
Margin over Processing Costs	20%	25%	30%	

The agreement includes a safeguard, the upfront payment is refundable if the plant is not operating by 31 July 2025 unless the delay is caused by matters related to the mining or extraction of mineralised material from the Hualilan Project or matters otherwise beyond Casposo's control or; if additional testwork indicates that the gold recovery of Hualilan material via Toll Processing will be below 70%.

Funding

The Toll Mill Operator recently completed a successful financing of US\$7 million to refurbish the Casposo Plant for toll milling material from Challenger's Hualilan Project (refer to ASX release dated 23 December 2024).

The funding was secured by Austral Gold's subsidiary, Casposo Argentina Mining Ltd. (Casposo), through a US\$7 million secured loan from Banco San Juan S.A. in Argentina. These funds will be allocated to refurbish the Casposo Plant, supporting the Agreement with CEL to process mineralised material from CEL's Hualilan Project.

The Mill Operator will use its best commercial efforts to provide working capital funding - directly or through third parties - to cover CEL's costs of mining, ore transport, and processing under an approved budget.



Operational Details

Initial processing will operate in 90-day campaign cycles over a 12-month period, unless the Technical Committee decides to extend such a term. The Technical Committee, with equal representation from both parties, will evaluate transitioning to continuous processing after the first year based on operational performance.

CEL retains full ownership of ore and all resulting products throughout the process, with established protocols for gold-in-circuit measurement.

CEL representatives at the plant may request adjustments to treatment rate and metallurgical parameters as needed. Additional terms remain unchanged from the 5 December 2024 ASX Release.

STRATEGIC PLACEMENT

Final binding Agreements for the completion of the \$6.6m Strategic Placement to a part of the Elsztain Group (the "Purchaser") have been executed. The terms of the placement are as previously announced with the placement expected to be completed on 2 January 2025.

On the Closing Date the Purchaser will acquire from the Company 147,726,678 ordinary shares in the capital of the Company (each, a "Share") at a price of 4.5 cents, each Share with an attached ordinary share purchase warrant of the Company. The share purchase warrants will have a period of two (2) years from the Closing Date at a strike price equivalent to the Placement Price plus 10% during the first 12 months then the Placement Price plus 20% after 12 months.

The warrants may not be exercised to the extent that the issue of the underlying CEL's shares would cause the Purchaser (alone or together with any affiliates and/or any other person with whom the Purchaser may have an agreement or arrangement providing voting power) to exceed voting power in excess of 19.99% in CEL unless CEL shareholders resolve to approve the issue in accordance with section 611 item 7 of the Corporations Act or another exception contained in section 611 of the Corporations Act from the prohibitions in section 606 of the Corporations Act is available and applies to the issue. The Purchaser acknowledges that CEL is not proposing to seek shareholder approval under item 7 of section 611 of the Corporations Act to allow the Purchaser to exceed 19.99% voting power.

The Elsztain Group is controlled by Mr Eduardo Elsztain. Mr Elsztain is a prominent Argentine businessman who has built a huge portfolio of real assets in the last 3 decades, including rental and mixed-use properties in Argentina and farmland in Latin America. Mr Elsztain also has extensive interests in mining both inside and outside Argentina.

This ASX release was approved by the Board of Directors.



For further information contact:

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Previous announcements referred to in this release include:

The Mineral Resource Estimate for the Hualilan Gold Project was first announced to the ASX on 1 June 2022 and updated 29 March 2023. The Mineral Resource Estimate for the El Guayabo Project was first announced to the ASX on 14 June 2023. The Company confirms it is not aware of any information or assumptions that materially impacts the information included in that announcement and that the material assumptions and technical parameters underpinning the Mineral Resource Estimate continue to apply and have not materially changed.

ADDITIONAL INFORMATION

COMPETENT PERSON STATEMENT – EXPLORATION RESULTS AND MINERAL RESOURCES

The information that relates to sampling techniques and data, exploration results, geological interpretation and Mineral Resource Estimate has been compiled Dr Stuart Munroe, BSc (Hons), PhD (Structural Geology), GDip (AppFin&Inv) who is a full-time employee of the Company. Dr Munroe is a Member of the AusIMM. Dr Munroe has over 20 years experience in the mining and metals industry and qualifies as a Competent Person as defined in the JORC Code (2012).

Dr Munroe has sufficient experience of relevance to the styles of mineralisation and the types of deposits under consideration, and to the activities undertaken, to qualify as a Competent Person as defined in the 2012 Edition of the Joint Ore Reserves Committee (JORC) Australasian Code for Reporting of Exploration Results and Mineral Resources. Dr Munroe consents to the inclusion in this report of the matters based on information in the form and context in which it appears. The Australian Securities Exchange has not reviewed and does not accept responsibility for the accuracy or adequacy of this release.

FORWARD LOOKING STATEMENTS

The announcement may contain certain forward-looking statements. Words 'anticipate', 'believe', 'expect', 'forecast', 'estimate', 'likely', 'intend', 'should', 'could', 'may', 'target', 'plan', 'potential' and other similar expressions are intended to identify forward-looking statements. Indication of, and guidance on, future costings, earnings and financial position and performance are also forward-looking statements.

Such forward looking statements are not guarantees of future performance, and involve known and unknown risks, uncertainties and other factors, many of which are beyond the control of Challenger Gold Ltd, its officers, employees, agents and associates, which may cause actual results to differ materially from those expressed of implied in such forward-looking statements. Actual results, performance, or outcomes may differ materially from any projections or forward-looking statements or the assumptions on which those statements are based.

You should not place any undue reliance on forward-looking statements and neither. Challenger nor its directors, officers, employees, servants or agents assume any responsibility to update such information. The stated Production Targets are based on the Company's current expectations of future results or events and should not be relied upon by investors when making investment decisions. Further evaluation work and appropriate studies are required to establish sufficient confidence that this target will be met.

Challenger Gold Limited ACN 123 591 382 ASX: CEL **Issued Capital** 1,532m shares 126.7m options 52.1m perf rights **Australian Registered Office** Level 1 100 Havelock Street West Perth WA 6005 Directors
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Financial numbers, unless stated as final, are provisional and subject to change when final grades, weight and pricing are agreed under the terms of the offtake agreement. Figures in this announcement may not sum due to rounding.

The Company confirms that it is not aware of any new information or data that materially affects the information included in the original market announcements and that all material assumptions and technical parameters underpinning the estimates in the relevant original market announcements continue to apply and have not materially changed. The Company confirms that the form and context in which the Competent Person's findings are presented have not been materially modified from the original market announcements.

HUALILAN GOLD PROJECT MRE AND SCOPING STUDY

All references to the Scoping Study and its outcomes in this announcement relate to the ASX Announcement of 8 November 2023 'Hualilan Gold Project Scoping Study'. Please refer to that announcement for full details and supporting documentation.

Table 2: Hualilan Hold Project Mineral Resource Estimate (March 2023)

Domain	Category	Mt	Au (g/t)	Ag (g/t)	Zn (%)	Pb (%)	AuEq (g/t)	AuEq (Mozs)
US\$1800 optimised shell > 0.30 ppm AuEq	Indicated	45.5	1.0	5.1	0.38	0.06	1.3	1.9
	Inferred	9.6	1.1	7.3	0.43	0.06	1.4	0.44
Below US\$1800 shell >1.0ppm AuEq	Indicated	2.7	2.0	9.0	0.89	0.05	2.5	0.22
	Inferred	2.8	2.1	12.4	1.1	0.07	2.8	0.24
Total		60.6	1.1	6.0	0.4	0.06	1.4	2.8

Note: Some rounding errors may be present

¹ Gold Equivalent (AuEq) values - Requirements under the JORC Code

- Assumed commodity prices for the calculation of AuEq is Au US\$1900 Oz, Ag US\$24 Oz, Zn US\$4,000/t, Pb US\$2000/t
- Metallurgical recoveries are estimated to be Au (95%), Ag (91%), Zn (67%) Pb (58%) across all ore types (see JORC Table 1 Section 3 Metallurgical assumptions) based on metallurgical test work.
- The formula used: $AuEq(g/t) = Au(g/t) + [Ag(g/t) \times 0.012106] + [Zn(%) \times 0.46204] + [Pb(%) \times 0.19961]$
- CEL confirms that it is the Company's opinion that all the elements included in the metal equivalents calculation have a reasonable potential to be recovered and sold.

