

Hualilan drilling continues to extend mineralisation well beyond the limits of the current Mineral Resource Estimate

Highlights

- Extension drilling at Hualilan continues to deliver outstanding results including:
 - 65.3m at 2.4 g/t AuEq¹
 12.2m at 11.0 g/t AuEq¹

8.3m at 16.0 g/t AuEq1

- 2.3 g/t Au, 1.7 g/t Ag, 0.2% Zn from 209.0m and
- 10.1 g/t Au, 11.7 g/t Ag, 1.5% Zn from 324.9m including
 14.8 g/t Au, 17.1 g/t Ag, 2.2% Zn from 324.9m (GNDD-684);
- 8.4m at 5.5 g/t AuEq¹
 2.6m at 17.4 g/t AuEq¹
 20.0m at 1.1 g/t AuEq¹
 5.5m at 2.6 g/t AuEq¹
 1.9m at 23.3 g/t AuEq¹
- 4.5 g/t Au, 5.3 g/t Ag, 2.0% Zn from 362.4 including.
- 14.3 g/t Au, 16.1 g/t Ag, 6.3% Zn from 362.4m and
 1.0 g/t Au, 0.9 g/t Ag, 0.2% Zn from 409.0m including
 2.4 g/t Au, 1.1 g/t Ag, 0.4% Zn from 413.0m and;
- 20.0 g/t Au, 8.3 g/t Ag, 6.8% Zn from 651.2m(GNDD-685)
- 94.0m at 0.7 g/t AuEq¹
 2.0m at 8.8 g/t AuEq¹
 45.0m at 0.7 g/t AuEq¹
- 0.6 g/t Au, 1.2 g/t Ag, 0.1% Zn from 17.0m including
 8.8 g/t Au, 0.2 g/t Ag, 0.1% Zn from 109.0m and
- 0.5 g/t Au, 6.3 g/t Ag, 0.2% Zn from 314.0m (GNDD-661);
- Drilling extends the Magnata Fault mineralisation significantly deeper and defines a new zone, with the same orientation as the Magnata Fault, 200 metres south of the Magnata Fault.
- Several holes extend the Verde Zone mineralisation up-dip to near surface
- Exploration holes up to 1 kilometre west of the resource boundary intersect mineralisation
- Drilling continues to point to a significant increase in the footprint of the mineralisation with the mineralisation at Hualilan remaining open in all directions

Commenting on the first drilling results after the resource, CEL Managing Director, Mr Kris Knauer, said

"These results continue to give us confidence that our maiden resource estimate for Hualilan will increase significantly as we continue to drill.

The high-grade intercept near the bottom of hole GNDD-685 opens an exciting drill target that covers at least 700 metres vertically. We are also continuing to find new zones of mineralisation above and below the existing mineralisation and importantly, the majority of this is within the current Mineral Resource Estimate pit shell.

Our infill drill holes are generally surprising on the upside both in terms of mineralisation width and grade which should positively impact on the resource upgrade. At this stage the size of Hualilan is only limited by the amount of drill metres"



Challenger Exploration (ASX: CEL) ("CEL" the "Company") is pleased to announce results from drilling at its flagship Hualilan Gold Project, San Juan Argentina. The results are from the Company's ongoing drill program targeting extensions to the current 2.1 million ounce AuEq¹ Mineral Resource Estimate ("MRE")². The holes reported in this release primarily focus on extensions to the Verde Zone and Magnata Fault mineralisation. The drilling continues to show the presence of coherent zones of significantly higher-grade mineralisation at depth in the Verde Zone.

All results were received after the completion of the Company's maiden Hualilan Gold Project MRE. The current MRE, which includes a high-grade core of 1.1 Moz at 5.6 g/t AuEq¹, was based on 125,700 metres drilling and will be updated once assays from the recently completed 204,000 metre drill program have been received. Following its \$24.7 million dollar financing in September CEL has committed to a further 50,000 metres which will take total CEL drill metres at Hualilan to 254,000 metres.

¹ Reported as Gold Equivalent (AuEg) values – for requirements under the JORC Code see page 14

² Refer to the about Challenger section for MRE

Drillhole	Intercept	Comment	Gram x					
Diminoic	(AuEq)	comment	Metres					
GNDD-557	61.7m at 0.3 g/t AuEq	extends the southern Verde Zone mineralisation over 200 metres	18.0					
GNDD-337	and 25.6m at 0.5 g/t AuEq	west of the current MRE boundary	12.3					
GNDD-613	31.3m at 0.6 g/t AuEq	31.3m at 0.6 g/t AuEq Verde Zone infill hole with the mineralisation intersected 50% wider						
GNDD-013	inc 9.3 m at 1.6 g/t AuEq	than anticipated from the current MRE model	15.0					
GNDD-624	64.0m at 0.5 g/t AuEq	Verde Zone infill hole intersecting a wider zone than expected based	31.0					
GNDD-024	inc 10.0m at 1.4 g/t AuEq	on the MRE block model	14.0					
GNDD-644	43.0m at 0.9 g/t AuEq	Confirms continuity of mineralisation in upper zone and extends the	38.7					
GNDD-044	and 12.5m at 1.5 g/t AuEq	deeper zone of mineralisation 50 metres up-dip	18.4					
GNDD-648	28.0m at 1.0 g/t AuEq	extends Verde Zone mineralisation 80 metres up-dip to surface with	27.4					
GNDD 040		intersection significantly higher-grade than adjacent drilling						
	9.8m at 0.9 g/t AuEq and		8.7					
GNDD-655	7.8m at 1.2 g/t AuEq and	Extends Verde Zone mineralisation 40 metres up dip to near surface	9.2					
	4.9m at 1.7 g/t AuEq		6.2					
	94.0m at 0.7 g/t AuEq and	Verde Zone infill hole that intersected several stacked zones of	63.8					
GNDD-661	4.0m at 3.0 g/t AuEq and	mineralisation over a significantly wider interval than predicted by	12.0					
	45.0m at 0.7 g/t AuEq	the current MRE block model	29.2					
	65.3m at 2.4 g/t AuEq inc	Verde Zone intersection materially wider and higher-grade than the	156.9					
	1.9m at 74.5 g/t AuEq and	current MRE block model in the upper 65.5m intersection.	141.6					
GNDD-684	12.2m at 11.0 g/t AuEq inc	The decomplishes with a sector of high condenses at Wards 50	133.1					
	8.3m at 16.0 g/t AuEq and	The deeper intersections extend high-grade zones at Verde 50 metres up dip from the current MRE boundary.	132.4					
	7.6m at 1.2 g/t AuEq and	metres up tip from the current with boundary.	8.8 6.5					
	10.0m at 0.6 g/t AuEq							
	8.4m at 5.5 g/t AuEq inc	Extends the Magnata Fault mineralisation 125 metres below the	46.4					
GNDD-685	2.6m at 17.4 g/t AuEq and 20.0m at 1.1 g/t AuEq inc	current MRE boundary.	45.2 21.7					
כאס-טטאוט	5.5m at 2.6 g/t AuEq and	Deep intersection correlates with 2.8m at 9.8 g/t AuEq in GNDD-134. These define a second zone of mineralisation parallel to the	21.7 14.3					
	1.9m at 23.3 g/t AuEq	Magnata Fault with up to 700 metres vertical extent	43.1					
GNDD-697	63.5m at 0.7 g/t AuEq inc 4.5m at 3.3 g/t AuEq and	75 metres north of the MRE boundary. Extends the Verde Zone along strike and 100 metres up-dip and demonstrates that the	47.7 15.0					
GNDD-097	3.6m at 5.0 g/t AuEq and	mineralisation is strong and open to the north along strike	13.0 17.9					
	J.OH at J.O g/t AuEq	eransacion is strong and open to the north diong strike	17.3					

Table 1. Summary of significant Intercepts reported this ASX Release

Challenger Exploration Limited ACN 123 591 382 ASX: CEL

Issued Capital 1,044.9m shares 10m options 120m perf shares 16m perf rights

Australian Registered Office 1205 Hay Street West Perth WA 6005

Directors Mr Kris Knauer, MD and CEO Mr Scott Funston, Finance Director Mr Fletcher Quinn, Chairman Mr Sergio Rotondo, Exec. Director



HIGHLIGHTS

The results continue to exceed the Company's expectations and confirm that mineralisation remains open in all directions and there is clear potential for the MRE to grow significantly via continued extension drilling.

THE VERDE ZONE

The Verde Zone contributes almost 1 million ounces gold equivalent¹ to the current Hualilan MRE when the new high-grade zones at depth are included. The Verde Zone was a CEL discovery targeted using surface magnetics and IP (Induced Polarisation) at the Hualilan Gold Project. The discovery hole (ASX release 2/3/21) returned 125.5 metres at 1.1 g/t AuEq including 71.0 metres at 1.8 g/t AuEq (GNDD-169). The Verde Zone covers 2.0 kilometres of strike and mineralisation remains open along strike and at depth.

Mineralisation in the Verde Zone is oriented north-south, is 50 to 100 metres wide, and hosted by bedding parallel fault-fracture zones in sediments and steeply dipping fracture zones in intrusives. A lower grade halo of mineralisation extends into the overlying sedimentary rocks which have been locally brecciated by the hydrothermal fluids during mineralisation. The overlying mineralisation in the sedimentary rocks dips to the west at 30-50° and is up to 50 metres thick. This overlying halo of lower grade mineralisation is a useful exploration guide to vector to the deeper intrusion-hosted mineralisation. As drilling extends deeper, zones of high-grade skarn mineralisation are being intersected at both limestone-intrusive contacts and also within limestone which is analogous to the Main Norte and Sentazon Manto mineralisation.

The infill and extension drilling at the Verde and Gap Zones is designed as a series of fences of holes at 40 metre spacing along strike. Holes on each fence were collared to target the mineralisation 40 metres below the previous hole. The intention is to drill the entire 2.0 kilometre Verde Zone down to 400 metres vertically on 40 x 40 metre spacing. The infill portion of this program is ongoing as, mineralisation continues to be extended further north and south along strike, and at depth. Accordingly, the focus is to continue to expand the footprint of the mineralisation rather than infill drilling.

GNDD-684

GNDD-684 was a 40 metre spaced infill hole In the Verde zone between GNDD-361 (69.0 metres at 0.3 g/t AuEq) and GNDD-350 (0.6 metres at 4.9 g/t AuEq). GNDD-684 materially upgrades the existing MRE block model intersecting significantly wider and higher grade mineralisation in the main Verde Zones. Additionally, the hole intersected a new zone of mineralisation above the main Verde zones within the \$1800 MRE pit shell and two new zones of mineralisation at depth.

The intersection of 65.3m at 2.4 g/t AuEq (2.3 g/t gold, 1.7 g/t silver, 0.1% lead, 0.2% zinc) from 209.0m including 1.9m at 74.5 g/t AuEq (71.6 g/t gold, 41.1 g/t silver, 3.5% lead 3.8% zinc) is materially thicker and higher in grade than the MRE block model (Figure 1) and is another example of infill drilling significantly upgrading the MRE.



The deeper Intersection of 12.2m at 11.0 g/t AuEq (10.1 g/t gold, 11.7 g/t silver, 0.1% lead, 1.5% zinc) from 324.9m including 8.3m at 16.0 g/t AuEq (14.8 g/t gold, 17.1 g/t silver, 0.1% lead, 2.2% zinc) extends the deeper high-grade zone of Verde mineralisation 60 metres up-dip from GNDD-361 (4.8 metres at 8.8 g/t AuEq) and is again, substantially thicker and higher grade.

The deeper intersection of **7.6m** at **1.2** g/t AuEq (**1.1** g/t gold, **0.7** g/t silver, **0.1%** zinc) from 354.5m including **2.2m** at **3.2** g/t AuEq (**3.2** g/t gold, **0.9** g/t silver) correlates with the intersection of 2.0 metres at 9.4 g/t in GNDD-361 down-dip and will allow the Inclusion of this deeper zone in an MRE upgrade. The deepest intersection of **4.0m** at **1.4** g/t AuEq (**1.4** g/t gold, **1.8** g/t silver) from 408.8m is a new zone that will require follow up drilling. The near surface intersection of **27.0m** at **0.3** g/t AuEq (**0.3** g/t gold, **0.7** g/t silver) from 115.4m including **2.0** at **1.4** g/t AuEq (**1.1** g/t gold, **6.8** g/t silver) which lies within the \$1800 MRE pit shell is a new zone of mineralisation.

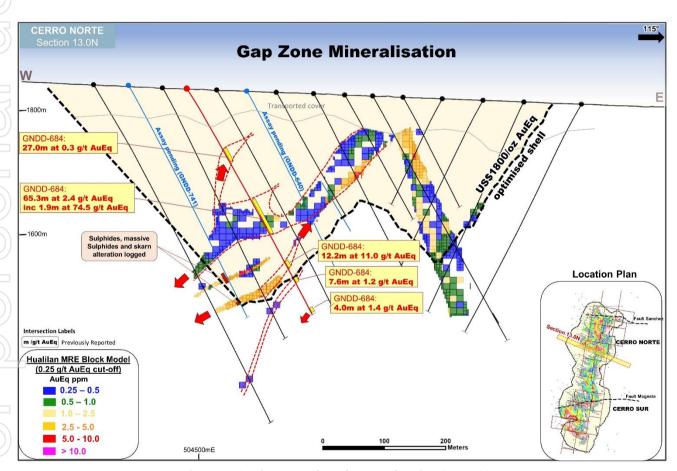


Figure 1 - Section central Verde Zone showing GNDD-684

GNDD-624 and GNDD-648

GNDD-624 was a 40 metre spaced infill hole targeting the Verde Zone between GNDD-287 and GNDD-387. The intersection of **64.0m** at **0.5** g/t AuEq (**0.4** g/t gold, **4.2** g/t silver, **0.2%** zinc) from 79.0m including **10.0m** at **1.4** g/t AuEq (**1.2** g/t gold, **9.7** g/t silver, **0.1%** lead **0.2%** zinc) is considerably thicker than that predicted from the MRE block model (Figure 2).

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Mr Scott Funston, Finance Director
Mr Fletcher Quinn, Chairman
Mr Sergio Rotondo, Exec. Director



The deeper Intersection of **4.3m at 1.8 g/t AuEq (0.9 g/t gold, 17.2 g/t silver, 0.3% lead 1.4% zinc)** from 107.0m is in line with the existing MRE model.

GNDD-648 was drilled as a 100 metre up-dip step-out from GNDD-284. The intersection of **28.0m** at **1.0** g/t AuEq (**0.9** g/t gold, **1.9** g/t silver, **0.1%** zinc) from 2.0m including **4.0m** at **2.7** g/t AuEq (**2.6** g/t gold, **3.0** g/t silver, **0.1%** zinc) from 10.0m and **4.0m** at **2.8** g/t AuEq (**2.7** g/t gold, **2.3** g/t silver, **0.1%** zinc) from 26.0m extends the Verde zone to surface, will upgrade the MRE and provide more confidence in near surface mineralisation. An infill hole GNDD-727 (assays pending) has been collared to test between GNDD-624 and GNDD-284 and then deeper towards the Magnata Fault Zone.

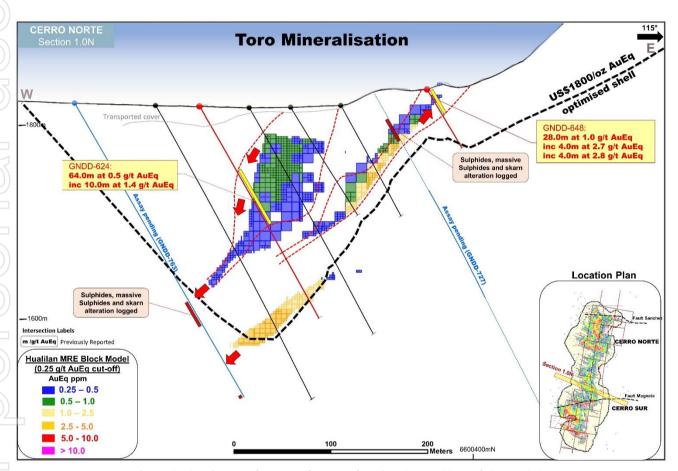


Figure 2 - Section southern Verde Zone showing GNDD-624 and GNDD-648

GNDD-661

GNDD-661 was collared with two objectives, a deep test of the Magnata Fault structure to the east, and to infill the intrusion hosted mineralisation north of the Magnata Fault. The hole intersected intrusion-hosted southern Verde Zone mineralisation over three zones from near surface. Intercepts included 94.0m at 0.7 g/t AuEq (0.6 g/t gold, 1.2 g/t silver, 0.1% zinc) from 17.0m including 7.5m at 2.3 g/t AuEq (2.2 g/t gold, 2.2 g/t silver, 0.1 % lead, 0.2% zinc), 4.0m at 3.0 g/t AuEq (2.9 g/t gold, 1.3 g/t silver, 0.2% zinc) from 135.0m, and 45.0m at 0.6 g/t AuEq (0.5 g/t gold, 6.3 g/t silver,



0.2% zinc) from 163.0 including **1.4m at 6.4 g/t AuEq (5.4 g/t gold, 50.1 g/t silver, 0.4% lead, 0.8% zinc)**. This zone of Intrusion-hosted mineralisation was significantly thicker than anticipated.