JORC Code, 2012 Edition – Table 1 report template Section 1 Sampling Techniques and Data -Hualilan Project

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

Criteria	JORC Code explanation	Commentary
Sampling techniques	 Nature and quality of sampling (eg cut channels, random chips, or specific specialised industry 	Diamond core (HQ3 and NQ3) was cut longitudinally on site using a diamond saw or split using a hand operated hydraulic core sampling splitter. Samples lengths are generally from 0.5m to 2.0m in length (average 1.74m). Sample lengths are selected according to lithology, alteration, and mineralization contacts.
	standard measurement tools appropriate to the minerals under investigation, such as down hole	For reverse circulation (RC) drilling, 2-4 kg sub-samples from each 1m drilled were collected from a face sample recovery cyclone mounted on the drill machine.
	gamma sondes, or handheld XRF instruments, etc). These examples should not be taken as	Channel samples are cut into underground or surface outcrop using a hand-held diamond edged cutting tool. Parallel saw cuts 3-5cm apart are cut 2-4cm deep into the rock which allows for the extraction of a representative sample using a hammer and chisel. The sample is collected onto a plastic mat and collected into a sample bag.
	limiting the broad meaning of sampling. - Include reference to measures taken to ensure sample representivity and the appropriate	Core, RC and channel samples were crushed to approximately 85% passing 2mm. A 500g or a 1 kg sub-sample was taken and pulverized to 85% passing 75 μ m. A 50g charge was analysed for Au by fire assay with AA determination. Where the fire assay grade is > 10 g/t gold, a 50g charge was analysed for Au by Fire assay with gravimetric determination.
	 calibration of any measurement tools or systems used. Aspects of the determination of mineralisation that are Material to the Public Report. In cases where 'industry standard' work has been done this would be relatively simple (eg 'reverse circulation drilling was used to obtain 1 m samples from which 3 kg was pulverised to produce a 30 g charge for fire assay'). In other cases, more explanation may be required, such as where there is coarse gold that has inherent sampling problems. Unusual commodities or mineralisation types (eg 	A 10g charge was analysed for at least 48 elements by 4-acid digest and ICP-MS determination. Elements determined include Ag, As, Ba, Be, Bi, Ca, Ce, Co, Cr, Cs, Cu, Fe, Ga, Ge, Hf, In, K, La, Li, Mg, Mn, Mo, Na, Nb, Ni, P, Pb, Rb, Re, S, Sb Sc, Se, Sn, Sr, Ta, Te, Th, Ti, Tl, U, V, W, Y, Zn and Zr. For Ag > 100 g/t, Zn, Pb and Cu > 10,000 ppm and S > 10%, overlimit analysis was done by the same method using a different calibration.
		Unused pulps are returned from the laboratory to the Project and stored in a secure location, so they are available for any further analyses. Remaining drill core is stored undercover for future use if required.
		Visible gold observed has been observed in only 1 drill core sample only. Coarse gold is not likely to result in sample bias.
		Historic Data: There is little information provided by previous explorers to detail sampling techniques. Selected drill core was cut with a diamond saw longitudinally and one half submitted for assay. Assay was generally done for Au. In some drill campaigns, Ag and Zn were also analysed. There is limited multielement data available. No information is available for RC drill techniques and sampling.
	submarine nodules) may warrant disclosure of detailed information.	

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JORC Code explanation Criteria Commentary CEL drilling of HQ3 core (triple tube) was done using various truck and track mounted drill machines that are **Drilling techniques** Drill type (eg core, reverse operated by various drilling contractors based in Mendoza and San Juan. The core has not been oriented as the circulation, open-hole hammer, rock is commonly too broken to allow accurate core orientation. rotary air blast, auger, Bangka, sonic, etc) and details (ea core CEL drilling of reverse circulation (RC) drill holes was done using a track-mounted LM650 universal drill rig set diameter, triple or standard tube. up for reverse circulation drilling. Drilling was done using a 5.25 inch hammer bit. depth of diamond tails, face-Collar details for historic drill holes, DD drill holes, RC drill holes completed by CEL that are used in the resource sampling bit or other type, estimate are detailed in CEL ASX releases: whether core is oriented and if 1 June 2022 (Maiden MRE): https://announcements.asx.com.au/asxpdf/20220601/pdf/459ifk8g7x2mty.pdf so, by what method, etc). and 29 March 2023 (MRE update): https://announcements.asx.com.au/asxpdf/20230329/pdf/45n49ilm02grm1.pdf Collar locations for drill holes are surveyed using DGPS. Three DD holes and 3 RC holes have hand-held GPS collar survevs. Historic Data: Historic drill hole data is archival, data cross checked with drill logs and available plans and sections where available. Collar locations have been checked by CEL using differential GPS (DGPS) to verify if the site coincides with a marked collar, tagged drill site or likely drill pad location. In most cases the drill collars coincide with historic drill site, some of which (but not all) are tagged. The collar check surveys were reported in POSGAR (2007) projection and converted to WGS84. Method of recording and Drill core is placed into wooden boxes by the drillers and depth marks are indicated on wooden blocks at the end Drill sample recovery of each run. These depths are reconciled by CEL geologists when measuring core recovery and assessing core assessing core and chip sample loss. Triple tube drilling has been being done by CEL to maximise core recovery. recoveries and results assessed. Measures taken to maximise 761 CEL diamond drill holes completed have been used for the CEL resource estimate. Some of these sample recovery and ensure holes are located outside the resource area. representative nature of the Total drilled is 224,180.60 metres, including cover drilled of 22,041.30 metres (9.8 %). samples. Of the remaining 202,139.30 metres of bedrock drilled, core recovery is 96.8%. Whether a relationship exists RC sub-samples are collected from a rotary splitter mounted to the face sample recovery cyclone. A 2-4 kg subbetween sample recovery and samples is collected for each metre of RC drilling. Duplicate samples are taken at the rate of I every 25-30 grade and whether sample bias samples using a riffle splitter to split out a 2-4 kg sub-sample. The whole sample recovered is weighed to may have occurred due to measure sample recovery and consistency in sampling. preferential loss/gain of 37 CEL RC drill holes have been used in the CEL resource estimate. fine/coarse material. Total metres drilled is 2.923m. Cover drilled is 511 m (17.5%) Channel samples have been weighed to ensure a consistency between sample lengths and weights. The channel samples are collected from saw-cut channels and the whole sample is collected for analysis. There is no correlation between sample length and assay values. 193 surface and underground channels have been used in the CEL resource estimate.

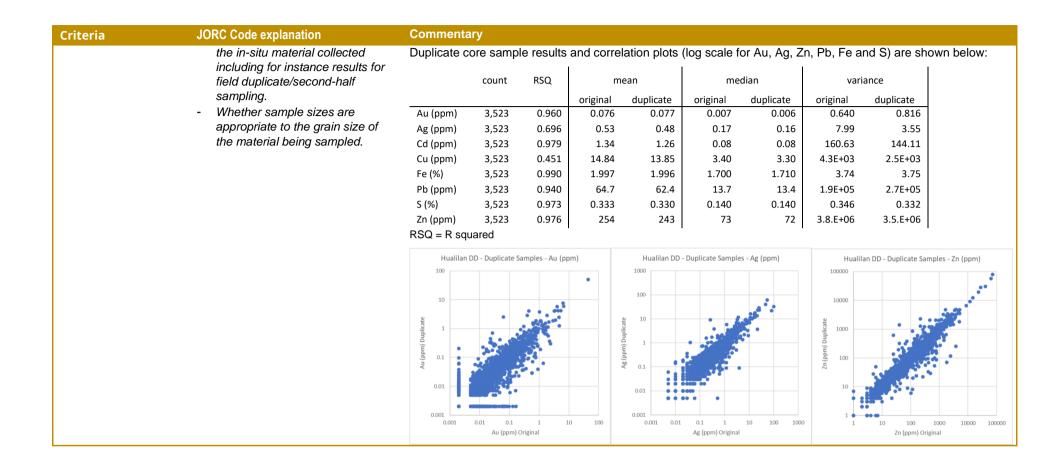
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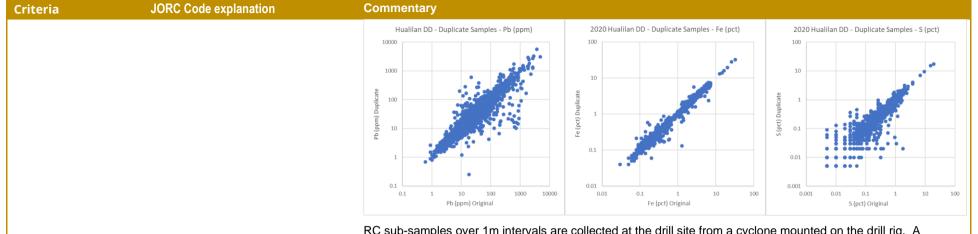
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Criteria	JORC Code explanation	Commentary
		Channels total 2597.70 metres in length. The average weight per metre sampled is 3.7 kg/m which is adequate for the rock being sampled and compares well with the expected weight for ½ cut HQ3 drill core of 4.1 kg/m. A possible relationship has been observed in historic drilling between sample recovery and Au Ag or Zn values whereby low recoveries have resulted lower reported values. Historic core recovery data is incomplete. Core recovery is influenced by the intensity of natural fracturing in the rock. A positive correlation between recovery and RQD has been observed. The fracturing is generally post mineral and not directly associated with the mineralisation.
Logging	 Whether core and chip samples have been geologically and geotechnically logged to a level of detail to support appropriate Mineral Resource estimation mining studies and metallurgical studies. Whether logging is qualitative or quantitative in nature. Core (or costean channel etc) photography. The total length and percentage of the relevant intersections logged. 	For CEL drilling, all the core (100%) is photographed and logged for recovery, RQD, weathering, lithology, alteration, mineralization, and structure to a level that is suitable for geological modelling, Mineral Resource Estimation and metallurgical test work. RC drill chips are logged for geology, alteration and mineralisation to a level that is suitable for geological modelling resource estimation and metallurgical test work. Where possible logging is quantitative. Geological logging is done into MS Excel in a format that can readily be cross-checked and is back-up transferred to a secure, offsite, cloud-based database which holds all drill hole logging sample and assay data. No specialist geotechnical logging has been undertaken. Detailed logs are available for most of the historical drilling. Some logs have not been recovered. No core photographs from the historic drilling have been found. No drill core has survived due to poor storage and neglect. No historic RC sample chips have been found.
Sub-sampling techniques and sample preparation	 If core whether cut or sawn and whether quarter half or all core taken. If non-core whether riffled tube sampled rotary split etc and whether sampled wet or dry. For all sample types the nature quality and appropriateness of the sample preparation technique. Quality control procedures adopted for all sub-sampling stages to maximise representivity of samples. Measures taken to ensure that the sampling is representative of 	CEL samples have been submitted to the MSA laboratory in San Juan, the ALS laboratory in Mendoza and the former SGS laboratory in San Juan for sample preparation. The sample preparation technique is considered appropriate for the style of mineralization present in the Project. Sample sizes are appropriate for the mineralisation style and grain size of the deposit. Sample intervals are selected based on lithology, alteration, and mineralization boundaries. Representative samples of all of the core are selected. Sample length averages 1.74m. Second-half core or ¼ core samples have been submitted for a mineralised interval in 1 drill hole only and for some metallurgical samples. The second half of the core samples has been retained in the core trays for future reference. Competent drill core is cut longitudinally using a diamond saw for sampling of ½ the core. Softer core is split using a wide blade chisel or a manual core split press. The geologist logging the core, marks where the saw cut or split is to be made to ensure half-core sample representivity. From GNDD073 and later holes, duplicate core samples consisting of two ¼ core samples over the same interval have been collected approximately every 30-50m drilled.