The hole failed to intersect significant mineralisation at depth on the eastern Magnata fault however drillhole GNDD-703 (assays pending) which was designed as deep test of the eastern Magnata Fault structure 60 metres to the west of GNDD-661 Is logged as intersecting sulphides and skarn alteration which is interpreted as mineralisation on the Magnata Fault.

GNDD-644

GNDD-644 was drilled as a 40 metre spaced infill hole in the northern Verde Zone between GNRC-091 (24.0 metres at 0.5 g/t AuEq) and GNDD-298 (21.0 metres at 0.8 g/t AuEq). The upper intercept of 43.0m at 0.9 g/t AuEq (0.8 g/t gold, 1.6 g/t silver, 0.2% lead, 0.1% zinc) from 42.0m including 2.0 at 7.3 g/t AuEq (7.1 g/t gold, 2.9 g/t silver, 0.7% lead 0.1% zinc) is significantly wider and higher in grade than the intercepts in the holes up and down-dip.

The deeper intercept of 12.5m at 1.5 g/t AuEq (0.9 g/t gold, 3.6 g/t silver, 1.1% zinc) extends the mineralisation 40 metres up-dip, within the optimised pit shell, from the intercept of and 2.0 metres at 1.5 g/t in GNDD-298. GNDD-671 (assays pending) has been completed to test with a deeper hole planned down-dip of GNDD-671 and an infill drill hole up-dip of GNDD-671.

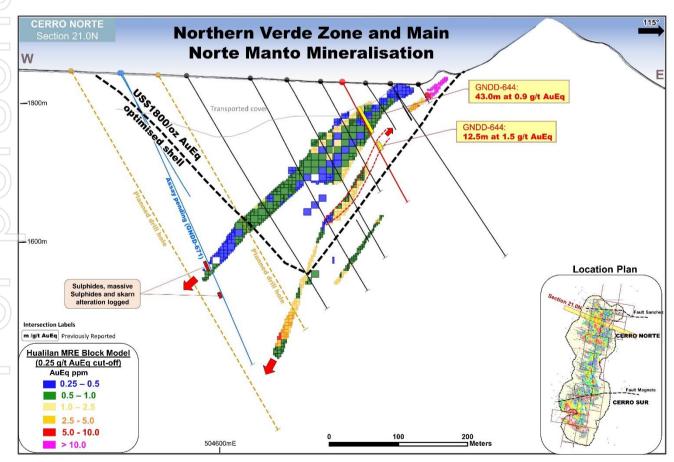


Figure 3 - Section norther Verde Zone showing GNDD-644 and planned drilling

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Mr Fletcher Quinn, Chairman
Mr Sergio Rotondo, Exec. Director



GNDD-697

GNDD-697 intersected **65.3m at 0.7 g/t AuEq (0.5 g/t gold, 2.0 g/t silver, 0.2% lead, 0.4% zinc)** from 41.0m including **4.5m at 3.3 g/t AuEq (2.4 g/t gold, 6.6 g/t silver, 0.8% lead 1.4% zinc)** and **3.6m at 5.0 g/t AuEq (2.9 g/t gold, 14.0 g/t silver, 2.0% lead 3.3% zinc)**.

GNDD-697 was drilled on the northern most section of drilling included in the current MRE. As can be seen from the MRE block model (Figure 4) prior to the MRE cut-off results had only been received for one Verde Zone drill hole on this section, GNDD-433 which intersected 22.0 metres at 0.7 g/t AuEq from 178.0m. GNDD-697 joins GNDD-563, which intersected 34.4 metres at 0.8 g/t AuEq, in confirming that mineralisation at Hualilan is strong and open to the north along strike.

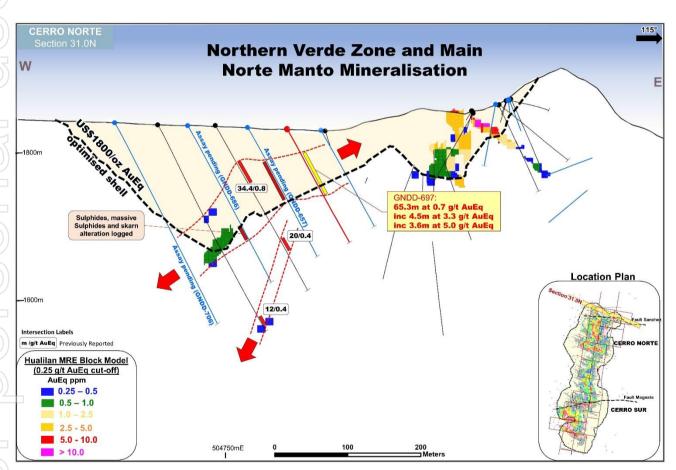


Figure 4 - Section northern Verde Zone showing GNDD-697

EXPLORATION DRILLING WEST OF THE MRE BOUNDARY

GNDD-557 was collared at the southern end of the Verde Zone approximately 750 metres west of the MRE boundary. The hole intersected two zones of mineralisation; 61.7m at 0.3 g/t AuEq (0.3 g/t gold, 1.2 g/t silver) from 271.3m including 0.9m at 2.0 g/t AuEq (1.9 g/t gold, 9.0 g/t silver) from 286.0m and 1.3 at 2.5 g/t AuEq (2.4 g/t gold, 6.7 g/t silver) from 330.3 hosted in interbedded intrusives and sediments.



The hole also intersected **25.6m at 0.5 g/t AuEq (0.4 g/t gold, 5.1 g/t silver, and 0.1% zinc)** from 460.0m. This deeper zone was hosted in fine grained sandstones and is interpreted as analogous to the sediment hosted mineralisation often located above the main Verde Zone mineralisation.

GNDD-576 was collared 300 metres north of GNDD-557 and almost 1 kilometre west of the existing MRE. This hole intersected intersecting 10.0m at 0.6 g/t AuEq (0.4 g/t gold, 20.6 g/t silver) from 182.0m including 2.0m at 1.4 g/t AuEq (0.8 g/t gold, 44.1 g/t silver, 0.1% zinc) hosted in intrusives and 0.6m at 0.8 g/t AuEq (0.4 g/t gold, 7.0 g/t silver, 0.1% lead, 0.5% zinc) from 551.7 at the contact between shale and limestone.

GNDD-585 was collared 1.5 kilometres to the north approximately 800 metres north-west of the MRE boundary. The hole intersected 3.0m at 1.8 g/t AuEq (1.6 g/t gold, 3.4 g/t silver, 0.1% lead, 0.3% zinc) from 244.0m including 1.0m at 4.7 g/t AuEq (4.2 g/t gold, 8.8 g/t silver, 0.2% lead, 0.8% zinc) hosted at the contact between shale and limestone.

GNDD-557, GNDD576, and GNDD-585 are three in a series of exploration holes collared up to one kilometre west of the Verde Zone mineralisation most of which intersected mineralisation. Given the area to the west of the Verde Zone is covered by recent cover this series of stratigraphic/exploration holes was drilled to allow the geology to be better integrated with the ground magnetic and IP data. These holes confirm that mineralisation remains open at least 1 kilometer west of the current MRE.

THE MAGNATA FAULT

The Magnata and Sanchez Faults are two east-west striking sub-vertical faults. The faults can be seen in outcrop and magnetic data extending for tens of kilometres to the east and west of Hualilan. The Magnata Fault is located at Cerro Sur approximately 1.5 kilometres south of the Sanchez Fault and separates into the M1 and M2 Magnata Faults, both of which host high-grade shoots.

The Magnata and Sanchez Faults were historically recognised as hosting mineralisation at Hualilan. The mineralising fluids are interpreted to have migrated from a source below or along strike, within the faults forming steeply dipping zones of mineralisation in the Magnata and Sanchez Faults. These fluids migrating up the faults also formed nearby replacement Manto-style high grade lenses, oriented parallel to the limestone beds, dipping to the west.

GNDD-685

GNDD-685 returned several important intersections that significantly open the mineralisation potential in the vicinity of the Magnata Fault.

The intersection of **8.4m** at **5.5** g/t AuEq (**4.5** g/t gold, **5.3** g/t silver, and **2.0%** zinc) from 362.4m including **2.6** metres at **17.4** g/t AuEq (**14.3** g/t gold, **16.1** g/t silver, and **6.3%** zinc) and **20.0m** at **1.1** g/t AuEq (**1.0** g/t gold, **0.9** g/t silver, and **0.2%** zinc) from 409.0m including **5.5m** at **2.6** g/t AuEq (**2.4** g/t gold, **1.1** g/t silver, and **0.4%** zinc) are in the Magnata Fault. This extends the mineralisation on



the Magnata Fault 100 metres below the current base of the MRE and importantly below GNDD-181 which was previously believed to close the Magnata Fault mineralisation at depth on this section. The Magnata Fault mineralisation remains strong and open at depth.

The deepest intersection in GNDD-685 was **1.9m at 23.3 g/t AuEq (20.0 g/t gold, 8.3 g/t silver, and 6.8% zinc)** from 651.2m. This is believed to correlate with a deep intersection of 20.0 metres at 1.5 g/t AuEq Including 2.8 metres at 9.8 g/t AuEq in GNDD-134 that is 250 metres below the MRE boundary. Given the depth of this intersection in GNDD-134 limited drilling had been conducted to follow up this zone of mineralisation which occurs at the contact between intrusive and limestone. GNDD-685 provides the first indication that this mineralisation is likely steeply dipping.

Surface mapping 700 metres vertically above these intersection in GNDD-685 and GNDD-134 indicates the presence of the same brecciated limestone-intrusive contact which is mineralised at surface. This limestone-intrusive contact is located approximately 200 metres south of the Magnata Fault.

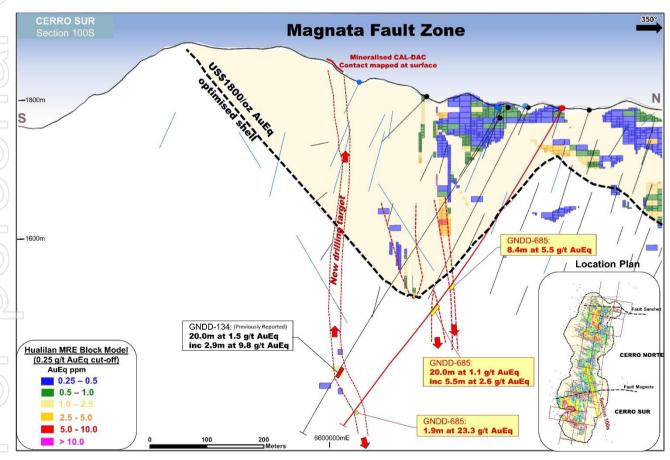


Figure 5 - Cross Section GNDD-685 Eastern Magnata Fault Zone and new target zone

This new drill target has at least 700 metres of vertical extent (Figure 5) and has been mapped over 200 metres of strike and remains open. A series of drill holes and surface channel samples have been completed to test this intrusive-limestone contact over the 700 vertical metres.

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Assays remain pending however several are logged as intersecting zones of massive and semi massive sulphides with skarn alteration which is consistent with high-grade gold mineralisation in surrounding drill holes.



Core Photo GNDD-685 including the Interval 651.1-653.0m (20.0 g/t gold, 8.3 g/t silver, and 6.8% zinc)



Ends

This ASX release was approved by the CEL Board

For further information contact:

Kris Knauer Scott Funston Media Enquiries
Managing Director Chief Financial Officer Jane Morgan
+61 411 885 979 +61 413 867 600 +61 405 555 618

kris.knauer@challengerex.com scott.funston@challengerex.com jm@janemorganmanagement.com.au

Previous announcements referred to in this release include:

21 Sept 2022 - Ongoing drilling at Challenger's Hualilan Gold Project continues to significantly extend the high-grade mineralisation

24 June 2022 - Drilling at Hualilan Solidifies Outlook to Uplift Maiden MRE

1 June 2022 - 2.1M Ounce AuEq Maiden Resource at Hualilan Gold Project



Table 2 - New Intercepts Reported this Release

Drill Holo	Erom	To			-	Pb	Zn	AuEa		Cuome v
Drill Hole	From	To	Interval	Gold	Ag	_	(%)	AuEq	Comments	Gram x
(#)	(m)	(m)	(m)	(g/t)	(g/t)	(%)		(g/t)	0.2 σ/+ ΔυΓα αυ+	metres
GNDD551A	353.0	361.0	8.0	0.3	0.1	0.0	0.0	0.3	0.2 g/t AuEq cut	2.6
GNDD557	271.3	333.0	61.7	0.3	1.2	0.0	0.0	0.3	0.2 g/t AuEq cut	18.0
inc	286.0	286.8	0.9	1.9	9.0	0.0	0.0	2.0	1.0 g/t AuEq cut	1.7
inc	330.3	331.6	1.3	2.4	6.7	0.0	0.0	2.5	1.0 g/t AuEq cut	3.1
and	460.0	485.6	25.6	0.4	5.1	0.0	0.1	0.5	0.2 g/t AuEq cut	12.3
GNDD564	40.0	47.0	7.0	0.2	11.6	0.2	0.2	0.4	0.2 g/t AuEq cut	2.8
and	453.0	457.0	4.0	0.5	2.0	0.0	0.0	0.6	0.2 g/t AuEq cut	2.3
and	484.0	486.0	2.0	0.2	3.7	0.0	2.3	1.3	1.0 g/t AuEq cut	2.7
GNDD576	182.0	192.0	10.0	0.4	20.6	0.0	0.0	0.6	0.2 g/t AuEq cut	6.2
inc	188.0	190.0	2.0	0.8	44.1	0.0	0.1	1.4	1.0 g/t AuEq cut	2.8
and	551.7	552.3	0.6	0.4	7.0	0.1	0.5	0.8	0.2 g/t AuEq cut	0.5
GNDD585	244.0	247.0	3.0	1.6	3.4	0.1	0.3	1.8	0.2 g/t AuEq cut	5.4
inc	246.0	247.0	1.0	4.2	8.8	0.2	0.8	4.7	1.0 g/t AuEq cut	4.7
GNDD602	351.0	355.0	4.0	0.2	0.9	0.0	0.1	0.3	0.2 g/t AuEq cut	1.1
GNDD607	215.6	226.0	10.4	0.3	3.4	0.1	0.2	0.4	0.2 g/t AuEq cut	4.5
inc	215.6	216.3	0.7	0.6	15.3	0.8	2.2	2.0	1.0 g/t AuEq cut	1.4
and	348.5	350.0	1.5	2.6	21.1	0.1	0.8	3.2	0.2 g/t AuEq cut	4.8
inc	348.5	349.4	0.9	4.2	33.5	0.2	1.3	5.2	1.0 g/t AuEq cut	4.4
and	368.0	370.0	2.0	0.1	97.8	0.1	0.0	1.3	1.0 g/t AuEq cut	2.6
GNDD609	76.0	90.0	14.0	0.2	0.6	0.0	0.0	0.3	0.2 g/t AuEq cut	3.6
and	123.5	127.0	3.5	0.3	0.4	0.0	0.1	0.3	0.2 g/t AuEq cut	1.1
and	151.0	171.0	20.0	0.4	0.2	0.0	0.0	0.4	0.2 g/t AuEq cut	8.2
inc	165.0	171.0	6.0	0.9	0.4	0.0	0.1	1.0	1.0 g/t AuEq cut	5.8
and	359.2	359.7	0.5	0.8	13.2	0.0	1.2	1.5	1.0 g/t AuEq cut	0.8
GNDD611	68.0	72.0	4.0	0.3	1.7	0.1	0.3	0.5	0.2 g/t AuEq cut	1.8
and	213.3	215.4	2.0	0.5	6.9	0.1	0.4	0.7	0.2 g/t AuEq cut	1.5
inc	214.2	215.4	1.2	0.8	2.6	0.1	0.4	1.0	1.0 g/t AuEq cut	1.2
GNDD613	122.0	153.3	31.3	0.5	1.5	0.1	0.2	0.6	0.2 g/t AuEq cut	18.2
inc	144.0	153.3	9.3	1.3	3.3	0.2	0.6	1.6	1.0 g/t AuEq cut	15.0
GNDD615	142.0	146.0	4.0	0.2	1.9	0.0	0.0	0.2	0.2 g/t AuEq cut	1.0
and	176.0	189.2	13.2	0.2	5.1	0.0	0.0	0.2	0.2 g/t AuEq cut	3.0
and	231.0	272.4	41.4	0.2	0.6	0.0	0.0	0.2	0.2 g/t AuEq cut	9.0
and	288.0	290.9	2.9	0.2	1.5	0.0	0.5	0.5	0.2 g/t AuEq cut	1.3
and	321.0	344.0	23.0	0.4	3.3	0.0	0.1	0.5	0.2 g/t AuEq cut	11.0
inc	321.0	323.0	2.0	1.4	16.7	0.0	0.0	1.6	1.0 g/t AuEq cut	3.3
inc	339.0	340.1	1.1	2.3	4.5	0.4	0.7	2.8	1.0 g/t AuEq cut	3.1
and	360.0	376.0	16.0	0.2	1.9	0.0	0.1	0.3	0.2 g/t AuEq cut	4.2
GNDD617	49.0	53.0	4.0	0.8	2.1	0.1	0.2	0.9	0.2 g/t AuEq cut	3.6
inc	51.0	53.0	2.0	1.1	2.7	0.2	0.2	1.3	1.0 g/t AuEq cut	2.6
and	65.0	72.0	7.0	0.6	0.4	0.0	0.0	0.6	0.2 g/t AuEq cut	4.1
inc	67.0	69.0	2.0	1.0	0.4	0.0	0.0	1.1	1.0 g/t AuEq cut	2.1
and	82.0	89.1	7.1	0.3	4.3	0.0	0.0	0.4	0.2 g/t AuEq cut	2.8
GNDD619	149.5	157.0	7.5		0.8	0.0	0.1	0.4	0.2 g/t AuEq cut	1
				0.4	0.8	0.1	0.1		1.0 g/t AuEq cut	3.2
inc GNDD631	149.5	151.0	1.5	1.0				1.0		1.6
GNDD621	94.0	97.0	3.0	0.3	6.9	0.0	0.0	0.4	0.2 g/t AuEq cut	1.3
and	205.0	207.0	2.0	0.8	0.3	0.0	0.1	0.8	0.2 g/t AuEq cut	1.7