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RC sub-samples over 1m intervals are collected at the drill site from a cyclone mounted on the drill rig. A duplicate RC sample is collected for every 25-30m drilled.

The duplicate RC sample results and correlation plots (log scale for Au, Ag, Zn, Pb, Fe and S) are shown below:

	count	RSQ	mean		me	dian	variance		
			original	duplicate	original	duplicate	original	duplicate	
Au (ppm)	85	0.799	0.101	0.140	0.017	0.016	0.041	0.115	
Ag (ppm)	85	0.691	1.74	2.43	0.59	0.58	13.59	64.29	
Cd (ppm)	85	0.989	15.51	16.34	0.41	0.44	4189	4737	
Cu (ppm)	85	0.975	47.74	53.86	5.80	5.70	2.4E+04	3.1E+04	
Fe (%)	85	0.997	1.470	1.503	0.450	0.410	7.6	7.6	
Pb (ppm)	85	0.887	296.0	350.6	26.3	32.4	6.0E+05	7.4E+05	
S (%)	85	0.972	0.113	0.126	0.020	0.020	0.046	0.062	
Zn (ppm)	. 85	0.977	3399	3234	158	177	2.5.E+08	2.1.E+08	
RSQ = R squ	ıared								

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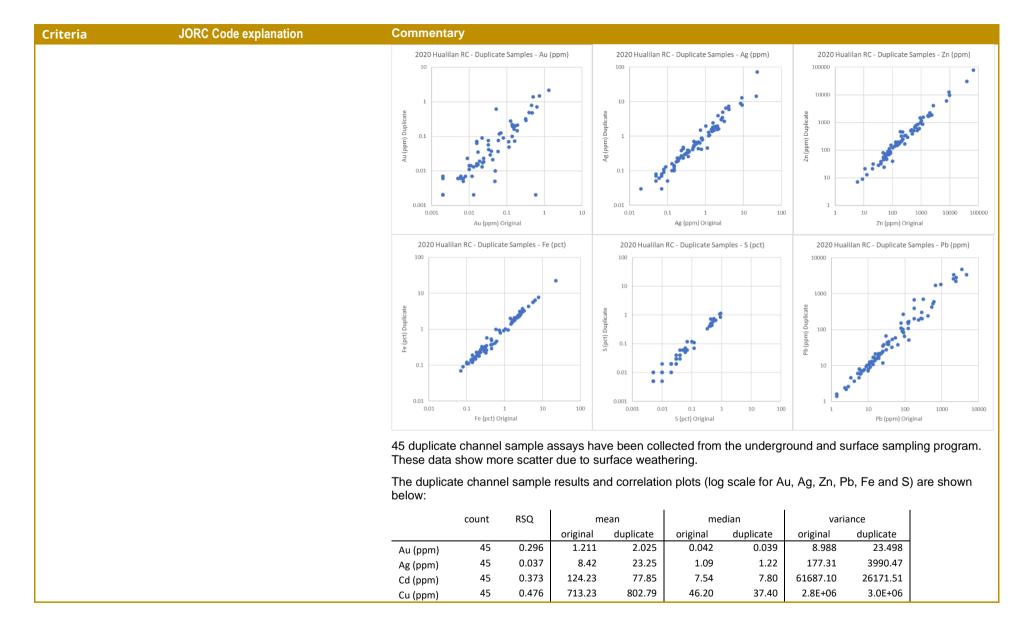
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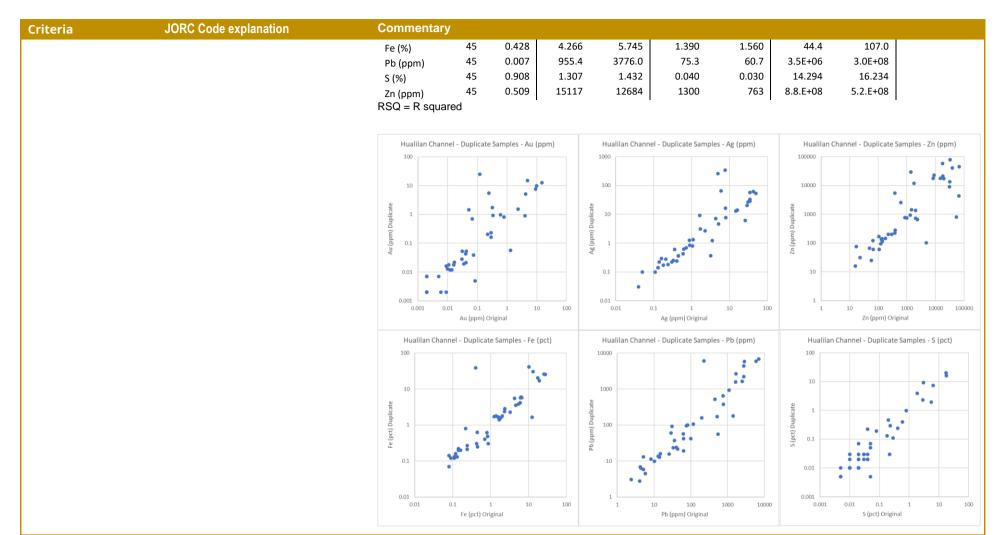
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Quality of assay data and laboratory tests

The nature quality and appropriateness of the assaying and laboratory procedures used and whether the technique is

The MSA laboratory used for sample preparation in San Juan was inspected by Stuart Munroe (Exploration Manager) and Sergio Rotondo (CEL Director) prior to any samples being submitted. The laboratory has been visited and revied most recently by Stuart Munroe (Exploration Manager) in May 2022. The laboratory procedures are consistent with international best practice and are suitable for samples from the Project. The SGS laboratory in San Juan and the ALS laboratory in Mendoza has not yet been inspected by CEL

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LS laboratory in Mendoza has not yet been inspected by C

Contact
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Criteria

JORC Code explanation

considered partial or total. represe

 For geophysical tools spectrometers handheld XRF instruments etc the parameters used in determining the analysis including instrument make and model reading times calibrations

factors applied and their

derivation etc.

Nature of quality control procedures adopted (eg standards blanks duplicates external laboratory checks) and whether acceptable levels of accuracy (i.e. lack of bias) and precision have been established.

Commentary

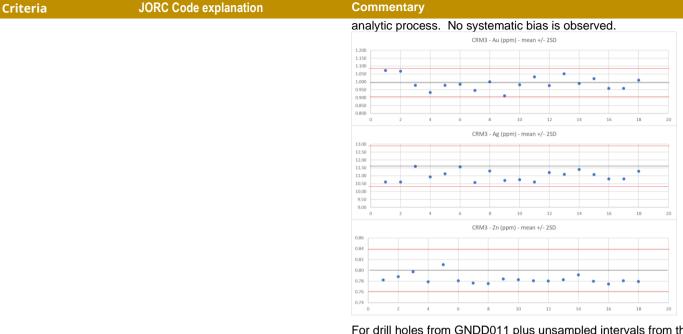
representatives due to COVID-19 restrictions. Each laboratory presents internal laboratory standards for each job to gauge precision and accuracy of assays reported.

CEL have used two different blank samples, submitted with drill core and subjected to the same preparation and assay as the core samples, RC sub-samples and channel samples. The blank samples are sourced from surface gravels in the Las Flores area of San Juan and from a commercial dolomite quarry near San Juan. In both cases the blank material is commonly for construction. Commonly, the blank samples are strategically placed in the sample sequence immediately after samples that were suspected of containing higher grade Au, Ag, S or base metals to test the lab preparation and contamination procedures. The values received from the blank samples suggest only rare cross contamination of samples during sample preparation.

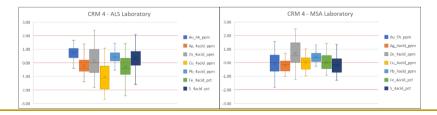


For GNDD001 – GNDD010 samples analysed by MSA in 2019, three different Certified Standard Reference pulp samples (CRM) with known values for Au Ag Pb Cu and Zn were submitted with samples of drill core to test the precision and accuracy of the analytic procedures MSA laboratory in Canada. 26 reference analyses were analysed in the samples submitted in 2019. The standards demonstrate suitable precision and accuracy of the

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For drill holes from GNDD011 plus unsampled intervals from the 2019 drilling, 17 different multi-element Certified Standard Reference pulp samples (CRM) with known values for Au Ag Fe S Pb Cu and Zn. 7 different CRM's with known values for Au only have been submitted with samples of drill core, RC chips and channel samples to test the precision and accuracy of the analytic procedures of the MSA,ALS and SGS laboratories used. In the results received to date there has been no systematic bias is observed. The standards demonstrate suitable precision and accuracy of the analytic process. A summary of the standard deviations from the expected values for CRM's used is summarised below. Generally, an average of standard deviations close to zero indicates a high degree of precision.