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Mr Sergio Rotondo, Exec. Director



and	296.4	299.5	3.1	0.2	2.5	0.1	0.2	0.3	0.2 g/t AuEq cut	1.0
GNDD623	157.0	161.0	4.0	0.2	2.3	0.1	0.2	0.4	0.2 g/t AuEq cut	1.5
and	195.0	213.0	18.0	0.3	0.7	0.1	0.2	0.4	0.2 g/t AuEq cut	7.4
inc	206.0	208.0	2.0	0.5	2.6	0.5	0.9	1.0	1.0 g/t AuEq cut	2.0
inc	212.0	213.0	1.0	1.5	1.9	0.3	0.6	1.9	1.0 g/t AuEq cut	1.9
GNDD624	79.0	143.0	64.0	0.4	4.2	0.0	0.2	0.5	0.2 g/t AuEq cut	31.0
inc	81.0	91.0	10.0	1.2	9.7	0.1	0.2	1.4	1.0 g/t AuEq cut	14.0
inc	107.0	111.3	4.3	0.9	17.2	0.3	1.4	1.8	1.0 g/t AuEq cut	7.6
GNDD632	NSI									0.0
GNDD634	94.6	106.5	11.9	0.3	2.5	0.0	0.1	0.3	0.2 g/t AuEq cut	3.7
and	116.6	120.7	4.1	1.3	3.7	0.2	0.6	1.6	0.2 g/t AuEq cut	6.8
inc	118.0	120.7	2.7	1.8	5.3	0.3	0.9	2.4	1.0 g/t AuEq cut	6.3
and	142.0	152.0	10.0	1.1	0.7	0.0	0.1	1.1	0.2 g/t AuEq cut	11.4
inc	142.0	145.1	3.1	2.8	1.1	0.1	0.2	2.9	1.0 g/t AuEq cut	9.0
GNDD638	317.0	321.0	4.0	0.7	3.1	0.1	0.3	0.9	0.2 g/t AuEq cut	3.5
inc	319.0	321.0	2.0	1.3	3.6	0.1	0.4	1.5	1.0 g/t AuEq cut	3.0
GNDD644	42.0	85.0	43.0	0.8	1.6	0.1	0.1	0.9	0.2 g/t AuEq cut	38.7
inc	49.0	51.0	2.0	7.1	2.9	0.7	0.1	7.3	1.0 g/t AuEq cut	14.7
inc	65.0	73.0	8.0	1.4	2.7	0.3	0.1	1.5	1.0 g/t AuEq cut	12.2
and	101.0	113.5	12.5	0.9	3.6	0.0	1.1	1.5	0.2 g/t AuEq cut	18.4
inc	101.0	112.4	11.4	1.0	3.8	0.0	1.1	1.6	1.0 g/t AuEq cut	18.0
GNDD648	2.0	30.0	28.0	0.9	1.9	0.0	0.1	1.0	0.2 g/t AuEq cut	27.4
inc	10.0	14.0	4.0	2.6	3.0	0.0	0.1	2.7	1.0 g/t AuEq cut	10.7
inc	26.0	30.0	4.0	2.7	2.3	0.0	0.0	2.8	1.0 g/t AuEq cut	11.2
GNDD655	19.9	29.7	9.8	0.4	4.6	0.2	0.9	0.9	0.2 g/t AuEq cut	8.7
inc	21.7	26.5	4.9	0.6	6.2	0.4	1.0	1.3	1.0 g/t AuEq cut	6.2
and	46.0	54.6	8.6	0.3	0.6	0.0	0.0	0.3	0.2 g/t AuEq cut	2.6
and	69.5	72.0	2.5	0.4	0.8	0.0	0.1	0.5	0.2 g/t AuEq cut	1.2
and	96.9	104.7	7.8	0.9	4.3	0.1	0.6	1.2	0.2 g/t AuEq cut	9.2
inc	96.9	101.8	4.9	1.2	6.4	0.1	0.8	1.7	1.0 g/t AuEq cut	8.4
GNDD656	70.0	86.0	16.0	0.1	0.9	0.0	0.1	0.2	0.2 g/t AuEq cut	3.1
GNDD661	17.0	111.0	94.0	0.6	1.2	0.0	0.1	0.7	0.2 g/t AuEq cut	63.8
inc	17.0	24.5	7.5	2.2	2.2	0.1	0.2	2.3	1.0 g/t AuEq cut	17.2
inc	44.0	46.0	2.0	1.0	0.3	0.0	0.1	1.0	1.0 g/t AuEq cut	2.1
inc	64.0	66.0	2.0	2.8	0.2	0.0	0.0	2.8	1.0 g/t AuEq cut	5.6
inc	74.0	76.0	2.0	8.8	0.2	0.0	0.1	8.8	1.0 g/t AuEq cut	17.6
inc	109.0	111.0	2.0	1.0	3.5	0.1	0.2	1.2	1.0 g/t AuEq cut	2.3
and	135.0	139.0	4.0	2.9	1.3	0.0	0.2	3.0	0.2 g/t AuEq cut	12.0
and	163.0	208.0	45.0	0.5	6.3	0.0	0.2	0.6	0.2 g/t AuEq cut	29.2
inc	183.0	185.0	2.0	1.1	4.5	0.0	0.0	1.2	1.0 g/t AuEq cut	2.4
inc	191.0	192.4	1.4	5.4	50.1	0.3	0.7	6.4	1.0 g/t AuEq cut	9.0
inc	197.6	202.0	4.4	2.0	10.5	0.1	1.0	2.6	1.0 g/t AuEq cut	11.4
GNDD664	37.5	38.3	0.8	2.8	47.2	0.1	0.3	3.5	1.0 g/t AuEq cut	2.6
and	57.8	62.0	4.2	0.1	4.2	0.1	0.8	0.5	0.2 g/t AuEq cut	2.1
inc	57.8	58.2	0.4	0.1	4.4	0.0	3.2	1.7	1.0 g/t AuEq cut	0.7
GNDD684	115.0	142.0	27.0	0.3	0.7	0.0	0.0	0.3	0.2 g/t AuEq cut	7.9
inc	125.0	127.0	2.0	1.1	6.8	0.2	0.3	1.4	1.0 g/t AuEq cut	2.7
and	209.0	274.3	65.3	2.3	1.7	0.1	0.2	2.4	0.2 g/t AuEq cut	156.9
inc	256.0	257.3	1.3	1.1	0.8	0.0	0.0	1.1	1.0 g/t AuEq cut	1.4

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inc	265.6	267.5	1.9	71.6	41.1	3.5	3.8	74.5	10 g/t AuEq cut	141.6
and	324.9	337.0	12.2	10.1	11.7	0.1	1.5	11.0	0.2 g/t AuEq cut	133.1
inc	324.9	333.1	8.3	14.8	17.1	0.1	2.2	16.0	10 g/t AuEq cut	132.4
and	354.5	362.0	7.6	1.1	0.7	0.0	0.1	1.2	0.2 g/t AuEq cut	8.8
inc	359.8	362.0	2.2	3.2	0.9	0.0	0.0	3.2	1.0 g/t AuEq cut	7.1
and	408.0	418.0	10.0	0.6	0.7	0.0	0.0	0.6	0.2 g/t AuEq cut	6.5
inc	408.0	412.0	4.0	1.4	1.8	0.0	0.0	1.4	1.0 g/t AuEq cut	5.6
GNDD685	3.0	57.0	54.0	0.2	2.8	0.1	0.1	0.3	0.2 g/t AuEq cut	15.4
and	91.0	104.6	13.6	0.1	8.7	0.0	0.1	0.3	0.2 g/t AuEq cut	3.9
and	362.4	370.8	8.4	4.5	5.3	0.0	2.0	5.5	0.2 g/t AuEq cut	46.4
inc	362.4	365.0	2.6	14.3	16.1	0.0	6.3	17.4	1.0 g/t AuEq cut	45.2
inc	362.4	363.7	1.3	26.5	29.5	0.0	12.3	32.5	10 g/t AuEq cut	42.3
and	409.0	429.0	20.0	1.0	0.9	0.0	0.2	1.1	0.2 g/t AuEq cut	21.7
inc	413.0	418.5	5.5	2.4	1.1	0.0	0.4	2.6	1.0 g/t AuEq cut	14.3
inc	425.0	427.0	2.0	1.5	1.4	0.0	0.1	1.6	1.0 g/t AuEq cut	3.2
and	548.0	549.7	1.7	0.4	8.1	0.0	2.1	1.5	0.2 g/t AuEq cut	2.5
and	624.6	627.0	2.4	0.2	0.6	0.0	1.2	0.8	0.2 g/t AuEq cut	1.9
and	651.2	653.0	1.9	20.0	8.3	0.0	6.8	23.3	0.2 g/t AuEq cut	43.1
GNDD697	41.0	106.3	65.3	0.5	2.0	0.2	0.4	0.7	0.2 g/t AuEq cut	47.7
inc	52.5	57.0	4.5	2.4	6.6	0.8	1.4	3.3	1.0 g/t AuEq cut	15.0
inc	65.0	68.6	3.6	2.9	14.0	2.0	3.3	5.0	1.0 g/t AuEq cut	17.9

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.



About Challenger Exploration

Challenger Exploration Limited's (ASX: CEL) aspiration is to become a globally significant gold producer. The Company is developing two complementary gold/copper projects in South America with a maiden **2.1 million ounce gold Resource Estimate** recently announced for the Hualilan Gold Project in San Juan, Argentina. Three rigs are currently drilling at Hualilan with 2-rigs operating at the Company's El Guayabo project in Ecuador.

The Company strategy is for the 100% owned Hualilan Gold Project to provide a high-grade low capex operation in the near term while it prepares for much larger bulk gold operations at both Hualilan and El Guaybo in Ecuador.

- 1. Hualilan Gold Project, located in San Juan Province Argentina, is a near term development opportunity. It has extensive drilling with over 150 historical and 700 CEL drill-holes and the Company has released an Interim JORC 2012 Compliant resource of 2,133,065 ounces which remains open in most directions. This resource contains a Skarn component 6.3 Mt at 5.6 g/t AuEq for 1.1 Moz AuEg and an intrusion/sediment-hosted component of 41.5Mt at 0.8 g/t AuEg for 1.0 Moz AuEq. The resource was based on 126,000 metres of CEL's 264,000 metre drill program. The project was locked up in a dispute for the 15 years prior to the Company's involvement and as a consequence had seen no modern exploration until CEL acquired the project in 2019. In the past 2 years CEL has completed over 700 drill holes for more than 200,000 metres of drilling. Results have included 6.1m @ 34.6 g/t Au, 21.9 g/t Ag, 2.9% Zn, 67.7m @ 7.3 g/t Au, 5.7 g/t Ag, 0.6% Zn, and 63.3m @ 8.5 g/t Au, 7.6 g/t Ag, 2.8% Zn. This drilling intersected high-grade gold over 3.5 kilometres of strike and extended the known mineralisation along strike and at depth in multiple locations. Recent drilling has demonstrated this high-grade skarn mineralisation is underlain by a significant intrusion-hosted gold system with intercepts including 209.0m at 1.0 g/t Au, 1.4 g/t Ag, 0.1% Zn and 110.5m at 2.5 g/t Au, 7.4 g/t Au, 0.90% Zn in intrusives. CEL's current program which is fully funded will include an additional 60,000 metres of drilling, an updated JORC Compliant Mineral Resource Estimate, and Scoping Study.
- 2. El Guayabo Gold/Copper Project covers 35 sq kms in southern Ecuador and is located 5 kilometres along strike from the 22-million ounce Cangrejos Gold Project¹. Prior to CEL the project was last drilled by Newmont Mining in 1995 and 1997 targeting gold in hydrothermal breccias. Historical drilling demonstrated potential to host significant gold and associated copper and silver mineralisation. Historical drilling has returned a number of intersections including 156m @ 2.6 g/t Au, 9.7 g/t Ag, 0.2% Cu and 112m @ 0.6 % Cu, 0.7 g/t Au, 14.7 g/t Ag which have never been followed up. CEL's maiden drilling program confirmed the discovery of a major Au-Cu-Ag-Mo gold system spanning several zones of significant scale. results from CEL's maiden drill program included 257.8m at 1.4 g/t AuEq including 53.7m at 5.3 g/t AuEq and 309.8m at 0.7 g/t AuEq including 307.1m at 0.8 g/t AuEq, and 528.7m at 0.5 g/t AuEq from surface to the end of the hole including 397.1m at 0.6 g/t AuEq from surface. The Company has drilled five of fifteen regionally significant Au-soil anomalies with over 500 metres of mineralisation intersected at all anomalies, confirming the potential for a major bulk gold system at El Guayabo. The Company has two rigs on site completing an additional 25,000 metres of diamond core drilling designed to allow the reporting of a maiden JORC 2012 Compliant resource for the main GY-A discovery zone.

¹ Source : Lumina Gold (TSX : LUM) July 2020 43-101 Technical Report



Mineralisation Style	Mt (0.25 g/t AuEq cut-off)	Au (g/t)	Ag (g/t)	Zn (%)	Pb (%)	Au Eq (g/t)
Skarn (limestone hosted)	6.3	4.4	19.4	2.0	0.2	5.6
intrusion/sediment hosted	41.4	0.6	4.0	0.2	0.04	0.8
Mineralisation Style	Contained Metal	Au (Moz)	Ag (Moz)	Zn (kt)	Pb (kt)	Au Eq (kOz
Skarn (limestone hosted)		0.9	3.9	123	11	1.13
intrusion/sediment hosted		0.8	5.3	95	19	1.00
Total Contained metal		1.7	9.2	218	29	2.13

Table 3 Interim MRE reported as Skarn and Intrusion/sediment hosted components of mineralisation

Domain	Category	Mt	Au g/t	Ag g/t	Zn %	Pb %	AuEq g/t	AuEq (Mozs)
US\$1800 optimised shell	Indicated	18.7	1.1	5.4	0.41	0.07	1.3	0.80
> 0.25ppm AuEq	Inferred	25.0	1.0	5.6	0.39	0.06	1.2	1.00
Below US\$1800 shell >1.0ppm AuEq		4.0	1.9	11.5	1.04	0.07	2.6	0.33
Total Indicated and Inferred		47.7	1.1	6.0	0.45	0.06	1.4	2.13

Note: Some rounding errors may be present

Table 3 Total Interim MRE (Combined skarn and Intrusion hosted domains)

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.

The Mineral Resource Estimate for the Hualilan Gold Project was first announced to the ASX on 1 June 2022. 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.

JORC Code, 2012 Edition – Table 1 report template

Section 1 Sampling Techniques and Data - Hualilan Project

Criteria	JORC Code explanation	Commentary
Sampling techniques	 Nature and quality of sampling (eg cut channels, random chips, or specific specialised industry standard 	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.
	measurement tools appropriate to the minerals under investigation, such as down hole gamma sondes,	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.
	or handheld XRF instruments, etc). These examples should not be taken as limiting the broad meaning of	Channel samples are cut into underground or surface outcrop using a hand-held diamond edged cutting tool. Parallel saccuts 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.
	sampling. - Include reference to measures taken to ensure sample representivity and the appropriate calibration of any	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.
	the appropriate calibration of any measurement tools or systems used Aspects of the determination of mineralisation that are Material to	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.
	the Public Report In cases where 'industry standard'	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.
	work has been done this would be relatively simple (eg 'reverse	Unused pulps are returned from the laboratory to the Project and stored in a secure location, so they are available for an further analyses. Remaining drill core is stored undercover for future use if required.
	circulation drilling was used to obtain 1 m samples from which 3 kg	Visible gold observed has been observed in only 1 drill core sample only. Coarse gold is not likely to result in sample bias.
	was pulverised to produce a 30 g charge for fire assay'). In other cases, more explanation may be	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,
	required, such as where there is coarse gold that has inherent sampling problems. Unusual	Ag and Zn were also analysed. There is limited multielement data available. No information is available for RC drill techniques and sampling.
	commodities or mineralisation types	
	(eg submarine nodules) may warrant disclosure of detailed	
	information.	

Challenger Exploration Limited ACN 123 591 382 ASX: CEL 1,044.9m shares 10m options 120m perf shares 16m perf rights Australian Registered Office Level 1 1205 Hay Street West Perth WA 6005 **Directors**Mr Kris Knauer, MD and CEO
Mr Scott Funston, Finance Director
Mr Fletcher Quinn, Chairman
Mr Sergio Rotondo, Exec. Director

Criteria

JORC Code explanation

Commentary

Drilling techniques

 Drill type (eg core, reverse circulation, open-hole hammer, rotary air blast, auger, Bangka, sonic, etc) and details (eg core diameter, triple or standard tube, depth of diamond tails, facesampling bit or other type, whether core is oriented and if so, by what method, etc). CEL drilling of HQ3 core (triple tube) was done using various truck and track mounted drill machines that are operated by various drilling contractors based in Mendoza and San Juan. The core has not been oriented as the rock is commonly too broken to allow accurate core orientation.

CEL drilling of reverse circulation (RC) drill holes was done using a track mounted LM650 universal drill rig set up for reverse circulation drilling. Drilling was done using a 5.25 inch hammer bit.

Collar details for DD drill holes not included in the 01 June 2022 Resource Estimate are shown below in WGS84, zone 19s projection. Collar locations for drill holes are surveyed using DGPS following drilling. For drill collar and channel sample details for holes that are included in the 01 June 2022 Resource Estimate, see CEL ASX release of 01 June 2022.

Hole_id	East (m)	North (m)	Elevation (m)	Dip (°)	Azimuth (°)	Depth (m)
GNDD316 EXT	504121.0	6599927.0	1804.4	-60	115	217.40
GNDD359 EXT	504408.4	6601161.1	1827.6	-60	115	118.00
GNDD483	504127.1	6599924.1	1804.4	-50	115	380.00
GNDD487	504284.6	6601262.1	1844.7	-60	115	602.00
GNDD495	504339.7	6599517.9	1787.6	-60	115	167.00
GNDD497	504339.7	6599517.9	1787.6	-60	060	293.00
GNDD501	504467.0	6599500.0	1797.0	-60	060	290.00
GNDD505	503976.2	6599818.0	1802.9	-60	112	635.00
GNDD506	504635.7	6600966.9	1817.2	-60	115	515.00
GNDD508	504276.1	6600340.1	1818.3	-60	112	560.00
GNDD509	504491.3	6599599.8	1794.7	-60	115	232.00
GNDD510	504517.3	6600933.8	1827.7	-60	115	500.00
GNDD511	504526.0	6600059.0	1833.3	-10	110	175.00
GNDD516	504723.4	6600793.6	1821.3	-60	115	188.00
GNDD518	504468.5	6600287.0	1818.4	-60	170	332.00
GNDD519	504491.2	6599622.0	1794.8	-50	115	101.00
GNDD521	504907.6	6600928.4	1814.5	-60	295	392.00
GNDD525	504331.6	6600372.6	1819.5	-60	170	437.00
GNDD526	504529.0	6599963.0	1840.1	-15	115	190.00
GNDD528	505056.2	6600903.2	1813.2	-60	295	489.00
GNDD529	504539.1	6600347.5	1817.5	-60	170	452.00
GNDD530	504038.0	6600143.0	1815.0	-60	115	557.00
GNDD531	504431.9	6600929.5	1833.0	-60	115	461.00

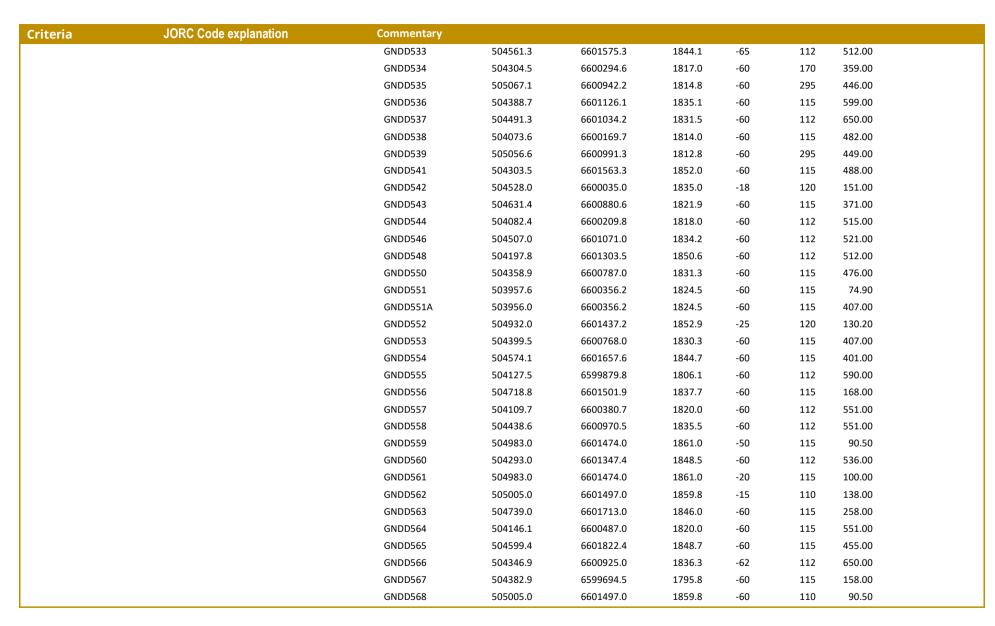
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Directors

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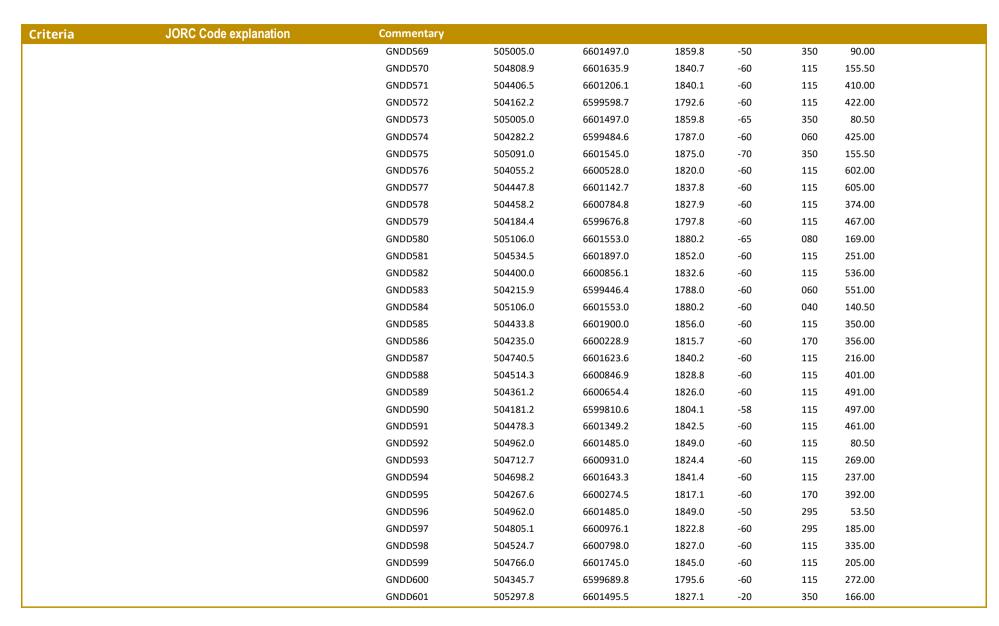


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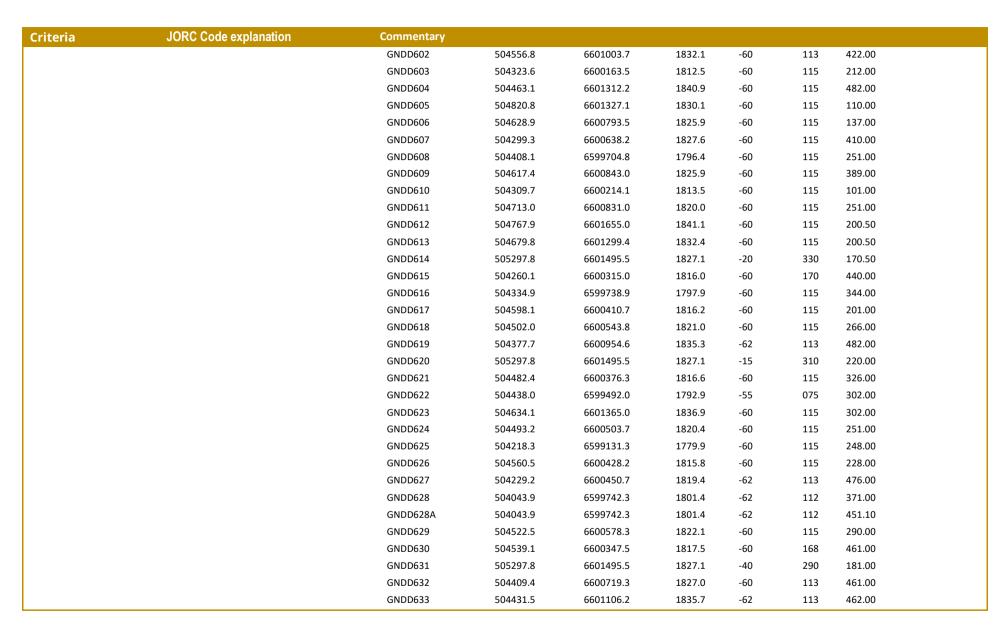


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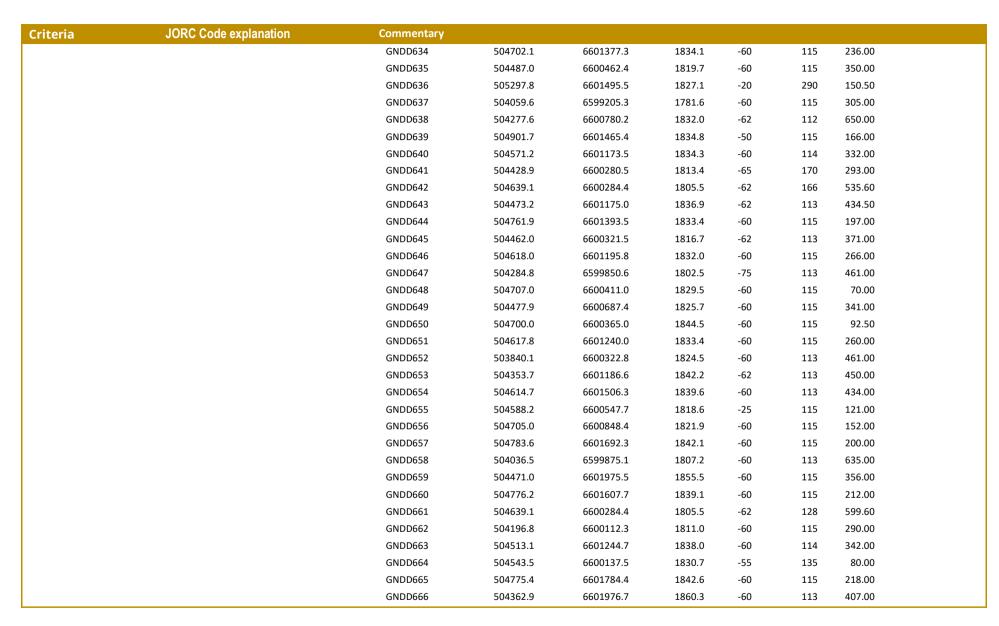


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Criteria	JORC Code explanation	Commentary						
		GNDD667	504798.1	6601509.2	1835.3	-65	115	131.00
		GNDD668	504237.4	6600093.4	1809.4	-60	115	255.80
		GNDD669	503877.3	6599290.3	1791.7	-60	115	446.00
		GNDD670	504478.0	6599646.0	1794.5	-10	130	125.00
		GNDD671	504473.6	6601528.4	1848.0	-62	113	461.00
		GNDD672	504539.4	6601541.4	1841.4	-60	112	500.00
		GNDD673	504638.0	6600392.0	1821.4	-58	145	620.00
		GNDD674	504711.0	6600272.0	1837.0	-65	115	74.00
		GNDD675	504126.9	6599880.5	1803.9	-65	113	536.00
		GNDD676	504719.0	6600247.0	1842.3	-65	115	61.00
		GNDD677	504732.8	6601539.7	1839.2	-60	115	185.00
		GNDD678	504417.3	6601863.6	1855.3	-65	115	350.00
		GNDD679	504662.8	6600270.1	1825.5	-40	115	116.00
		GNDD680	504690.2	6601692.0	1842.5	-60	115	251.00
		GNDD681	504269.8	6599901.7	1804.3	-67	115	226.00
		GNDD682	504402.0	6599064.0	1826.4	-15	125	194.50
		GNDD683	504471.8	6601794.0	1851.5	-62	113	464.00
		GNDD684	504486.7	6601212.9	1837.6	-60	113	420.00
		GNDD685	504638.0	6600392.0	1821.4	-60	165	671.00
		GNDD686	504702.2	6601730.3	1843.7	-60	115	242.00
		GNDD687	504289.0	6599892.8	1802.5	-60	115	195.40
		GNDD688	504392.5	6601389.2	1845.2	-62	113	476.00
		GNDD689	504371.3	6599854.4	1800.5	-60	115	140.00
		GNDD690	504700.3	6601245.7	1831.5	-60	115	177.00
		GNDD691	504664.3	6601659.7	1842.5	-60	114	304.00
		GNDD692	504407.3	6599749.3	1797.1	-57	115	143.00
		GNDD693	504523.7	6601460.4	1843.0	-60	113	415.00
		GNDD694	504402.0	6599064.0	1826.4	-55	125	114.50
		GNDD695	504371.6	6599810.1	1798.4	-60	115	161.00
		GNDD696	504543.7	6601362.8	1840.3	-60	114	339.00
		GNDD697	504826.9	6601672.1	1839.5	-60	115	176.00
		GNDD698	504402.0	6599064.0	1826.4	-55	030	161.00
		GNDD699	504414.2	6599834.4	1799.1	-60	115	116.00

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Criteria	JORC Code explanation	Commentary						
		GNDD700	504607.0	6601730.5	1846.4	-60	113	350.00
		GNDD701	504386.0	6599895.0	1801.2	-60	115	150.00
		GNDD702	504549.9	6600830.3	1826.3	-60	115	341.00
		GNDD703	504669.2	6600332.9	1809.1	-58	160	620.00
		GNDD704	504612.2	6601419.2	1839.4	-60	114	315.00
		GNDD705	504260.5	6599685.4	1797.6	-60	112	431.00
		GNDD706	504612.7	6601772.0	1848.7	-60	114	308.00
		GNDD707	504699.0	6600628.4	1821.3	-60	115	242.00
		GNDD708	504348.7	6599776.6	1797.6	-60	115	149.00
		GNDD709	504304.2	6600431.2	1817.7	-60	113	452.00
		GNDD710	504399.8	6601606.5	1850.8	-60	112	561.00
		GNDD711	504527.0	6601723.7	1847.2	-60	112	461.00
		GNDD712	504386.3	6600291.9	1815.0	-60	170	335.00
		GNDD713	504691.1	6601823.7	1847.4	-60	115	281.00
		GNDD714	504283.3	6600027.3	1806.3	-60	115	221.00
		GNDD715	504146.1	6600003.5	1808.8	-60	113	440.00
		GNDD716	504671.0	6599748.0	1906.7	-50	115	302.00
		GNDD717	504175.5	6599945.7	1809.7	-53	115	302.00
		GNDD718	504679.9	6601872.2	1850.5	-60	115	261.00
		GNDD719	504671.0	6599748.0	1906.7	-62	010	602.00
		GNDD720	504327.9	6600095.3	1809.3	-60	115	180.00
		GNDD721	504371.0	6600075.2	1808.1	-60	115	147.00
		GNDD722	504294.5	6599978.4	1805.9	-60	115	212.00
		GNDD723	504414.1	6600055.1	1807.4	-60	115	141.00
		GNDD724	504669.0	6599930.0	1907.2	-50	115	341.00
		GNDD725	504303.1	6601210.1	1841.1	-62	113	516.00
		GNDD726	504275.0	6599825.0	1802.4	-64	115	461.00
		GNDD727	504674.2	6600467.0	1825.8	-58	140	640.00
		GNDD728	504669.0	6599930.0	1907.2	-73	113	431.00
		GNDD729	504276.8	6599942.6	1807.2	-68	115	215.00
		GNDD730	504300.0	6599930.0	1804.1	-62	115	200.00
		GNDD731	504669.0	6599930.0	1907.2	-62	005	542.00
		GNDD732	504161.9	6599422.4	1787.8	-60	114	302.00