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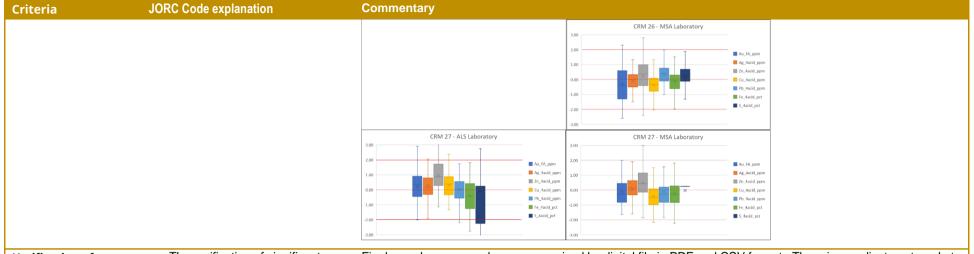
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Issued Capital 1,532m shares 126.7m options 52.1m perf rights Australian Registered Office Level 1 100 Havelock Street West Perth WA 6005 Directors
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Verification of sampling and assaying

- The verification of significant intersections by either independent or alternative company personnel.
- The use of twinned holes.
- Documentation of primary data entry procedures data verification data storage (physical and electronic) protocols.
- Discuss any adjustment to assay data.

Final sample assay analyses are received by digital file in PDF and CSV format. There is no adjustment made to any of the assay values received. The original files are backed-up and the data copied into a cloud-based drill hole database, stored offsite from the project. The data is remotely accessible for geological modelling and resource estimation.

Assay results summarised in the context of this report have been rounded appropriately to 2 significant figures. No assay data have been otherwise adjusted. Replicate assay of 186 coarse reject samples from 2019 drilling has been done to verify assay precision. Original core samples were from the 2019 DD drilling which were analysed by MSA (San Juan preparation and Vancouver analysis). Coarse reject samples were analysed by ALS (Mendoza preparation and Vancouver analysis). The repeat analysis technique was identical to the original. The repeat analyses correlate very closely with the original analyses providing high confidence in precision of results between MSA and ALS. A summary of the results for the 186 sample pairs for key elements is provided below:

	Mean		Median		Std Devia	ation	
Element	MSA	ALS	MSA	ALS	MSA	ALS	Correlation coefficient
Au (FA and GFA ppm)	4.24	4.27	0.50	0.49	11.15	11.00	0.9972
Ag (ICP and ICF ppm)	30.1	31.1	5.8	6.2	72.4	73.9	0.9903
Zn ppm (ICP ppm and ICF %)	12312	12636	2574	2715	32648	33744	0.9997
Cu ppm (ICP ppm and ICF %)	464	474	74	80	1028	1050	0.9994
Pb ppm (ICP ppm and ICF %)	1944	1983	403	427	6626	6704	0.9997
S (ICP and ICF %)	2.05	1.95	0.05	0.06	5.53	5.10	0.9987
Cd (ICP ppm)	68.5	68.8	12.4	12.8	162.4	159.3	0.9988

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Criteria	JORC Code explanation	Commentary							
		As (ICP ppm))	76.0	79.5	45.8	47.6	88.1	90.6	0.9983
		Fe (ICP %)	4.96	4.91	2.12	2.19	6.87	6.72	0.9994
		REE (ICP ppm)	55.1	56.2	28.7	31.6	98.2	97.6	0.9954
		Cd values >1000 are se	et at 1000.						

REE is the sum off Ce, La, Sc, Y. CE > 500 is set at 500. Below detection is set at zero

Replicate assay of 192 coarse reject samples from 2021 drilling has been done to verify assay precision. Original core samples were from the 2021 DD drilling which were analysed by SGS Laboratories (San Juan preparation and Lima analysis). Coarse reject samples were prepared and analysed by ALS (Mendoza preparation and Lima analysis). The repeat analysis technique was identical to the original. Except for Mo (molybdenum), the repeat analyses correlate closely with the original analyses providing confidence in precision of results between SGS and ALS. A summary of the results for the 192 sample pairs for key elements is provided below:

		Mean	n Median		Std Devia			
								Correlation
Element	count	SGS	ALS	SGS	ALS	SGS	ALS	coefficient
Au (FA and GFA ppm)	192	1.754	1.680	0.432	0.441	20.8	21.5	0.9837
Ag (ICP and ICF ppm)	192	12.14	11.57	0.93	1.03	7085	5925	0.9995
Zn (ICP and ICF ppm)	192	6829	7052	709	685	4.54E+08	5.34E+08	0.9942
Cu (ICP and ICF ppm)	192	203.4	202.9	25.7	24.5	3.30E+05	3.35E+05	0.9967
Pb (ICP and ICF ppm)	192	1768	1719	94.7	91.6	5.04E+07	4.39E+07	0.9959
S (ICP and ICF %)	192	2.23	2.10	0.94	0.87	16.51	15.56	0.9953
Cd (ICP ppm)	192	43.9	42.4	4.1	4.0	19594	18511	0.9956
As (ICP ppm))	192	45.4	45.2	16.0	16.9	10823	9893	0.9947
Fe (ICP %)	189	3.07	3.30	2.38	2.31	4.80	9.28	0.9781
REE (ICP ppm)	192	63.5	72.8	39.4	44.3	3414	4647	0.9096
Mo (ICP and ICF ppm)	192	7.69	1.68	6.74	0.97	85.83	10.33	0.3026

Values below detection were set to half the detection limit

Limit of detection for Fe was exceeded for 3 samples submitted to SGS with no overlimit analysis REE is the sum off Ce, La, Sc, Y. Vaues below detection were set at zero.

Replicate assay of 140 pulp reject samples from the 2022 drill (parts of drill holes GNDD654 and GNDD666) was done to check assay precision. The original pulps were analysed by MSA laboratories (San Juan preparation and Vancouver, Canada analysis). Replicate pulps were analysed by ALS (Lima, Peru). The analytic techniques were identical at both laboratories.

Mean

Median

Std Deviation

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Criteria	JORC Code explanation	Commentary								
										Correlation
		Element	count	SGS	ALS	SGS	ALS	SGS	ALS	coefficient
		Au (FA ppm)	140	0.27	0.30	0.01	0.02	0.98	1.05	0.9829
		Ag (ICP ppm)	140	1.16	1.14	0.16	0.16	6.15	6.31	0.9965
		Zn (ICP ppm)	140	555	565	50	56	2471	2469	0.9996
		Pb (ICP ppm)	140	92.3	95.4	13.6	13.5	338	351	0.9977
		S (ICP %)	140	0.64	0.61	0.17	0.17	1.22	1.12	0.9982
		Fe (ICP %)	140	1.62	1.59	0.64	0.66	1.91	1.88	0.9991
		exploration. A prelin assayed. The twin he GNDD003 – DDH34 GNRC110 – DDH53 GNDD144 – GNDD00 GNRC107 – GNDD00 GNDD206 – DDH54 GNDD421 – GNDD4	oles are: and 04HD08 21 – 05HD3 08/008A	3	win hole	s indicate	es similar	widths and gr	ades for l	key elements
Location of data points	 Accuracy and quality of surveys used to locate drill holes (collar and down-hole surveys) trenches 	Following completion of drilling, collars are marked and surveyed using a differential GPS (DGPS) relative to a nearby Argentinian SGM survey point. The collars have been surveyed in POSGAR 2007 zone 2 and converted to WGS84 UTM zone 19s.								
	mine workings and other locations used in Mineral Resource estimation.	Following completion of the channel sampling, the location of the channel samples is surveyed from a survey mark at the entrance to the underground workings, located using differential GPS. The locations have been surveyed in POSGAR 2007 zone 2 and converted to WGS84 UTM zone 19s.								
	Specification of the grid system used.Quality and adequacy of	The drill machine is set-up on the drill pad using hand-held survey equipment according to the proposed hole design.								
	topographic control.	Diamond core drill holes up to GNDD390 are surveyed down-hole at 30-40m intervals down hole using a down-hole compass and inclinometer tool. RC drill holes and diamond core holes from GNDD391 were continuously surveyed down hole using a gyroscope to avoid magnetic influence from the drill string and rocks. The gyroscope down-hole survey data is recorded in the drill hole database at 10m intervals.								
		Ten diamond drill ho loss of drilling equipr holes, a survey of the	nent. These	are GND	D036, 1	97, 212, 2	283, 376,	423, 425, 439), 445 and	d 465. For these
		All current and previousing DGPS to provimetre precision has	de topograpl	nic contro	I for the	Project. I	n additio	n, AWD3D DT	M model	with a nominal 2.5

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Criteria	JORC Code explanation	Commentary
		survey data with 0.1 meter precision is being acquired over the project to provide more detail where required.
Data spacing and distribution	 Data spacing for reporting of Exploration Results. Whether the data spacing and distribution is sufficient to establish the degree of geological and grade continuity appropriate for the Mineral Resource and Ore Reserve estimation procedure(s) and classifications applied. Whether sample compositing has been applied. 	Nominal 80m x 80m, 40m x 80m and 40m x 40m drill spacing is being applied to the drilling to define mineralised areas to Indicated Resource level of confidence, where appropriate. Drilling has been completed to check previous exploration, extend mineralisation along strike, and provide some information to establish controls on mineralization and exploration potential. Samples have not been composited.
Orientation of data in relation to geological structure	 Whether the orientation of sampling achieves unbiased sampling of possible structures and the extent to which this is known considering the deposit type. If the relationship between the drilling orientation and the orientation of key mineralised structures is considered to have introduced a sampling bias this should be assessed and reported if material. 	As far as is currently understood and where practicable, the orientation of sampling achieves unbiased sampling of structures and geology controlling the mineralisation. Some exploration holes have drilled at a low angle to mineralisation and have been followed up with drill holes in the opposite direction to define mineralised domains. For underground channel sampling, the orientation of the sample is determined by the orientation of the workings. Where the sampling is parallel with the strike of the mineralisation, plans showing the location of the sampling relative to the orientation of the mineralisation, weighted average grades and estimates of true thickness are provided to provide a balanced report of the mineralisation that has been sampled. Drilling has been designed to provide an unbiased sample of the geology and mineralisation targeted. In exceptional circumstances, where drill access is restricted, drilling may be non-optimally angled across the mineralised zone.
Sample security	 The measures taken to ensure sample security. 	Samples were under constant supervision by site security, senior technical personnel and courier contractors prior to delivery to the preparation laboratories in San Juan and Mendoza.
Audits or reviews	 The results of any audits or reviews of sampling techniques and data. 	There has not yet been any independent reviews of the sampling techniques and data.