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Mr Kris Knauer, MD and CEO
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Criteria	JORC Code explanation	Commentary						
		GNDD733	503981.8	6599506.4	1791.9	-60	113	410.00
		GNDD734	504669.0	6599930.0	1907.2	-60	345	311.00
		GNDD735	504339.7	6601149.0	1840.5	-60	113	501.00
		GNDD736	504211.1	6599576.0	1792.4	-62	115	206.00
		GNDD737	504671.0	6599748.0	1903.0	-50	035	302.00
		GNDD738	503877.8	6599378.3	1794.6	-60	112	467.00
		GNDD739	504671.0	6599748.0	1906.7	-60	345	524.00
		GNDD740	503877.3	6599202.0	1787.2	-60	112	452.00
		GNDD741	504396.1	6601255.1	1840.7	-60	113	423.00
		GNDD742	503960.0	6598987.0	1775.1	-60	114	305.00
		GNDD743	504663.2	6600046.2	1871.0	-60	115	332.00
		GNDD744	503783.0	6599069.0	1785.4	-60	113	401.00
		GNDD745	504530.0	6601325.1	1837.7	-60	113	381.00
		GNDD746	504688.0	6600039.0	1872.5	-50	115	302.00
		GNDD747	503938.5	6598643.9	1765.8	-60	114	332.00
		GNDD748	504844.6	6601708.0	1837.2	-60	115	165.00
		GNDD749	504860.9	6601744.6	1837.6	-60	115	180.00
		GNDD750	504839.9	6601798.5	1838.9	-60	115	201.00
		GNDD751	504542.8	6601054.2	1830.8	-60	113	422.00
		GNDD752	504688.0	6600039.0	1872.5	-50	295	122.00
		GNDD753	504663.2	6600046.2	1871.0	-73	115	413.00
		GNDD754	504869.0	6601652.5	1835.9	-60	115	150.00
		GNDD755	504532.3	6601677.1	1845.9	-60	113	420.00
		GNDD756	504444.7	6601055.9	1834.1	-60	113	509.00
		GNDD757	504657.0	6600103.0	1850.3	-50	115	341.00
		GNDD758	504571.3	6601703.0	1844.2	-60	113	366.00
		GNDD759	504657.0	6600103.0	1850.3	-65	115	290.00
		GNDD760	504907.1	6600840.3	1815.8	-62	292	479.00
		GNDD761	504657.0	6600103.0	1850.3	-50	90	341.00
		GNDD762	504500.7	6601647.7	1845.5	-60	113	480.00
		GNDD763	504376.4	6600558.2	1821.9	-60	115	422.00
		GNDD764	504985.9	6600891.8	1814.3	-62	292	470.00
		GNDD765	504136.1	6599699.2	1797.8	-60	114	270.00

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		Historic drill hole collar data: Collar details for pre-CEL diamond core drilling (DD) and reverse circulation (RC) that is relied on for exploration and resource estimation is the CEL ASX release date 01 June 2022
Drill sample recovery	 Method of recording and assessing core and chip sample recoveries and results assessed. 	Drill core is placed into wooden boxes by the drillers and depth marks are indicated on wooden blocks at the end of each run. These depths are reconciled by CEL geologists when measuring core recovery and assessing core loss. Triple tube drilling has been being done by CEL to maximise core recovery.
	 Measures taken to maximise sample recovery and ensure representative nature of the samples. Whether a relationship exists 	RC sub-samples are collected from a rotary splitter mounted to the face sample recovery cyclone. A 2-4 kg sub-samples collected for each metre of RC drilling. Duplicate samples are taken at the rate of I every 25-30 samples using a riffle splitter to split out a 2-4 kg sub-sample. The whole sample recovered is weighed to measure sample recovery and consistency in sampling.
	between sample recovery and grade and whether sample bias may have occurred due to preferential loss/gain of fine/coarse material.	Channel samples have been weighed to ensure a consistency between sample lengths and weights. The channel sample are collected from saw-cut channels and the whole sample is collected for analysis. There is no correlation between sample length and assay values.
		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 influence 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 	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.
	 quantitative in nature. Core (or costean channel etc) photography. The total length and percentage of the relevant intersections logged. 	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. 	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.
	- If non-core whether riffled tube sampled rotary split etc and whether	Sample sizes are appropriate for the mineralisation style and grain size of the deposit.
	sampled wet or dry.	Sample intervals are selected based on lithology, alteration, and mineralization boundaries. Representative samples of a of the core are selected. Sample length averages 1.74m. Second-half core or ¼ core samples have been submitted for a

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Criteria JORC Code explanation

- 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 the insitu material collected including for instance results for field duplicate/second-half sampling.
- Whether sample sizes are appropriate to the grain size of the material being sampled.

Commentary

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.

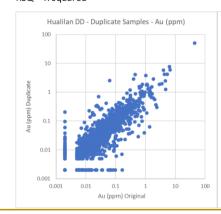
Competent drill core is cut longitudinally using a diamond saw for sampling of ½ the core. Softer core is split using a wide blade chisel or a manual core split press. The geologist logging the core, marks where the saw cut or split is to be made to ensure half-core sample representivity.

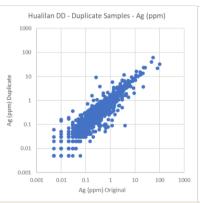
From GNDD073 and later holes, duplicate core samples consisting of two ¼ core samples over the same interval have been collected approximately every 30-50m drilled.

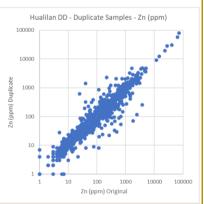
Duplicate core sample results and correlation plots (log scale for Au, Ag and Zn) are shown below:

	count	RSQ	mean		median		variance	
			original	duplicate	original	duplicate	original	duplicate
Au (ppm)	3,002	0.960	0.081	0.082	0.007	0.007	0.740	0.943
Ag (ppm)	3,002	0.691	0.56	0.50	0.17	0.16	9.05	3.86
Cd (ppm)	3,002	0.979	1.47	1.37	0.09	0.09	186.57	167.69
Cu (ppm)	3,002	0.433	14.86	13.80	3.30	3.20	4.8E+03	2.7E+03
Fe (%)	3,002	0.988	1.946	1.944	1.700	1.700	3.20	3.19
Pb (ppm)	3,002	0.942	67.6	64.8	14.1	13.8	2.2E+05	3.1E+05
S (%)	3,002	0.974	0.341	0.338	0.150	0.150	0.380	0.364
Zn (ppm)	3,002	0.976	271	258	74	72	4.4.E+06	4.1.E+06

RSQ = R squared

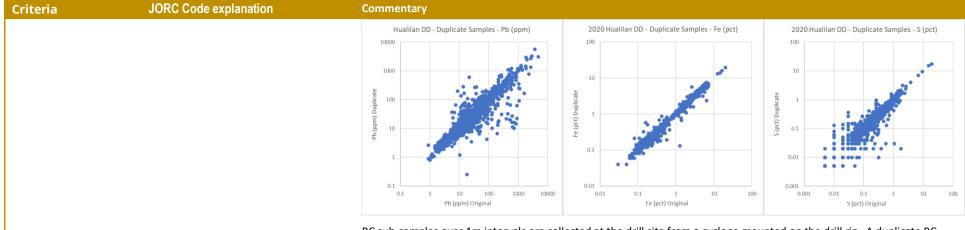






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RC sub-samples over 1m intervals are collected at the drill site from a cyclone mounted on the drill rig. A duplicate RC sample is collected for every 25-30m drilled.

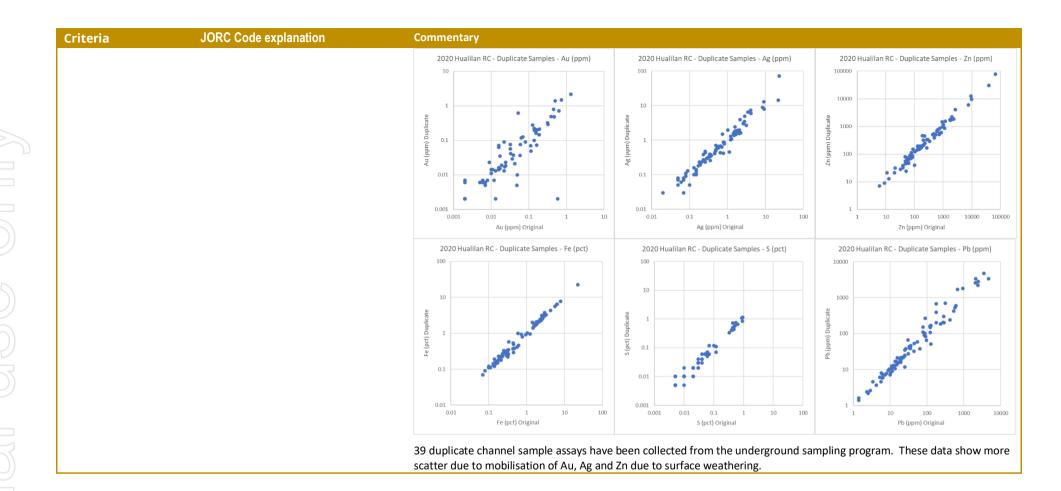
The duplicate RC sample results and correlation plots (log scale for Au, Ag and Zn) are shown below:

	count	RSQ	mean		median		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) SQ = R squared	85 d	0.977	3399	3234	158	177	2.5.E+08	2.1.E+08

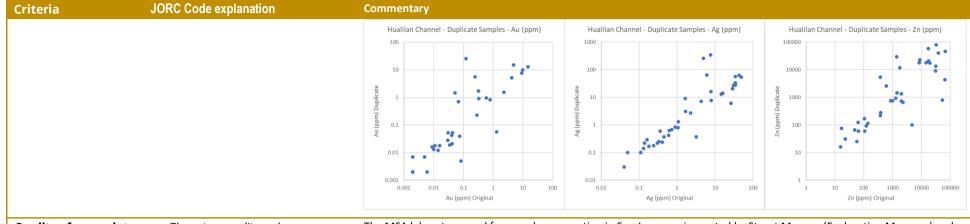
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Quality of assay data and laboratory tests

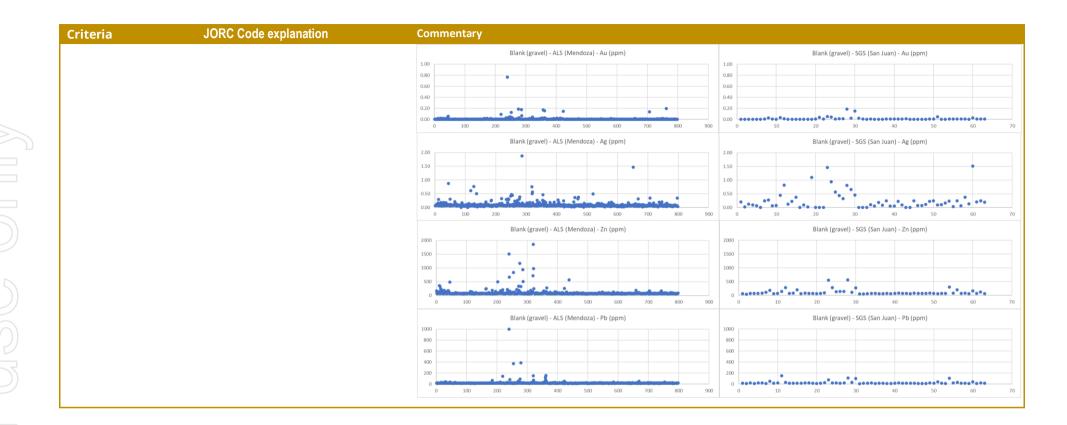
- The nature quality and appropriateness of the assaying and laboratory procedures used and whether the technique is considered partial or total.
- For geophysical tools spectrometers handheld XRF instruments etc the parameters used in determining the analysis including instrument make and model reading 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.

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 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.

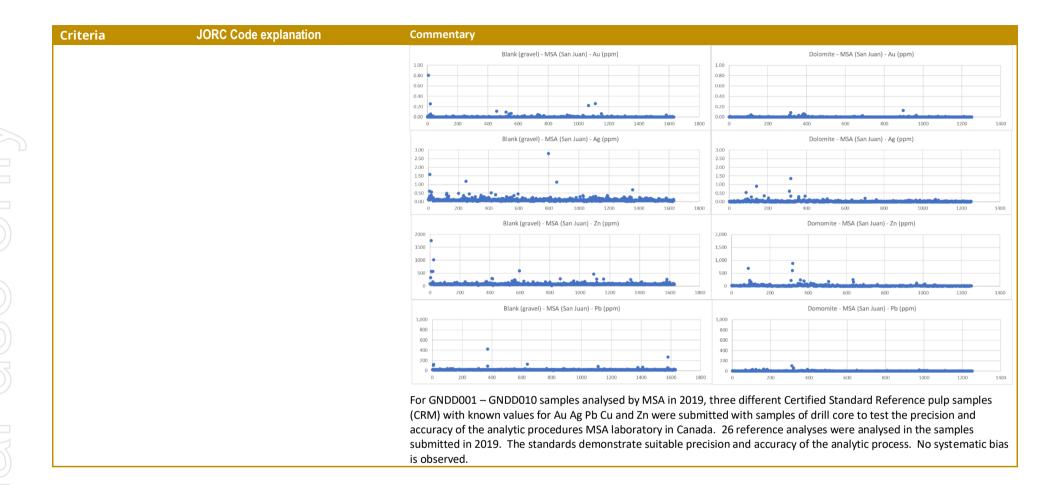
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Contact T: +61 8 6380 9235 E: admin@challengerex.com



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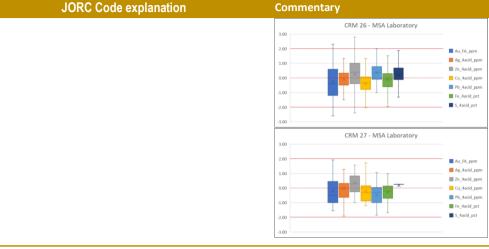
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Verification of sampling and assaying

Criteria

- 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 Deviation		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

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Criteria	JORC Code explanation	Commentary							
		Cd (ICP ppm)	68.5	68.8	12.4	12.8	162.4	159.3	0.9988
		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 set 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		Medi	an	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. Values below detection were set at zero

CEL have sought to twin and triplicate some of the historic and recent drill holes to check the results of previous

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Contact T: +61 8 6380 9235 E: admin@challengerex.com

Criteria	JORC Code explanation	Commentary
		exploration. A preliminary analysis of the twin holes indicates similar widths and grades for key elements assayed. The twin holes are: GNDD003 – DDH34 and 04HD08 GNRC110 – DDH53 GNDD144 – GNDD021 – 05HD39 GNRC107 – GNDD008/008A GNDD206 – DDH54 GNDD421 – GNDD424
Location of data points	 Accuracy and quality of surveys used to locate drill holes (collar and down-hole surveys) trenches mine workings and other locations used in Mineral Resource estimation. Specification of the grid system used. Quality and adequacy of 	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. 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. 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 holes have no down hole survey data due to drill hole collapse or blockage of the hole due to loss of
		drilling equipment. These are GNDD036, 197, 212, 283, 376, 423, 425, 439, 445 and 465. For these holes, a survey of the collar has been used with no assumed deviation to the end of the hole.
		All current and previous drill collar sites, Minas corner pegs and strategic surface points have been surveyed using DGPS to provide topographic control for the Project. In addition, AWD3D DTM model with a nominal 2.5 metre precision has been acquired for the project and greater surrounding areas. Drone-based topographic 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 	No regular drill hole spacing has been applied across the Project, although nominal 80m x 80m, 40m x 80m and 40m x 40m drill spacing is being applied to the drilling to define mineralised areas, 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. 80m x 80m drilling is designed for broad exploration of intrusion-hosted targets, whereas 40 m x 40m drilling is used to define and area that is expected to form part of a Mineral Resource Estimate in sedimentary and intrusive-hosted targets. Samples have not been composited for reporting.

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Criteria	JORC Code explanation	Commentary
Orientation of data in relation to geological structure	classifications applied. - Whether sample compositing has been applied. - 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.
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.

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Section 2 Reporting of Exploration Results

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

Criteria

status

JORC Code explanation

Mineral tenement and land tenure

- 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.

Granted mining leases (Minas Otorgadas) at the Hualilan Project

There are no royalties held over the tenements.

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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Criteria	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.	Application	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.	Application	14/02/1994	2.09

Requested Mining Leases (Minas Solicitados)

Name	Number	Status	Area (ha)
Elena	1124.328-G-2021	Application	2,799.24
Juan Cruz	1124.329-G-2021	Application	933.69
Paula (over "Lo Que Vendra")	1124.454-G-2021	Application	1,460.06
Argelia	1124.486-G-2021	Application	3,660.50
Ana Maria (over Ak2)	1124.287-G-2021	Application	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	Yes	Current	23.54
Marta	339.154-R-92	Yes	Current	478.50
Marta 1	339.153-R-92	Yes	Current	163.42
AK4	1124.299-R-18	Yes	Current	1,498.39
Solitario 1-5	545.604-C-94	Yes	Current	685.00
Solitario 1-4	545.605-C-94	Yes	Current	310.83
Solitario 1-1	545.608-C-94	Yes	Subject to Approval	TBA
Solitario 6-1	545.788-C-94	Yes	Subject to Approval	TBA
AGU 3	11240114-2014	Yes	Registered	1,500.00
AGU 5	1124.0343-2014	Yes	Registered	1,443.58
AGU 6	1124.0623-2017	Yes	Registered	1,500.00
AGU 7	1124.0622-S-17	Yes	Registered	1,500.00
Guillermina	1124.045-S-2019	Yes	Registered	2,921.05

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JORC Code explanation Criteria Commentary 1124.2478-71 El Petiso Yes 18.00 Registered Exploration Licence (Cateo) Farmin Agreements Name Number Transfrred to CEL **Status** Area (ha) 295.122-R-1989 Yes Current 1.882.56 228.441-R-1993 Yes Subject to Approval 2,800.00 545.880-0-1994 Yes Current 149.99 Exploration Licence (Cateo) Held (Direct Award) Grant Name Number Transfrred to CEL **Status** Area (ha) Date 1124.495-I-20 Yes 2,059.60 Aven Current 1124-248G-20 Yes Current 933.20 1124-188-G-20 (2 zones) Yes 327.16 Current 1124.313-2021 Yes Current 986.41 1124.564-G-2021 Yes 1,521.12 Current 1124.632-G-2022 Yes Current 4.287.38 There are no known impediments to obtaining the exploration licenses or operating the Project. **Exploration done** Intermittent historic sampling has produced a large volume of information and data including sampling, Acknowledgment and appraisal of geological maps, reports, trenching data, underground surveys, drill hole results, geophysical surveys, non-JORC by other parties exploration by other parties. resource estimates plus property examinations and detailed studies by multiple geologists. Prior to exploration by CEL, no work has been completed on the Project since 2006. There is at least 6 km of underground workings that pass through mineralised zones at Hualilan. Surveys of the workings are likely to be incomplete. Commonly incomplete records of the underground geology and sampling have been compiled and digitised as has sample data geological mapping adit exposures and drill hole results. Historic geophysical surveys exist but have been superseded by surveys completed by CEL. Historic drilling on or near the Hualilan Project (Cerro Sur and Cerro Norte combined) extends to over 150 drill holes. The key historical exploration drilling and sampling programs are: 1984 – Lixivia SA channel sampling & 16 RC holes (AG1-AG16) totalling 2,040m 1995 - Plata Mining Limited (TSE: PMT) 33 RC holes (Hua- 1 to 33) + 1,500 RC chip samples 1998 – Chilean consulting firm EPROM (on behalf of Plata Mining) systematic underground mapping and channel sampling 1999 – Compania Mineral El Colorado SA ("CMEC") 59 diamond core holes (DDH-20 to 79) plus 1,700m RC program

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Criteria	JORC Code explanation	Commentary
		- 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 (1996)
		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 on-going 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: 	Significant intersections reported by previous explorers and used in the Hualilan Mineral Resource Estimate of 01 June 2022 are included in the CEL ASX release date 01 June 2022. 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 reporting historic drill hole intersections.
	 easting and northing of the drill hole collar elevation or RL (Reduced Level – elevation above sea level in metres) of the drill hole collar 	The significant intersections from CEL drill holes and channel samples that have been used in the Mineral Resource Estimate are reported in the CEL ASX release date 01 June 2022. Significant intersections are listed below for drill holes that are not included in the Resource Estimate. Significant intersections are reported to a cut-off of 1.0 g/t AuEq (gold equivalent) unless otherwise indicated. Drill collar location is provided in the previous section.
	 dip and azimuth of the hole 	The following metals and metal prices have been used to report gold grade equivalent (AuEq): Au US\$ 1900 / oz

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Criteria

JORC Code explanation

Commentary

- 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.

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).

Hole_id	from (m)	to (m)	int (m)	Au (g/t)	Ag (g/t)	Pb (%)	Zn (%)	AuEq (g/t)	Note
GNDD375 EXT	490.70	491.20	0.50	1.1	13.0	0.00	0.64	1.6	
and	508.00	508.50	0.50	6.4	55.0	0.05	2.1	8.0	
and	521.35	524.70	3.35	1.5	15.7	0.02	0.58	1.9	
GNDD487	358.00	362.00	4.00	0.43	0.11	0.00	0.01	0.43	2
and	373.20	376.00	2.80	0.41	5.1	0.01	0.03	0.48	2
and	495.50	518.00	22.5	0.42	0.47	0.00	0.01	0.43	2
inc	497.00	497.50	0.50	4.0	5.8	0.00	0.01	4.1	
and	545.40	547.00	1.60	0.55	3.1	0.00	1.05	1.1	
GNDD495	NSI								
GNDD497	NSI								
GNDD501	35.00	53.25	18.2	0.22	32.7	0.02	0.07	0.65	2
inc	39.00	41.00	2.00	1.15	78.7	0.03	0.05	2.1	
inc	52.50	53.25	0.75	0.93	276	0.18	0.88	4.7	
and	187.65	189.00	1.35	2.5	2.0	0.00	0.02	2.5	2
inc	187.65	188.35	0.70	4.4	2.5	0.00	0.03	4.4	
GNDD505	443.00	445.00	2.00	0.29	25.9	0.04	0.41	0.80	2
GNDD506	116.10	118.20	2.10	0.02	4.5	0.09	1.9	0.98	2
inc	117.00	118.20	1.20	0.03	5.2	0.07	2.2	1.1	
and	205.40	216.00	10.6	0.87	1.1	0.00	0.10	0.93	2
inc	205.40	214.00	8.60	0.90	1.3	0.00	0.09	1.0	
and	238.40	273.60	35.2	0.32	1.4	0.01	0.49	0.57	2
inc	238.40	239.60	1.20	0.24	4.1	0.02	2.2	1.3	
inc	267.50	273.60	6.10	0.93	3.1	0.01	1.5	1.7	
and	294.00	302.00	8.00	0.42	0.52	0.01	0.07	0.46	2
and	318.00	323.50	5.50	0.34	0.71	0.01	0.09	0.39	2
and	430.35	438.65	8.30	0.29	0.26	0.02	0.03	0.31	2
GNDD508	89.75	91.10	1.35	0.85	2.01	0.10	0.32	1.0	•
and	125.00	128.40	3.40	0.24	8.6	0.00	0.19	0.43	2
and	167.00	191.00	24.0	0.33	0.47	0.04	0.06	0.37	2

Challenger Exploration Limited ACN 123 591 382 ASX: CEL

1,044.9m shares 10m options 120m perf shares 16m perf rights **Australian Registered Office** Level 1

1205 Hay Street West Perth WA 6005 Directors

Mr Kris Knauer, MD and CEO Mr Scott Funston, Finance Director Mr Fletcher Quinn, Chairman Mr Sergio Rotondo, Exec. Director Contact

Criteria	JORC Code explanation	Commentary									
		and	331.00	333.00	2.00	1.1	7.0	0.02	0.09	1.2	
		and	388.35	389.00	0.65	1.0	40.0	0.03	1.6	2.2	
		and	498.80	499.30	0.50	2.6	30.6	0.01	3.1	4.4	
		GNDD509	17.00	19.00	2.00	0.72	8.0	0.01	0.04	0.83	2
		and	61.00	63.00	2.00	2.0	15.5	0.00	0.01	2.2	
		and	223.75	227.30	3.55	2.3	2.5	0.00	0.03	2.4	
		GNDD510	167.00	169.00	2.00	1.4	0.30	0.00	0.01	1.4	
		and	224.00	284.00	60.0	0.24	2.0	0.03	0.07	0.31	2
		inc	238.00	240.00	2.00	0.78	7.8	0.06	0.44	1.1	
		and	348.00	350.00	2.00	3.7	5.9	0.44	1.2	4.4	
		and	430.00	447.00	17.0	0.91	0.43	0.00	0.00	0.91	2
		inc	439.60	447.00	7.40	1.8	0.32	0.00	0.00	1.8	
		and	461.00	465.00	4.00	0.40	0.82	0.00	0.01	0.41	2
		GNDD511	68.00	70.00	2.00	0.54	2.9	0.06	0.07	0.62	2
		and	130.00	132.00	2.00	0.26	26.5	0.03	0.07	0.62	2
		GNDD513	148.00	172.00	24.0	0.24	1.2	0.00	0.02	0.26	2
		and	186.00	188.00	2.00	0.96	15.2	0.23	0.30	1.3	
		and	239.00	243.00	4.00	0.34	1.0	0.00	0.01	0.36	2
		and	484.00	486.00	2.00	2.1	4.8	0.01	0.01	2.20	
		and	508.00	512.00	4.00	0.46	0.23	0.00	0.00	0.47	2
		and	532.00	542.00	10.0	0.32	1.0	0.04	0.08	0.37	2
		and	644.10	653.00	8.90	0.13	3.2	0.01	0.53	0.42	2
		inc	644.10	644.70	0.60	0.40	12.4	0.00	5.4	3.0	
		GNDD514	294.00	295.40	1.40	0.60	268	0.63	1.45	4.6	
		and	307.80	315.85	8.05	1.0	12.7	0.07	1.0	1.6	
		and	324.10	326.45	2.35	8.5	59.1	0.14	5.2	11.6	
		and	349.30	351.15	1.85	0.69	11.0	0.06	2.6	2.0	
		and	401.50	406.05	4.55	0.53	5.3	0.03	1.3	1.2	2
		inc	402.60	404.45	1.85	0.94	8.7	0.02	2.4	2.1	
		and	418.10	419.00	0.90	1.5	2.9	0.00	0.21	1.7	
		and	548.95	549.50	0.55	0.76	11.7	0.00	1.4	1.5	
		GNDD516	NSI								
		GNDD518	172.00	175.00	3.0	0.39	1.3	0.00	0.00	0.40	2
		and	183.50	185.00	1.50	1.5	25.0	0.58	0.79	2.3	
		and	201.00	206.00	5.00	0.83	2.5	0.17	0.21	1.0	2
		inc	203.00	204.25	1.25	2.2	0.87	0.05	0.14	2.2	
		GNDD519	NSI								
		GNDD521	82.00	86.00	4.00	0.26	0.20	0.00	0.0	0.26	2
		and	267.00	307.00	40.0	0.22	2.0	0.04	0.1	0.31	2
		inc	302.00	307.00	5.00	0.78	3.4	0.08	0.3	1.0	
		GNDD525	157.00	160.50	3.50	0.29	5.2	0.01	0.3	0.50	2
		and	268.00	274.00	6.00	0.62	1.6	0.10	0.2	0.73	2
		and	330.00	331.00	1.00	1.6	7.9	0.23	0.6	2.0	

Issued Capital 1,044.9m shares 10m options 120m perf shares 16m perf rights Australian Registered Office

Level 1 1205 Hay Street West Perth WA 6005 Directors

Mr Kris Knauer, MD and CEO Mr Scott Funston, Finance Director Mr Fletcher Quinn, Chairman Mr Sergio Rotondo, Exec. Director Contact

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Criteria	JORC Code explanation	Commentary									
		and	353.55	359.30	5.75	0.43	0.91	0.01	0.0	0.47	2
		inc	358.30	359.30	1.00	1.1	1.5	0.04	0.1	1.2	
		and	428.00	429.00	1.00	0.37	28.8	0.44	0.8	1.2	
		GNDD526	0.00	0.55	0.55	0.75	19.7	0.03	0.2	1.1	
		GNDD528	412.00	437.95	25.9	0.29	0.60	0.00	0.04	0.31	2
		inc	426.80	428.00	1.20	1.4	0.40	0.00	0.01	1.4	
		and	448.00	462.00	14.0	0.24	0.42	0.00	0.02	0.26	2
		GNDD529	144.00	150.00	6.00	0.42	1.0	0.06	0.07	0.48	2
		and	248.90	249.95	1.05	0.17	11.9	1.5	1.9	1.5	
		and	311.00	311.80	0.80	1.4	4.5	0.06	0.1	1.5	
		GNDD530	107.00	130.00	23.0	0.27	1.2	0.01	0.02	0.29	2
		and	159.00	213.00	54.0	0.30	2.0	0.01	0.06	0.35	2
		inc	196.00	198.90	2.90	1.8	12.2	0.05	0.51	2.2	
		and	357.50	386.00	28.5	5.0	23.9	0.02	0.03	5.3	
		inc	358.80	360.00	1.20	116	536	0.31	0.25	122	1
		GNDD531	283.00	295.00	12.0	0.20	2.3	0.01	0.03	0.25	2
		and	319.50	324.00	4.50	0.41	2.4	0.01	0.02	0.45	2
		inc	319.50	320.00	0.50	1.7	18.1	0.00	0.02	2.0	
		and	348.10	348.60	0.50	0.22	7.2	0.03	2.3	1.4	
		and	402.15	403.25	1.10	1.6	14.8	0.02	2.6	3.0	
		and	416.20	416.70	0.50	2.6	11.4	0.00	0.16	2.8	
		GNDD533	213.00	225.60	12.6	0.26	0.13	0.01	0.02	0.27	2
		inc	224.50	225.60	1.10	1.1	0.59	0.08	0.05	1.1	
		and	254.00	267.00	13.0	0.21	0.26	0.00	0.02	0.23	2
		and	362.00	363.35	1.35	67.0	101	0.04	15.0	75.1	1
		and	378.15	378.80	0.65	16.6	5.7	0.00	0.74	17.0	1
		and	403.50	404.00	0.50	3.0	32.6	0.04	1.4	4.0	
		and	473.00	494.00	21.0	0.43	0.89	0.00	0.01	0.44	2
		inc	481.00	483.00	2.00	1.2	0.33	0.00	0.01	1.2	
		GNDD534	88.00	92.00	4.00	0.18	1.4	0.06	0.19	0.29	2
		and	219.00	236.00	17.0	0.58	7.6	0.01	0.08	0.71	2
		inc	228.00	234.00	6.00	1.3	15.1	0.03	0.07	1.5	
		and	247.00	249.00	2.00	1.2	10.4	0.00	0.05	1.3	
		and	261.00	277.00	16.0	0.20	1.9	0.04	0.17	0.31	2
		and	312.00	321.35	9.35	0.22	1.8	0.04	0.08	0.29	2
		and	334.00	337.00	3.00	1.3	0.30	0.00	0.01	1.3	
		inc	334.00	335.00	1.00	3.5	0.63	0.01	0.02	3.5	
		GNDD535	88.00	90.00	2.00	0.69	0.13	0.00	0.01	0.69	2
		and	392.00	414.25	22.2	0.22	0.43	0.00	0.10	0.27	2
		inc	401.75	403.00	1.25	1.5	2.9	0.00	0.59	1.8	
		and	428.00	440.00	12.0	0.44	0.10	0.00	0.00	0.44	2
		GNDD536	188.85	213.00	24.1	0.74	1.7	0.02	0.23	0.87	2
		inc	201.20	203.00	1.80	2.9	13.4	0.01	2.2	4.1	

Issued Capital 1,044.9m shares 10m options 120m perf shares 16m perf rights **Australian Registered Office** Level 1