Section 2 Reporting of Exploration Results

(Criteria listed in the preceding section also apply to this section.)

Criteria

JORC Code explanation

Mineral tenement and land tenure status

- Type reference name/number location and ownership including agreements or material issues with third parties such as joint ventures partnerships overriding royalties native title interests historical sites wilderness or national park and environmental settings.
- The security of the tenure held at the time of reporting along with any known impediments to obtaining a licence to operate in the area.

Commentary

The Hualilan Project comprises fifteen Minas (equivalent of mining leases) and five Demasias (mining lease extensions) held under an farmin agreement with Golden Mining SRL (Cerro Sur) and CIA GPL SRL (Cerro Norte).

Fourteen additional Minas and eight exploration licences (Cateos) have been transferred to CEL under a separate farmin agreement. Six Cateos and eight requested mining leases are directly held. This covers all of the currently defined mineralization and surrounding prospective ground. There are no royalties held over the tenements.

Granted mining leases (Minas Otorgadas) at the Hualilan Project

Name	Number	Current Owner	Status	Grant Date	Area (ha)
Cerro Sur					
Divisadero	5448-M-1960	Golden Mining S.R.L.	Granted	30/04/2015	6
Flor de Hualilan	5448-M-1960	Golden Mining S.R.L.	Granted	30/04/2015	6
Pereyra y Aciar	5448-M-1960	Golden Mining S.R.L.	Granted	30/04/2015	6
Bicolor	5448-M-1960	Golden Mining S.R.L.	Granted	30/04/2015	6
Sentazon	5448-M-1960	Golden Mining S.R.L.	Granted	30/04/2015	6
Muchilera	5448-M-1960	Golden Mining S.R.L.	Granted	30/04/2015	6
Magnata	5448-M-1960	Golden Mining S.R.L.	Granted	30/04/2015	6
Pizarro	5448-M-1960	Golden Mining S.R.L.	Granted	30/04/2015	6
Cerro Norte					
La Toro	5448-M-1960	CIA GPL S.R.L.	Granted	30/04/2015	6
La Puntilla	5448-M-1960	CIA GPL S.R.L.	Granted	30/04/2015	6
Pique de Ortega	5448-M-1960	CIA GPL S.R.L.	Granted	30/04/2015	6
Descrubidora	5448-M-1960	CIA GPL S.R.L.	Granted	30/04/2015	6
Pardo	5448-M-1960	CIA GPL S.R.L.	Granted	30/04/2015	6
Sanchez	5448-M-1960	CIA GPL S.R.L.	Granted	30/04/2015	6
Andacollo	5448-M-1960	CIA GPL S.R.L.	Granted	30/04/2015	6

Mining Lease extensions (Demasias) at the Hualilan Project

Name	Number	Current Owner	Status	Grant date	Area (ha)
Cerro Sur					
North of "Pizarro" Mine	195-152-C-1981	Golden Mining S.R.L.	Granted	29/12/1981	2.42
Cerro Norte					

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JORC Code explanation	Commentary					
	South of "Andacollo" Mine	545.208-B-94	CIA GPL S.R.L.	Pending Reconsideration	14/02/1994	1.83
	South of "Sanchez" Mine	545.209-B-94	CIA GPL S.R.L.	Registered	14/02/1994	3.50
	South of "La Toro" Mine	195-152-C-1981	CIA GPL S.R.L.	Granted	29/12/1981	2.42
	South of "Pizarro" Mine	545.207-B-94	Golden Mining S.R.L.	Registered	14/02/1994	2.09

Requested Mining Leases (Minas Solicitados)

Name	Number	Status	Area (ha)
Elena	1124.328-G-2021	Registered	2,799.24
Juan Cruz	1124.329-G-2021	Granted	933.69
Paula (over "Lo Que Vendra")	1124.454-G-2021	Application	1,460.06
Argelia	1124.486-G-2021	Registered	3,660.50
Ana Maria (over Ak2)	1124.287-G-2021	Registered	5,572.80
Erica (Over "El Peñón")	1124.541-G-2021	Application	6.00
Silvia Beatriz (over "AK3")	1124.572-G-2021	Application	2,290.75
Soldado Poltronieri (over 1124188-20,	1124.108-2022	Application	777.56
545867-R-94 and 545880-O-94)			

Mining Lease Farmin Agreements

Name	Number	Transfrred to CEL	Status	Area (ha)
Marta Alicia	2260-S-58	In Process	Granted	23.54
Marta	339.154-R-92	In Process	Granted	478.50
Solitario 1-5	545.604-C-94	In Process	Application	685.00
Solitario 1-4	545.605-C-94	In Process	Registered	310.83
Solitario 1-1	545.608-C-94	In Process	Application	TBA
Solitario 6-1	545.788-C-94	In Process	Application	TBA
AGU 3	11240114-2014	No	Granted	1,500.00
AGU 5	1124.0343-2014	No	Granted	1,443.58
AGU 6	1124.0623-2017	No	Granted	1,500.00
AGU 7	1124.0622-S-17	No	Granted	1,500.00
Guillermina	1124.045-S-2019	No	Granted	2,921.05
El Petiso	1124.2478-71	No	Granted	18.00
Ayen/Josefina	1124.495-I-20	No	Granted	2059.6

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Criteria	JORC Code explanation	Commentary							
		Exploration Lice	ence (Cateo) Farmin	Agreeme	ents				
		Name	Number	Transfrr	ed to CEL		Status	Area (ł	na)
		-	295.122-R-1989	In proces	SS	Re	gistered	1,882.	
		-	338.441-R-1993	In proces	SS	G	Granted	2,800.0	00
		-	545.880-0-1994	In proces	SS	Re	gistered	149.9	9
		-	414.998-2005	Yes		G	Granted	721.9	0
		-	1124.011-I-07	No		G	Granted	2552	-
		-	1124.012-I-07	No		Re	gistered	6677	
		-	1124.013-I-07	No			Granted	5818	
		-	1124.074-I-07	No		G	Granted	4484.	5
		Exploration Lice	ence (Cateo) Held (D	Direct Awa	ard)				
		Name	Number		Transfrred	to CEL	Status	Area (ha)	1
		-	1124-248G-20		Yes		Current	933.20	I
		-	1124-188-G-20 (2 zo	ones)	Yes		Current	327.16	I
		-	1124.313-2021		Yes		Current	986.41	I
		-	1124.564-G-2021		Yes		Current	1,521.12	I
		-	1124.632-G-2022		Yes		Current	4,287.38	1
			own impediments to						•
Exploration done by other parties	 Acknowledgment and appraisal of exploration by other parties. 	geological maps non-JORC reso Prior to explorat	oric sampling has proson, reports, trenching urce estimates plus ion by CEL, no work	data, und property of thas been	lerground s examination n complete	urveys, ons and d	drill hole re etailed stud Project sin	sults, geophy dies by multip ace 2006.	ysical surveys, ple geologists.
		Surveys of the v geology and sar	t 6 km of undergrour vorkings are likely to mpling have been co drill hole results. Hist ted by CEL.	be incon impiled a	nplete. Con nd digitised	nmonly ii I as has :	ncomplete sample dat	records of thata geological	e underground mapping adit
			on or near the Hualil The key historical ex						tends to over
		- 1995 - F - 1998 —	Lixivia SA channel s Plata Mining Limited Chilean consulting fi g and channel samp	(TSE: PN rm EPRC	/IT) 33 RC I	holes (H	ua- 1 to 33) + 1,500 RC	chip samples