1205 Hay Street West Perth WA 6005 Directors

Mr Kris Knauer, MD and CEO Mr Scott Funston, Finance Director Mr Fletcher Quinn, Chairman Mr Sergio Rotondo, Exec. Director Contact

T: +61 8 6380 9235 E: admin@challengerex.com

riteria	JORC Code explanation	Commentary									
		inc	211.00	213.00	2.00	4.4	0.13	0.00	0.01	4.4	
		and	240.50	252.70	12.2	0.40	0.38	0.00	0.01	0.41	2
		and	508.30	512.00	3.70	1.0	1.7	0.03	0.40	1.2	
		inc	508.30	510.05	1.75	1.7	1.3	0.02	0.15	1.8	
		and	552.00	558.60	6.60	4.2	50.0	0.01	3.4	6.4	
		inc	556.80	558.60	1.80	14.2	183	0.04	12.5	22.1	1
		inc	556.80	558.10	1.30	19.2	252	0.06	17.1	30.2	1
		GNDD537	78.00	94.30	16.30	0.30	1.2	0.01	0.02	0.32	2
		and	144.00	150.00	6.00	0.24	0.64	0.03	0.03	0.27	2
		and	308.00	336.50	28.50	0.21	1.0	0.02	0.05	0.25	2
		GNDD538	115.50	122.00	6.50	0.18	3.4	0.05	0.08	0.27	2
		and	134.70	141.00	6.30	0.45	1.4	0.01	0.03	0.48	2
		and	176.00	186.00	10.00	1.02	0.7	0.00	0.01	1.0	2
		inc	182.00	184.00	2.00	3.11	0.7	0.00	0.01	3.1	
		and	198.50	200.00	1.50	1.60	1.7	0.00	0.06	1.6	
		and	331.00	410.00	79.00	0.21	1.3	0.03	0.05	0.26	2
		inc	404.00	405.00	1.00	4.0	11.2	0.60	1.1	4.7	
		GNDD539	315.00	321.00	6.00	0.22	0.11	0.00	0.01	0.22	2
		and	373.00	394.00	21.00	0.88	1.0	0.00	0.04	0.91	2
		inc	379.00	381.00	2.00	1.8	0.21	0.00	0.01	1.8	
		inc	388.00	392.00	4.00	1.7	2.4	0.00	0.01	1.7	
		and	410.00	426.00	16.00	0.30	0.14	0.00	0.01	0.31	2
		inc	424.00	426.00	2.00	1.07	0.53	0.00	0.01	1.1	
		GNDD540	134.00	186.50	52.5	0.29	5.1	0.00	0.06	0.38	2
		inc	136.60	137.40	0.80	0.77	49.5	0.03	0.14	1.5	
		inc	150.00	152.00	2.00	1.2	19.4	0.00	0.08	1.5	
		and	224.00	254.20	30.2	0.40	4.5	0.06	0.26	0.57	2
		inc	234.00	236.00	2.00	3.8	41.8	0.17	2.4	5.4	
		and	309.15	311.65	2.50	4.0	67.5	0.45	7.5	8.1	
		GNDD541	398.00	399.60	1.60	0.72	0.01	0.00	0.00	0.72	2
		and	436.00	441.00	5.00	0.07	62.3	0.06	0.10	0.88	2
		inc	439.90	441.00	1.10	0.24	222	0.18	0.35	3.1	
		and	464.20	464.70	0.50	1.4	48.7	0.00	3.7	3.7	
		GNDD542	NSI								
		GNDD543	90.30	106.00	15.7	0.18	1.7	0.01	0.1	0.24	2
		and	179.60	181.00	1.40	0.87	1.2	0.16	0.4	1.1	
		GNDD544	48.00	58.60	10.6	0.10	3.6	0.23	1.0	0.65	2
		inc	57.00	58.60	1.60	0.12	11.0	0.91	3.5	2.1	
		and	152.00	160.00	8.00	0.22	1.4	0.00	0.0	0.25	2
		and	299.00	318.00	19.0	0.25	1.0	0.00	0.0	0.27	2
		and	333.45	338.00	4.55	0.31	1.8	0.00	0.0	0.34	2
		and	409.00	410.40	1.40	1.1	12.0	0.13	0.6	1.5	
		and	422.00	426.00	4.00	0.43	2.9	0.07	0.0	0.49	2

1,044.9m shares 10m options 120m perf shares 16m perf rights Australian Registered Office Level 1

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		GNDD546	55.00	59.00	4.00	1.2	0.34	0.00	0.0	1.2	
		and	134.00	138.00	4.00	0.48	0.40	0.00	0.0	0.48	2
		and	316.00	330.00	14.00	0.56	1.8	0.03	0.2	0.66	2
		inc	326.00	328.00	2.00	1.8	2.9	0.01	0.1	1.9	
		and	437.00	439.35	2.35	2.2	8.1	0.00	0.1	2.3	2
		inc	438.30	438.80	0.50	9.7	35.7	0.01	0.0	10.2	1
		GNDD548	NSI								
		GNDD549	2.00	17.50	15.5	0.31	5.9	0.01	0.05	0.41	2
		and	28.10	39.00	10.9	4.0	71.5	0.51	0.81	5.3	
		inc	29.20	31.75	2.55	15.4	245	1.7	2.1	19.4	1
		inc	29.80	30.85	1.05	31.1	381	2.8	3.2	37.4	1
		inc	37.00	39.00	2.00	1.6	44.3	0.60	0.32	2.3	
		GNDD550	373.30	377.70	4.40	1.0	16.0	0.03	4.5	3.3	
		inc	374.00	377.70	3.70	1.1	18.7	0.03	5.4	3.8	
		and	425.00	427.10	2.10	3.7	27.0	0.01	1.7	4.8	
		and	437.50	443.00	5.50	0.49	15.3	0.02	3.3	2.2	
		GNDD551A	353.0	361.0	8.0	0.3	0.1	0.0	0.0	0.3	2
		GNDD552	2.20	36.00	33.8	0.75	12.1	0.10	0.15	1.0	2
		inc	9.00	12.35	3.35	6.0	82.4	0.80	0.58	7.4	
		inc	11.40	12.35	0.95	15.6	254	1.1	0.07	18.9	1
		GNDD553	300.00	306.00	6.00	0.21	1.1	0.10	0.18	0.33	2
		and	323.50	325.35	1.85	2.2	11.2	0.02	1.0	2.9	
		and	343.00	343.50	0.50	0.19	5.8	0.07	2.1	1.2	
		GNDD554	232.90	240.00	7.10	0.26	0.62	0.08	0.18	0.37	2
		and	259.90	306.00	46.10	0.79	0.86	0.02	0.12	0.86	2
		inc	259.90	261.00	1.10	0.89	2.2	0.21	0.45	1.2	
		inc	272.50	279.00	6.50	3.7	2.6	0.01	0.31	3.9	
		inc	286.20	287.30	1.10	1.3	0.51	0.01	0.12	1.3	
		inc	295.40	296.65	1.25	1.1	2.0	0.06	0.13	1.2	
		and	318.80	323.00	4.20	0.43	0.16	0.00	0.01	0.43	2
		and	338.00	362.00	24.00	0.88	0.80	0.00	0.05	0.91	2
		inc	344.70	350.20	5.50	2.8	1.9	0.00	0.21	2.9	
		GNDD555	68.55	69.10	0.55	0.03	79.0	0.09	0.12	1.1	
		and	284.00	288.00	4.00	0.37	4.4	0.13	0.51	0.69	
		and	314.00	327.70	13.7	0.29	8.0	0.25	0.76	0.79	
		inc	314.00	316.00	2.00	0.32	34.9	0.23	0.72	1.1	
		inc	326.85	327.70	0.85	1.0	32.5	3.3	10.1	6.7	
		and	468.70	470.00	1.30	1.0	19.5	0.01	2.7	2.4	
		and	481.10	482.55	1.45	0.59	11.5	0.04	2.2	1.7	
		and	489.75	490.25	0.50	0.23	6.0	0.05	1.7	1.1	
		and	495.00	498.70	3.70	0.90	11.3	0.01	1.2	1.6	
		inc	496.35	498.70	2.35	1.1	15.6	0.01	1.6	2.0	
		and	520.85	522.50	1.65	1.3	16.5	0.00	0.20	1.6	2

Issued Capital 1,044.9m shares 10m options 120m perf shares 16m perf rights Australian Registered Office Level 1

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T: +61 8 6380 9235 E: admin@challengerex.com

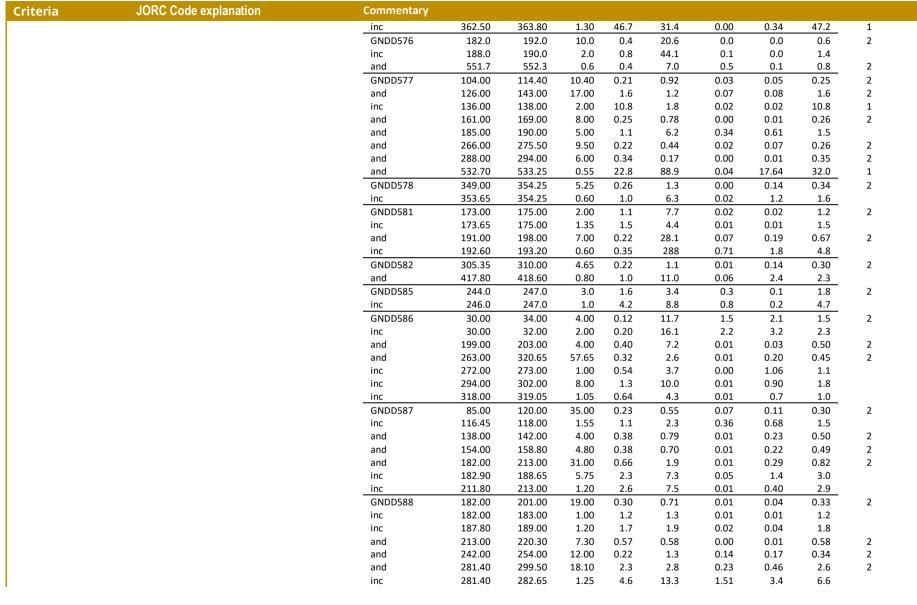
riteria	JORC Code explanation	Commentary									
		inc	521.80	522.50	0.70	2.3	26.7	0.00	0.42	2.8	
		and	531.80	532.40	0.60	9.4	19.8	0.02	1.6	10.4	1
		and	538.80	539.55	0.75	1.7	20.0	0.00	0.92	2.4	
		GNDD556	83.20	97.00	13.8	0.35	1.3	0.09	0.14	0.45	2
		inc	86.00	87.50	1.5	1.0	1.1	0.09	0.17	1.1	
		inc	94.60	95.80	1.2	1.0	2.2	0.11	0.17	1.1	
		and	115.00	124.00	9.00	0.25	0.35	0.03	0.10	0.30	2
		GNDD557	271.3	333.0	61.7	0.3	1.2	0.0	0.0	0.3	2
		inc	286.0	286.8	0.9	1.9	9.0	0.0	0.0	2.0	
		inc	330.3	331.6	1.3	2.4	6.7	0.0	0.0	2.5	
		and	460.0	485.6	25.6	0.4	5.1	0.1	0.0	0.5	2
		GNDD558	310.00	314.00	4.00	0.25	0.62	0.05	0.11	0.32	2
		and	348.00	353.00	5.00	0.25	1.1	0.10	0.14	0.35	2
		and	380.00	382.50	2.50	0.29	0.88	0.04	0.29	0.45	2
		GNDD559	14.00	18.00	4.00	0.23	0.45	0.01	0.10	0.28	2
		GNDD560	407.00	409.00	2.00	0.55	0.74	0.00	0.00	0.56	2
		and	483.40	486.00	2.60	0.15	4.9	0.05	0.05	0.25	2
		GNDD561	NSI								
		GNDD563	59.00	93.40	34.4	0.46	2.0	0.23	0.48	0.75	2
		inc	76.00	82.30	6.30	1.1	7.7	1.1	2.2	2.4	
		inc	90.00	92.00	2.00	3.0	0.39	0.04	0.05	3.1	
		and	125.00	128.10	3.10	0.43	0.57	0.02	0.07	0.48	2
		and	148.00	154.00	6.00	0.11	2.0	0.07	0.25	0.26	2
		and	182.00	202.00	20.0	0.31	1.7	0.04	0.07	0.37	2
		inc	184.00	184.50	0.50	5.1	16.8	1.2	2.1	6.5	
		GNDD564	40.0	47.0	7.0	0.2	11.6	0.2	0.2	0.4	2
		and	453.0	457.0	4.0	0.5	2.0	0.0	0.0	0.6	2
		and	484.0	486.0	2.0	0.2	3.7	2.3	0.0	1.3	
		GNDD565	NSI								
		GNDD566	434.05	452.25	18.20	0.05	0.65	0.00	0.3	0.19	2
		and	608.15	608.65	0.50	6.4	79.8	0.00	0.81	7.7	
		GNDD570	55.80	78.00	22.20	0.60	3.7	0.36	0.43	0.91	2
		inc	55.80	57.00	1.20	0.60	2.8	0.41	0.80	1.1	
		inc	63.00	70.30	7.30	1.4	9.0	0.82	0.41	1.8	
		and	95.00	105.00	10.00	0.33	1.4	0.02	0.15	0.43	2
		inc	103.00	105.00	2.00	0.94	2.6	0.08	0.32	1.1	
		GNDD571	213.00	260.00	47.00	0.34	1.1	0.00	0.08	0.39	2
		and	280.00	312.00	32.00	0.19	0.71	0.00	0.02	0.21	2
		and	328.00	338.00	10.00	0.59	0.49	0.00	0.01	0.61	2
		inc	334.00	336.00	2.00	1.6	1.4	0.00	0.01	1.6	
		and	356.00	367.00	11.00	9.0	5.7	0.00	0.05	9.1	2
		inc	356.00	363.80	7.80	12.5	7.9	0.00	0.06	12.6	-
		inc	357.90	359.40	1.50	15.7	9.7	0.00	0.01	15.8	1
Fynloration Limit	ted Issued Canital Aus	tralian Registered Office	Directors		Contac	-+					

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riteria	JORC Code explanation	Commentary									
		inc	289.70	290.70	1.00	32.6	18.1	1.57	1.9	34.0	1
		inc	298.85	299.50	0.65	2.1	4.4	0.33	1.9	3.1	
		and	314.00	401.00	87.00	0.67	1.4	0.00	0.01	0.69	2
		inc	315.00	323.00	8.00	2.9	3.4	0.01	0.01	3.0	
		inc	331.00	341.00	10.00	1.2	1.6	0.00	0.05	1.3	
		inc	379.00	381.00	2.00	2.2	0.4	0.00	0.00	2.3	
		inc	399.00	401.00	2.00	1.0	0.30	0.00	0.00	1.0	
		GNDD589	394.00	395.00	1.00	4.2	8.5	1.0	0.83	4.9	
		and	266.00	269.25	3.25	0.59	6.3	0.09	0.24	0.79	2
		and	273.80	274.40	0.60	0.93	9.6	0.03	0.14	1.1	
		GNDD591	224.00	238.00	14.00	1.2	0.91	0.02	0.0	1.2	2
		inc	229.25	232.00	2.75	4.4	3.5	0.05	0.1	4.5	
		inc	236.00	238.00	2.00	1.3	0.48	0.02	0.0	1.3	
		and	250.00	254.00	4.00	1.7	3.7	0.07	0.4	2.0	2
		inc	253.30	254.00	0.70	8.80	17.7	0.39	2.2	10.1	
		and	382.70	386.00	3.30	4.6	12.4	0.02	1.3	5.4	2
		inc	382.70	383.40	0.70	20.5	55.7	0.01	5.6	23.8	1
		and	425.00	429.60	4.60	0.53	0.63	0.00	0.01	0.5	2
		inc	429.00	429.60	0.60	3.1	0.56	0.00	0.02	3.1	
		and	436.40	437.00	0.60	1.4	13.1	0.00	2.3	2.6	
		GNDD593	105.50	124.00	18.50	0.16	2.2	0.00	0.08	0.23	2
		and	139.00	141.00	2.00	0.68	0.92	0.00	0.10	0.74	2
		and	153.00	164.00	11.00	0.83	1.7	0.02	0.10	0.90	2
		inc	153.00	157.00	4.00	1.7	4.0	0.05	0.20	1.8	
		GNDD594	104.00	116.00	12.00	0.72	1.8	0.21	0.51	1.0	2
		inc	108.00	110.00	2.00	3.1	6.5	0.48	1.5	3.9	
		and	162.00	163.40	1.40	2.1	0.30	0.00	0.01	2.1	
		and	198.00	204.00	6.00	0.63	3.3	0.02	0.13	0.73	2
		inc	198.00	198.50	0.50	1.7	3.3	0.12	0.32	2.0	
		GNDD595	198.35	212.10	13.75	0.32	2.5	0.00	0.02	0.36	2
		and	226.00	247.20	21.20	0.58	4.0	0.06	0.14	0.71	2
		inc	230.00	231.30	1.30	1.2	3.6	0.10	0.40	1.5	
		inc	240.45	242.00	1.55	3.2	20.3	0.28	0.86	3.9	
		and	266.00	305.80	39.80	0.26	2.9	0.08	0.30	0.45	2
		inc	266.00	268.00	2.00	1.6	8.5	0.01	0.04	1.7	
		inc	304.45	305.80	1.35	1.2	28.5	2.1	8.0	5.7	
		and	375.20	382.10	6.90	0.28	3.8	0.08	0.31	0.48	2
		inc	381.35	382.10	0.75	2.3	30.8	0.17	2.3	3.8	
		GNDD597	NSI								
		GNDD598	114.85	120.35	5.50	0.41	1.6	0.06	0.06	0.47	2
		inc	114.85	115.65	0.80	1.0	3.0	0.17	0.16	1.1	_
		and	168.00	240.00	72.00	0.24	1.0	0.01	0.10	0.30	2
		inc	204.00	206.00	2.00	1.4	0.86	0.00	0.00	1.4	-