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Criteria	JORC Code explanation	Commentary
	·	- 1999 – Compania Mineral El Colorado SA ("CMEC") 59 diamond core holes (DDH-20 to 79)
		plus 1,700m RC program
		 2003 – 2005 – La Mancha (TSE Listed) undertook 7,447m of DDH core drilling (HD-01 to HD-48)
		 Detailed resource estimation studies were undertaken by EPROM Ltd. (EPROM) in 1996 and CMEC (1999 revised 2000) both of which are well documented and La Mancha 2003 and 2006. The collection of all exploration data by the various operators was of a high standard and appropriate sampling techniques intervals and custody procedures were used. Not all the historic data has been archived and so there are gaps in the availability of the historic data.
Geology	 Deposit type geological setting and style of mineralisation. 	Mineralisation occurs in all rock types where it preferentially replaces limestone, shale and sandstone and occurs in fault zones and in fracture networks within dacitic intrusions.
		The mineralisation is Zn-(Pb-Cu-Ag) distal skarn (or manto-style skarn) overprinted with vein-hosted mesothermal to epithermal Au-Ag mineralisation. It has been divided into three phases – prograde skarn, retrograde skarn and a later quartz-rich mineralisation consistent with the evolution of a large hydrothermal system. Precise mineral paragenesis and hydrothermal evolution is the subject of ongoing work which is being used for exploration and detailed geometallurgical test work.
		Gold occurs in native form as inclusions with sulphide (predominantly pyrite) and in pyroxene. The mineralisation commonly contains pyrite, chalcopyrite sphalerite and galena with rare arsenopyrite, pyrrhotite and magnetite.
		Mineralisation is either parallel to bedding in bedding-parallel faults, in veins or breccia matrix within fractured dacitic intrusions, at lithology contacts or in east-west striking steeply dipping siliceous faults that cross the bedding at a high angle. The faults have thicknesses of 1–4 metres and contain abundant sulphides. The intersection between the bedding-parallel mineralisation and east-striking cross veins seems to be important in localising the mineralisation.
		Complete oxidation of the surface rock due to weathering is thin. A partial oxidation / fracture oxidation layer near surface is 1 to 40m thick and has been modelled from drill hole intersections.
Drill hole Information	 A summary of all information material to the understanding of the exploration results including a tabulation of the following information for all Material drill holes: easting and northing of the drill hole 	Significant intersections previous reported for historic drill holes, DD drill holes, RC drill holes completed by CEL are detailed in CEL ASX releases: 1 June 2022 (Maiden MRE): https://announcements.asx.com.au/asxpdf/20220601/pdf/459jfk8g7x2mty.pdf and 29 March 2023 (MRE update): https://announcements.asx.com.au/asxpdf/20230329/pdf/45n49jlm02grm1.pdf
	collar - elevation or RL (Reduced Level – elevation above sea level in metres) of the drill hole collar	A cut-off grade of 1 g/t Au equivalent has been used with up to 2m of internal diltion or a cut-off grade of 0.2 g/t Au equivalent and up to 4m of internal diltion has been allowed. No metallurcial or recovery factors have been used in the intersections reported.

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	 dip and azimuth of the hole down hole length and interception depth hole length. If the exclusion of this information is justified on the basis that the information is not Material and this exclusion does not detract from the understanding of the report the Competent Person should clearly explain why this is the case. 	
Data aggregation methods	 In reporting Exploration Results weig averaging techniques maximum and minimum grade truncations (eg cuttin high grades) and cut-off grades are used to the Material and should be stated. Where aggregate intercepts incorport short lengths of high-grade results at longer lengths of low-grade results the procedure used for such aggregations should be stated and some typical examples of such aggregations shown in detail. The assumptions used for any report metal equivalent values should be classified. 	are reported to cut-off grade of a 1.0 g/t Au equivalent and 10 g/t Au equivalent allowing for up to 2m of internal dilution between samples above the cut-off grade and 0.2 g/t Au equivalent allowing up to 10m of internal dilution between samples above the cut-off grade. The following metals and metal prices have been used to report gold grade equivalent (AuEq): Au US\$ 1780 / oz Ag US\$24 /oz and Zn US\$ 2800 /t. Metallurgical recoveries for Au, Ag and Zn have been estimated from the results of interim metallurgical test work completed by SGS Metallurgical Operations in Lakefield, Ontario using a combination of gravity and flotation of a combined metallurgical sample from 5 drill holes. Using data from the interim test results, and for the purposes of the AuEq calculation for drill hole significant intercepts, gold recovery is estimated For the AuEq calculation average metallurgical recovery is estimated to be 94.9% for gold, 90.9% for silver, 67.0% for Zn and 57.8% for Pb. Metal prices used to report AuEq are Au US\$ 1900 / oz, Ag US\$24 /oz, Zn US\$ 4,000 /t and Pb US 2,000/t Accordingly, the formula used for Au Equivalent is: AuEq (g/t) = Au (g/t) + [Ag (g/t) x (24/1900) x (0.578/.9490). Metallurgical test work and geological and petrographic descriptions suggest all the elements included in the metal equivalents calculation have reasonable potential of eventual economic recovery. While Cu and Pb are reported in the table above as they were not yet considered economically significant at the time of the interim metallurgical test results, these metals were not used in the Au equivalent calculation at this early stage of the Project.
		No top cuts have been applied to the reported grades.
Relationship between mineralisation	 These relationships are particularly important in the reporting of Exploration Results. 	The mineralisation is moderately or steeply dipping and strikes NNE and ENE. For some drill holes, there is insufficient information to confidently establish the true width of the mineralized intersections at this stage of the exploration program.

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Issued Capital

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52.1m perf rights

Criteria	JORC Code explanation	Commentary
widths and intercept lengths	 If the geometry of the mineralisation with respect to the drill hole angle is known its nature should be reported. If it is not known and only the down hole lengths are reported there should be a clear statement to this effect (eg 'down hole length true width not known'). 	Apparent widths may be thicker in the case where the dip of the mineralisation changes and/or bedding-parallel mineralisation intersects NW or ENE-striking cross faults and veins. Representative cross section interpretations have been provided periodically with releases of significant intersections to allow estimation of true widths from individual drill intercepts.
Diagrams	 Appropriate maps and sections (with scales) and tabulations of intercepts should be included for any significant discovery being reported These should include but not be limited to a plan view of drill hole collar locations and appropriate sectional views. 	Representative maps and sections are provided in the body of reports released to the ASX.
Balanced reporting	 Where comprehensive reporting of all Exploration Results is not practicable representative reporting of both low and high grades and/or widths should be practiced to avoid misleading reporting of Exploration Results. 	All available final data have been reported where possible.
Other substantive exploration data	- Other exploration data if meaningful and material should be reported including (but not limited to): geological observations; geophysical survey results; geochemical survey results; bulk samples – size and method of treatment; metallurgical test results; bulk density groundwater geotechnical and rock characteristics; potential deleterious or contaminating substances.	Specific gravity measurements have been taken from the drill core recovered during the drilling program. These data are used to estimate densities in Resource Estimates. Eight Induced Polarisation (IP) lines have been completed in the northern areas of the Project. Stage 1 surveying was done on 1 kilometre length lines oriented 115° azimuth, spaced 100m apart with a 50m dipole. The initial results indicate possible extension of the mineralisation with depth. Stage 2 surveying was done across the entire field on 1 – 3 kilometre length lines oriented 090°, spaced 400m apart with a 50m dipole. On-going data interpretation is being done as drilling proceeds. Three ground magnetic surveys and a drone magnetic survey have been completed. The results of these data and subsequent geological interpretations are being used to guide future exploration. Metallurgical test results are used to estimate the AuEq (gold equivalent) as detailed above in <i>Data Aggregation</i> and below in <i>Section 3: Metallurgical Factors or Assumptions</i> . The formula used for AuEq is: AuEq (g/t) = Au (g/t) + [Ag (g/t) x (24/1900) x (0.909/0.949)] + [Zn (%) x (40.00*31.1/1900) x (0.670/0.949)] + (Pb (%) x 20.00*31.1/1900) x (0.578/.9490). Point resistivity surveys have been completed east of the Project for the purposes of detecting the presence of groundwater. Three surveys (total of 22 points) have been completed. A water bore has been drilled approximately 4 kilometres to the east of the Project which found water in permeable Quaternary sedimentary deposits above hard-rock basement at 128 metres vertical depth. Testing and

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Criteria	JORC Code explanation	Commentary
Further work	 The nature and scale of planned further work (eg tests for lateral extensions or depth extensions or large-scale step-out drilling). Diagrams clearly highlighting the areas of possible extensions including the main geological interpretations and future drilling areas provided this information is not commercially 	commissioning of the bore has yet to be completed. Further geophysical test work is planned to determine the extent of the aquifer. • CEL Plans to undertake the following over the next 12 months • Additional resource extension, infill and exploration drilling; • Geophysical tests for undercover areas. • Structural interpretation and alteration mapping using high resolution satellite data and geophysics to better target extensions of known mineralisation. • Field mapping program targeting extensions of known mineralisation. • Further metallurgical test work.
	sensitive.	

Section 3 Estimation and Reporting of Mineral Resources

(Criteria listed in the preceding section also apply to this section.)

Criteria	JORC Code explanation	Commentary
Database integrity	 Measures taken to ensure that data has not been corrupted by for example transcription or keying errors between its initial collection and its use for Mineral Resource estimation purposes. Data validation procedures used. 	Geological logging completed by previous explorers was done on paper copies and transcribed into a series of excel spreadsheets. These data have been checked for errors. Checks have been made against the original logs and with follow-up twin and close spaced drilling. Only some of the historic drill holes have been used in the Resource Estimate, including the results presented in Section2. Some drill holes have been excluded where the geology indicates that the drill hole is likely mis-located or where the drill hole has been superseded by CEL drilling. For CEL drilled holes, assay data is received in digital format. Backup copies are backed up into a cloud-based file storage system and the data is entered into a drill hole database which is also securely backed up off site.
		The drill hole data is backed up and is updated periodically by the CEL GIS and data management team.
Site visits	 Comment on any site visits undertaken by the Competent Person and the outcome of those visits. If no site visits have been undertaken indicate why this is the case. 	The Competent Person has undertaken site visits during exploration. Site visits were undertaken in 2019 and 2020 before COVID-19 closed international travel. Post COVID numerous site visits have undertaken since November 2021. The performance of the drilling program, collection of data, sampling procedures, sample submission and exploration program were initiated and reviewed during these visits.
Geological interpretation	 Confidence in (or conversely the uncertainty of) the geological interpretation of the mineral deposit. Nature of the data used and of any assumptions made. The effect if any of alternative interpretations on Mineral Resource estimation. The use of geology in guiding and controlling Mineral Resource estimation. The factors affecting continuity both of grade and geology. 	The geological interpretation is considered appropriate given the drill core density of data that has been collected, access to mineralisation at surface and underground exposures. Given the data, geological studies past and completed by CEL, the Competent Person has a high level of confidence in the geological model that has been used to constrain the mineralised domains. It is assumed that networks of fractures controlled by local geological factors have focussed hydrothermal fluids and been the site of mineralisation in both the prograde zinc skarn and retrograde mesothermal – epithermal stages of hydrothermal evolution. The interpretation captures the essential geometry of the mineralised structure and lithologies with drill data supporting the findings from the initial underground sampling activities. Mineralised domains have been built using explicit wireframe techniques from 0.2 – 0.5 g/t AuEq mineralised intersections, joined between holes by the instruction from the geology and structure. Continuity of grade between drill holes is determined by the intensity of fracturing, the host rock contacts (particularly dacite – limestone contacts) and by bedding parallel faults, particularly within limestone, at the limestone and overlying sedimentary rock contact and within the lower sequences of the sedimentary rocks within 40m of the contact. No alternative interpretations have been made form which a Mineral Resource Estimate has been made.
Dimensions	 The extent and variability of the Mineral Resource expressed as length (along strike or otherwise) plan width and depth below surface to the upper and lower limits of the Mineral 	31 separate domains were interpreted over a strike length of 2.3kms. The domains vary in width and orientation from 2m up to 100m in width. The deepest interpreted domain extends from the surface down approximately 600m below surface.

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Criteria	JORC Code explanation	Commentary					
	Resource.						
Estimation and modelling techniques	 The nature and appropriateness of the estimation technique(s) applied and key assumptions including treatment of extreme grade values domaining interpolation 	Estimation was made for Au Ag, Zn and Pb being the elements that for pyrite which is of estimate the density for bocks in the Mineral Resource	economic and				
	parameters and maximum distance of extrapolation from data points. If a computer assisted estimation method was chosen include	No previous JORC Resource estimates or non-JORC For compare to the current Resource estimate. No produc	-				hods to
	a description of computer software and parameters used. - The availability of check estimates previous	A 2m composite length was selected after reviewing the average length of 1.54m for samples taken within the	•		rom the drillii	ng which shov	wed an
	estimates and/or mine production records and whether the Mineral Resource estimate takes appropriate account of such data. The assumptions made regarding recovery of	A statistical analysis was undertaken on the sample co domain-by-domain basis. The domains were then grout top cuts were applied in order to reduce the influence downgrading the high-grade composites too severely.	uped by host of extreme va	rock and mine alues on the r	eralisation sty esource estim	le and group on ates without	domain
	 by-products. Estimation of deleterious elements or other non-grade variables of economic significance (eg sulphur for acid mine drainage 	distribution of the grade population within each group became erratic. The following table shows the top cut No top cut was applied to estimation of Fe and S.	-				
	characterisation).	Group	Au (ppm)	Ag (ppm)	Zn (%)	Pb (%)	
	 In the case of block model interpolation the block size in relation to the average sample 	Fault Zone hosted (Magnata and Sanchez) and CAL (limestone) hosted	80	300	20	5	
	spacing and the search employed.	LUT (siltstone) hosted	20	100	5	1	
	 Any assumptions behind modelling of selective mining units. 	DAC (intrusive) hosted	15	70	5	1.8	
	 Any assumptions about correlation between variables. 	Block modelling was undertaken in Surpac™ V6.6 softv	vare.				
	 Description of how the geological interpretation was used to control the resource estimates. 	A block model was set up with a parent cell size of 10m (E) x 20m (N) x 10m (RL) with standard sub-celling to 2.5m (E) x 5.0m (N) x 2.5m (RL) to maintain the resolution of the mineralised domains. The 20m Y and vertical block dimensions were chosen to reflect drill hole spacing and to provide definition for potential mine planning. The shorter 10m X dimension was used to reflect the geometry and orientation of the majority of the domain					ck
	 Discussion of basis for using or not using grade cutting or capping. 	wireframes.	ieu y and orie	ntation of the	e majority of t	ne domain	
	 The process of validation the checking process used the comparison of model data to drill hole data and use of reconciliation data if available 	Group Variography was carried out using Leapfrog Edg the 31 domains for each variable.	e software or	n the two met	re composite	d data from e	ach of

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Criteria	J(ORC Code explanation	Commentary
			All relevant variables; Au, Ag, Pb, Zn, Fe and S in each domain were estimated using Ordinary Kriging using only data from within that domain. The orientation of the search ellipse and variogram model was controlled using surfaces designed to reflect the local orientation of the mineralized structures.
			An oriented "ellipsoid" search for each domain was used to select data for interpolation. A 3 pass estimation search was conducted, with expanding search ellipsoid dimensions and decreasing minimum number of samples with each successive pass. First passes were conducted with ellipsoid radii corresponding to 40% of the complete range of variogram structures for the variable being estimated. Pass 2 was conducted with 60% of the complete range of variogram structures for the variable being estimated. Pass 3 was conducted with dimensions corresponding to 200% of the semi-variogram model ranges. Blocks within the model where Au was not estimated during the first 3 passes were assigned as unclassified. Blocks for Ag, Pb, Zn, Fe and S that were not estimated were assigned the average values on a per-domain basis.
			Validation checks included statistical comparison between drill sample grades and Ordinary Kriging block estimate results for each domain. Visual validation of grade trends for each element along the drill sections was also completed in addition to swath plots comparing drill sample grades and model grades for northings, eastings and elevation. These checks show good correlation between estimated block grades and drill sample grades.
Moisture	-	Whether the tonnages are estimated on a dry basis or with natural moisture and the method of determination of the moisture content.	Tonnage is estimated on a dry basis.
Cut-off parameters	-	The basis of the adopted cut-off grade(s) or quality parameters applied.	The following metals and metal prices have been used to report gold grade equivalent (AuEq): Au US\$ 1900 / oz, Ag US\$24 /oz, Zn US\$ 4,000 /t and Pb US 2,000/t. Average metallurgical recoveries for Au, Ag, Zn and Pb have been estimated from the results of Stage 1 metallurgical test work completed by SGS Metallurgical Operations in Lakefield, Ontario using a combination of gravity and flotation combined metallurgical samples as detailed in the Criteria below. For the AuEq calculation average metallurgical recovery is estimated as 94.9% for gold, 90.9% for silver, 67.0% for Zn and 57.8% for Pb. Accordingly, the formula used for Au Equivalent is: AuEq (g/t) = Au (g/t) + [Ag (g/t) x (24/1900) x (0.909/0.949)] + [Zn (%) x (40.00*31.1/1900) x (0.670/0.949)] + (Pb (%) x 20.00*31.1/1900) x (0.578/.9490).
			Based on the break-even grade for an optimised pit shell for gold equivalent, a AuEq cut-off grade of 0.30 ppm is used to report the resource within an optimised pit shell run at a gold price of US\$1,800 per ounce and allowing for Ag, Zn and Pb credits. Under this scenario, blocks with a grade above the 0.30 g/t Au Eq cut off are considered to have reasonable prospects of mining by open pit methods. A AuEq cut-off grade of 1.0 ppm was used to report the resource beneath the optimised pit shell run as these blocks are considered to have reasonable prospects of future mining by underground methods.

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JORC Code explanation Commentary Criteria Assumptions made regarding possible mining The Resource estimate has assumed that near surface mineralisation would be amenable to open pit mining given Mining factors or methods minimum mining dimensions and that the mineralisation is exposed at surface and under relatively thin unconsolidated cover. A surface mine assumptions internal (or if applicable external) mining optimiser has been used to determine the proportion of the Resource estimate model that would be amenable to eventual economic extraction by open pit mining methods. The surface mine optimiser was bult using the dilution. It is always necessary as part of the process of determining reasonable prospects following parameters with prices in USD: for eventual economic extraction to consider Au price of \$1.800 per oz. Ag price of \$23.4 per oz. Zn price of \$3.825 per tonne and Pb price of \$1.980 potential mining methods but the assumptions per tonne made regarding mining methods and Average metallurgical recoveries of 94.9% for Au, 90.9 % for Ag and 67 % for Zn and 57.8 % for Pb. parameters when estimating Mineral Ore and waste mining cost of \$2.00 per tonne Resources may not always be rigorous. Where Unconsolidated cover removal cost of \$0.10 per tonne this is the case this should be reported with an Processing cost of \$10.00 per tonne explanation of the basis of the mining Transport and marketing of \$50 / oz of AuEq (road to Jan Juan then rail to Rosario Port) assumptions made. Royalty of \$60 per oz Au, 3% for Ag, Zn and Pb. Assumed concentrate payability of 94.1% for Au, 82.9% for Ag, 90 % for Zn and 95 % for Pb. 45° pit slopes on the western side of the pit and 55° on the eastern side of the pit Blocks above a 0.30 g/t AuEg within the optimised open pit shell are determined to have reasonable prospects of future economic extraction by open pit mining and are included in the Resource estimate on that basis. Blocks below the open pit shell that are above 1.0 g/t AuEq are determined to have reasonable prospects of future economic extraction by underground mining methods and are included in the Resource estimate on that basis. The basis for assumptions or predictions CEL has completed Stage 1 metallurgical test work on representative composite sample of mineralisation Metallurgical regarding metallurgical amenability. It is from: factors or always necessary as part of the process of 1. Two separate composite samples of limestone-hosted massive sulphide (manto) Sample A has a weighted assumptions determining reasonable prospects for eventual average grade of 10.4 g/t Au, 31.7 g/t Ag, 3.2 % Zn and 0.46 % Pb. Sample B has a weighted average grade of economic extraction to consider potential 9.7 g/t Au, 41.6 g/t Ag, 4.0% Zn and 0.48% Pb. 2. One dacite (intrusive) composite sample with a weighted average grade of 1.1 g/t Au, 8.1 g/t Ag and 0.10 % metallurgical methods but the assumptions regarding metallurgical treatment processes Zn and 0.04% Pb. 3. One sediment hosted (fine grained sandstone and siltstone) composite sample with a weighted average and parameters made when reporting Mineral Resources may not always be rigorous. Where grade of 0.68 g/t Au, 7.5 g/t Ag, 0.34 % Zn and 0.06 % Pb. this is the case this should be reported with an 4. One oxidised limestone (manto oxide) composite sample with a weighted average grade of 7.0 g/t Au, 45 explanation of the basis of the metallurgical g/t Ag, 3.7% Zn and 0.77% Pb. assumptions made. Gravity recovery and sequential flotation tests of the higher-grade limestone hosted mineralisation involved; 1. primary P80 = 51 micron primary grind, 2. gravity recovery, 3. Pb-Cu followed by Zn rougher flotation,

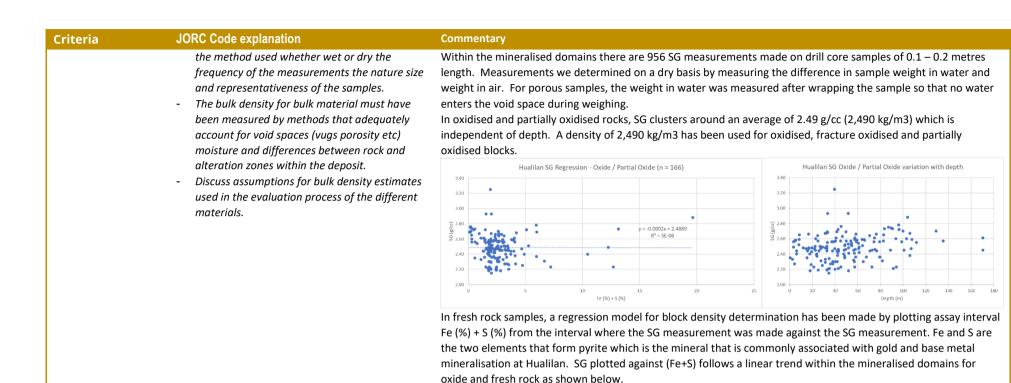
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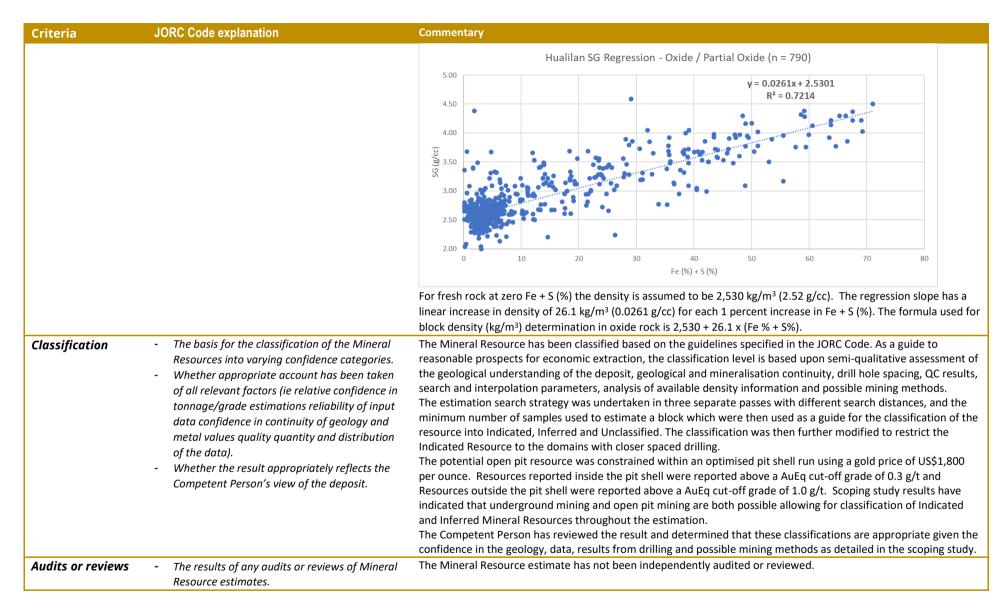
Criteria	JORC Code explanation	Commentary
		4. p80 = 29 micron regrind of the Zn rougher concentrate,
		5. two re-cleaning stages of the Pb/Cu rougher concentrate,
		6. four re-cleaning Sages on the Zn rougher concentrate, and
		7. additional gravity recovery stages added to the Zn Rougher concentrate
		This results in the following products that are likely to be saleable
		- Au-Ag concentrate (118 g/t Au, 286 g/t Ag) with low deleterious elements,
		- Pb concentrate (65% Pb, 178 g/t Au, 765 g/t Ag) with low deleterious elements, and
		- Zn concentrate (51% Zn, 10 g/t Au, 178 g/t Ag) with low deleterious elements, relatively high Cd, but at a
		level that is unlikely to attract penalties.
		- tailing grades of 2 to 3 g/t Au which respond to intensive cyanide leach with recoveries of 70-80% of any
		residual gold and silver to a gold doré bar.
		Two intensive leach tests of Au-Ag concentrate to doré have been completed using a representative sample of
		the Au-Ag concentrate. One split of the sample was finely ground to p80 of 16.7 μm and the second split
		finely ground to p80 of 40 μm. The 16.7 μm sample returned a recovery of 96.0% Au and the 40 μm sample
		returned a recovery of 92.8% Au. These results provide an option to eliminate concentrate transport costs and
		increase payability for the Au-Ag concentrate.
		Gravity recovery and flotation tests of the intrusive-hosted mineralisation involved;
		1. primary P80 = 120-80 micron primary grind,
		2. gravity recovery,
		3. single stage rougher sulphide flotation,
		4. P80 = 20-30 micron regrind of the rougher concentrate (5-10% mass),
		5. one or two re-cleaning stages of the Au-Ag Rougher concentrate
		At primary grind of p80 = 76 micron and regrind of p80 = 51 micron an Au-Ag concentrate can be produced
		grading 54 g/t Au and 284 g/t Ag with total recoveries of 97% (Au) and 85% (Ag).
		One test of a sediment hosted composite sample (5-10% of the mineralisation at the Project) was a repeat of
		the testing done on the intrusive-hosted mineralisation. This produced an Au-Ag concentrate grading 23.6 g/t
		Au and 234 g/t Ag at total recoveries of 85% (Au) and 87% (Ag). Further test work is likely to be done as part
		of more detailed studies. It is likely that the concentrate produced from the sediment-hosted mineralisation
		will be combined with the Au-Ag concentrate from the limestone and intrusive-hosted mineralisation.
		Applying recoveries of 70% for both gold and silver to the various concentrate tailings components
		where leaching is likely to be undertaken during production generates recoveries of:
		• 95% (Au), 93% (Ag), 89% (Zn), 70% (Pb) from the high-grade skarn (manto) component of the mineralisation;

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		 96% (Au) and 88% (Ag) from the intrusion-hosted component of the mineralisation;
		 85% (Au) and 87% (Ag) from the sediment-hosted component of the mineralisation;
		An intensive cyanide leach test of oxide (limestone and dacite hosted mineralisation has produced recoveries of 78% (Au) and 64% (Ag) which is expected to be recovered into gold doré bar. While the oxide component of the mineralisation comprises only a small percentage of the Hualilan mineralisation its lies in the top 30-40 metres and would be mined early in the case of an open pit operation.
		Based on the test work to date and the proportions of the various mineralisation types in the current geological model, it is expected that overall average recoveries for potentially saleable metals will be: - 94.9% Au, - 90.9% for Ag - 67.0% for Zn and - 57.8% for Pb As further results are obtained, these assumptions will be updated.
		Additional Stage 2 work involving column testing of low-grade material, improved recovery of Zn in lower-grade mineralisation, comminution and variability testing, blended test work, and pilot plant testing is ongoing and planned.
Environmental factors or	 Assumptions made regarding possible waste and process residue disposal options. It is 	It is considered that there are no significant environmental factors which would prevent the eventual extraction of gold from the project. Environmental surveys and assessments have been completed in the past and will form a
assumptions	always necessary as part of the process of determining reasonable prospects for eventual economic extraction to consider the potential environmental impacts of the mining and processing operation. While at this stage the determination of potential environmental impacts particularly for a greenfields project may not always be well advanced the status of early consideration of these potential	part of future pre-feasibility studies.
	environmental impacts should be reported. Where these aspects have not been considered this should be reported with an explanation of the environmental assumptions made.	
Bulk density	 Whether assumed or determined. If assumed the basis for the assumptions. If determined 	CEL has collected specific gravity (SG) measurements from drill core, which have been used to estimate block densities for the Resource estimate.

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Criteria JORC Code explanation Commentary Discussion of Where appropriate a statement of the relative There is sufficient confidence in the data quality drilling methods and analytical results that they can be relied accuracy and confidence level in the Mineral upon. The available geology and assay data correlate well. The approach and procedure is deemed appropriate relative accuracy/ Resource estimate using an approach or given the confidence limits. The main factors which could affect relative accuracy are: confidence procedure deemed appropriate by the domain boundary assumptions Competent Person. For example the application orientation of statistical or aeostatistical procedures to grade continuity quantify the relative accuracy of the resource top cut. within stated confidence limits or if such an Grade continuity is variable in nature in this style of deposit and has not been demonstrated to date and closer approach is not deemed appropriate a spaced drilling is required to improve the understanding of the grade continuity in both strike and dip directions. It qualitative discussion of the factors that could is noted that the results from the twinning of three holes by La Mancha are encouraging in terms of grade affect the relative accuracy and confidence of repeatability. the estimate. The deposit contains very high grades and there is need for the use of top cuts. The statement should specify whether it relates No production data is available for comparison. to global or local estimates and if local state the relevant tonnages which should be relevant to technical and economic evaluation. Documentation should include assumptions made and the procedures used. These statements of relative accuracy and confidence of the estimate should be compared with production data where available.