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Criteria	JORC Code explanation	Commentary									
		and	253.00	271.00	18.00	0.34	0.62	0.00	0.01	0.35	2
		and	283.00	295.00	12.00	0.40	1.5	0.00	0.01	0.42	2
		GNDD599	NSI								
		GNDD602	351.0	355.0	4.0	0.2	0.9	0.1	0.0	0.3	2
		GNDD603	61.00	85.90	24.90	0.20	7.7	0.00	0.0	0.31	2
		inc	81.00	83.00	2.00	0.88	17.4	0.00	0.1	1.1	
		and	124.00	132.00	8.00	0.22	2.3	0.03	0.1	0.29	2
		GNDD604	163.45	166.70	3.25	2.0	15.7	1.3	2.5	3.6	
		and	236.00	260.65	24.65	2.3	6.4	0.04	1.0	2.8	2
		inc	236.00	238.00	2.00	1.0	10.8	0.05	0.52	1.4	
		inc	247.50	249.00	1.50	5.0	3.4	0.18	0.26	5.2	
		inc	259.45	260.65	1.20	36.2	92.1	0.09	17.3	45.3	1
		and	375.00	375.90	0.90	24.9	15.3	0.01	3.5	26.7	1
		and	417.60	419.85	2.25	3.3	30.1	0.01	8.2	7.5	
		and	426.40	428.20	1.80	1.4	0.14	0.00	0.01	1.4	
		GNDD605	15.00	19.00	4.00	0.12	1.8	0.16	0.32	0.32	2
		and	46.00	70.00	24.00	0.13	2.6	0.01	0.54	0.41	2
		GNDD606	42.00	62.00	20.00	0.40	1.6	0.01	0.16	0.50	2
		inc	48.00	50.00	2.00	2.3	2.6	0.01	0.17	2.4	
		GNDD607	215.6	226.0	10.4	0.3	3.4	0.2	0.1	0.4	2
		inc	215.6	216.3	0.7	0.6	15.3	2.2	0.8	2.0	
		and	348.5	350.0	1.5	2.6	21.1	0.8	0.1	3.2	2
		inc	348.5	349.4	0.9	4.2	33.5	1.3	0.2	5.2	
		and	368.0	370.0	2.0	0.1	97.8	0.1	0.0	1.3	
		GNDD609	76.0	90.0	14.0	0.2	0.6	0.0	0.0	0.3	2
		and	123.5	127.0	3.5	0.3	0.4	0.1	0.0	0.3	2
		and	151.0	171.0	20.0	0.4	0.2	0.0	0.0	0.4	2
		inc	165.0	171.0	6.0	0.9	0.4	0.1	0.0	1.0	
		and	359.2	359.7	0.5	0.8	13.2	1.2	0.0	1.5	
		GNDD610	93.00	99.00	6.00	0.19	4.8	0.01	0.05	0.27	2
		GNDD611	68.0	72.0	4.0	0.3	1.7	0.3	0.1	0.5	2
		and	213.3	215.4	2.0	0.5	6.9	0.4	0.1	0.7	2
		inc	214.2	215.4	1.2	0.8	2.6	0.4	0.1	1.0	
		GNDD612	64.90	100.15	35.25	0.93	2.7	0.30	0.49	1.2	2
		inc	76.00	84.00	8.00	3.4	8.4	0.91	1.7	4.5	
		inc	99.00	100.15	1.15	0.76	3.8	0.70	0.82	1.3	
		and	117.00	131.00	14.00	1.0	1.1	0.01	0.22	1.1	2
		inc	117.00	121.00	4.00	3.0	3.1	0.02	0.61	3.3	
		and	148.00	162.50	14.50	0.93	6.3	0.04	0.10	1.1	2
		inc	154.00	158.00	4.00	2.4	4.7	0.05	0.11	2.5	
		and	176.20	180.00	3.80	0.28	0.52	0.00	0.03	0.30	2
		GNDD613	122.0	153.3	31.3	0.5	1.5	0.2	0.1	0.6	2
		inc	144.0	153.3	9.3	1.3	3.3	0.6	0.2	1.6	

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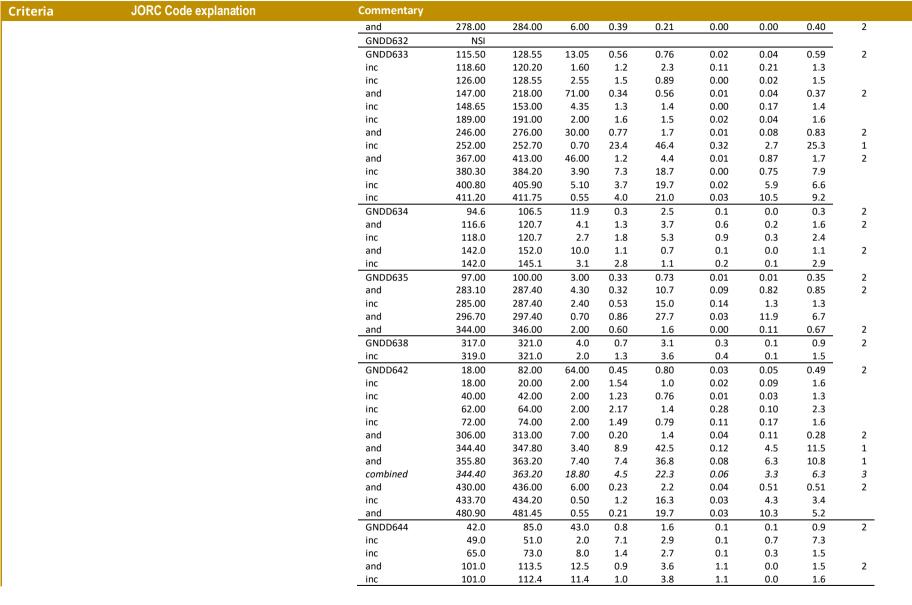
Criteria	JORC Code explanation	Commentary									
		GNDD615	142.0	146.0	4.0	0.2	1.9	0.0	0.0	0.2	2
		and	176.0	189.2	13.2	0.2	5.1	0.0	0.0	0.2	2
		and	231.0	272.4	41.4	0.2	0.6	0.0	0.0	0.2	2
		and	288.0	290.9	2.9	0.2	1.5	0.5	0.0	0.5	2
		and	321.0	344.0	23.0	0.4	3.3	0.1	0.0	0.5	2
		inc	321.0	323.0	2.0	1.4	16.7	0.0	0.0	1.6	
		inc	339.0	340.1	1.1	2.3	4.5	0.7	0.4	2.8	
		_ and	360.0	376.0	16.0	0.2	1.9	0.1	0.0	0.3	2
		GNDD617	49.0	53.0	4.0	0.8	2.1	0.2	0.1	0.9	2
		inc	51.0	53.0	2.0	1.1	2.7	0.2	0.2	1.3	
		and	65.0	72.0	7.0	0.6	0.4	0.0	0.0	0.6	2
		inc	67.0	69.0	2.0	1.0	0.4	0.0	0.0	1.1	
		and	82.0	89.1	7.1	0.3	4.3	0.1	0.0	0.4	2
		GNDD618	90.50	99.00	8.50	1.1	4.9	0.13	0.51	1.4	2
		inc	90.50	96.00	5.50	1.5	6.4	0.16	0.68	1.9	
		and	147.00	191.00	44.00	0.28	0.66	0.00	0.03	0.31	2
		inc	169.00	171.00	2.00	1.0	1.2	0.00	0.04	1.0	
		and	206.50	260.00	53.50	0.20	1.2	0.03	0.10	0.27	2
		inc	211.00	211.50	0.50	1.6	13.0	0.00	4.3	3.8	
		GNDD619	149.5	157.0	7.5	0.4	0.8	0.1	0.1	0.4	2
		inc	149.5	151.0	1.5	1.0	0.5	0.1	0.1	1.0	
		GNDD621	94.0	97.0	3.0	0.3	6.9	0.0	0.0	0.4	2
		and	205.0	207.0	2.0	8.0	0.3	0.1	0.0	8.0	2
		and	296.4	299.5	3.1	0.2	2.5	0.2	0.1	0.3	2
		GNDD623	157.0	161.0	4.0	0.2	2.3	0.2	0.1	0.4	2
		and	195.0	213.0	18.0	0.3	0.7	0.2	0.1	0.4	2
		inc	206.0	208.0	2.0	0.5	2.6	0.9	0.5	1.0	
		inc	212.0	213.0	1.0	1.5	1.9	0.6	0.3	1.9	
		GNDD624	79.0	143.0	64.0	0.4	4.2	0.2	0.0	0.5	2
		inc	81.0	91.0	10.0	1.2	9.7	0.2	0.1	1.4	
		inc	107.0	111.3	4.3	0.9	17.2	1.4	0.3	1.8	
		GNDD626	6.00	40.00	34.00	0.20	6.0	0.02	0.20	0.37	2
		inc	14.00	17.10	3.10	0.73	30.7	0.09	0.40	1.3	
		and	135.70	168.00	32.30	0.65	1.4	0.06	0.18	0.76	2
		inc	155.00	157.00	2.00	0.95	1.5	0.10	0.11	1.0	
		inc	166.45	168.00	1.55	8.3	13.2	0.81	2.6	9.8	
		and	202.00	209.20	7.20	0.45	1.1	0.03	0.14	0.54	2
		inc	205.00	206.80	1.80	1.1	2.6	0.09	0.48	1.4	
		GNDD629	117.00	215.00	98.00	0.36	1.6	0.01	0.10	0.43	2
		inc	129.00	131.00	2.00	1.9	2.3	0.00	0.13	2.0	
		inc	164.00	166.00	2.00	1.0	0.62	0.00	0.05	1.0	
		inc	191.75	194.65	2.90	3.1	19.1	0.26	1.4	4.1	
		and	262.55	266.50	3.95	0.26	2.0	0.03	0.14	0.36	2

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		GNDD648	2.0	30.0	28.0	0.9	1.9	0.1	0.0	1.0	2
		inc	10.0	14.0	4.0	2.6	3.0	0.1	0.0	2.7	
		inc	26.0	30.0	4.0	2.7	2.3	0.0	0.0	2.8	
		GNDD655	19.9	29.7	9.8	0.4	4.6	0.9	0.2	0.9	2
		inc	21.7	26.5	4.9	0.6	6.2	1.0	0.4	1.3	
		and	46.0	54.6	8.6	0.3	0.6	0.0	0.0	0.3	2
		and	69.5	72.0	2.5	0.4	0.8	0.1	0.0	0.5	2
		and	96.9	104.7	7.8	0.9	4.3	0.6	0.1	1.2	2
		inc	96.9	101.8	4.9	1.2	6.4	8.0	0.1	1.7	
		GNDD656	70.0	86.0	16.0	0.1	0.9	0.1	0.0	0.2	2
		GNDD661	17.0	111.0	94.0	0.6	1.2	0.1	0.0	0.7	2
		inc	17.0	24.5	7.5	2.2	2.2	0.2	0.1	2.3	
		inc	44.0	46.0	2.0	1.0	0.3	0.1	0.0	1.0	
		inc	64.0	66.0	2.0	2.8	0.2	0.0	0.0	2.8	
		inc	74.0	76.0	2.0	8.8	0.2	0.1	0.0	8.8	
		inc	109.0	111.0	2.0	1.0	3.5	0.2	0.1	1.2	
		and	135.0	139.0	4.0	2.9	1.3	0.2	0.0	3.0	2
		and	163.0	208.0	45.0	0.5	6.3	0.2	0.0	0.6	2
		inc	183.0	185.0	2.0	1.1	4.5	0.0	0.0	1.2	
		inc	191.0	192.4	1.4	5.4	50.1	0.7	0.3	6.4	
		inc	197.6	202.0	4.4	2.0	10.5	1.0	0.1	2.6	
		GNDD664	37.5	38.3	0.8	2.8	47.2	0.3	0.1	3.5	
		and	57.8	62.0	4.2	0.1	4.2	0.8	0.1	0.5	2
		inc	57.8	58.2	0.4	0.1	4.4	3.2	0.0	1.7	
		GNDD684	115.0	142.0	27.0	0.3	0.7	0.0	0.0	0.3	2
		inc	125.0	127.0	2.0	1.1	6.8	0.3	0.2	1.4	
		and	209.0	274.3	65.3	2.3	1.7	0.2	0.1	2.4	2
		inc	256.0	257.3	1.3	1.1	0.8	0.0	0.0	1.1	
		inc	265.6	267.5	1.9	71.6	41.1	3.8	3.5	74.5	10
		and	324.9	337.0	12.2	10.1	11.7	1.5	0.1	11.0	2
		inc	324.9	333.1	8.3	14.8	17.1	2.2	0.1	16.0	10
		and	354.5	362.0	7.6	1.1	0.7	0.1	0.0	1.2	2
		inc	359.8	362.0	2.2	3.2	0.9	0.0	0.0	3.2	
		and	408.0	418.0	10.0	0.6	0.7	0.0	0.0	0.6	2
		inc	408.0	412.0	4.0	1.4	1.8	0.0	0.0	1.4	
		GNDD685	3.0	57.0	54.0	0.2	2.8	0.1	0.1	0.3	2
		and	91.0	104.6	13.6	0.1	8.7	0.1	0.0	0.3	2
		and	362.4	370.8	8.4	4.5	5.3	2.0	0.0	5.5	2
		inc	362.4	365.0	2.6	14.3	16.1	6.3	0.0	17.4	
		inc	362.4	363.7	1.3	26.5	29.5	12.3	0.0	32.5	10
		and	409.0	429.0	20.0	1.0	0.9	0.2	0.0	1.1	2
		inc	413.0	418.5	5.5	2.4	1.1	0.4	0.0	2.6	
		inc	425.0	427.0	2.0	1.5	1.4	0.1	0.0	1.6	

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Level 1 1205 Hay Street West Perth WA 6005 Directors

Mr Kris Knauer, MD and CEO Mr Scott Funston, Finance Director Mr Fletcher Quinn, Chairman Mr Sergio Rotondo, Exec. Director Contact

T: +61 8 6380 9235 E: admin@challengerex.com

Criteria	JORC Code explanation	Commentary									
		and	548.0	549.7	1.7	0.4	8.1	2.1	0.0	1.5	2
		and	624.6	627.0	2.4	0.2	0.6	1.2	0.0	0.8	2
		and	651.2	653.0	1.9	20.0	8.3	6.8	0.0	23.3	2
		GNDD697 inc	41.0 52.5	106.3 57.0	65.3 4.5	0.5 2.4	2.0 6.6	0.2 0.8	0.4 1.4	0.7 3.3	2
		inc	52.5 65.0	68.6	4.5 3.6	2.4	14.0	2.0	3.3	5.0	
			10 g/t Au equiv		3.0	2.3	11.0	2.0	3.3	3.0	
		(2) cut off 0.2	2 g/t Au equiva	ent							
		(3) combined	zones with 0.2	g/t Au cut o	ff (grades	include ir	nternal diluti	on from bet	ween zone:	s)	
		• •	I zones with 1.0	_	off (grades	include ii	nternal dilati	on from bet	ween zone	s)	
			icant intersecti								
Data aggregation	 In reporting Exploration Results weight 		average signi								
methods	averaging techniques maximum and/o		o cut-off grac								
	minimum grade truncations (eg cutting		lution betwee								
	high grades) and cut-off grades are usu		lution betwee								
	Material and should be stated.	been used	to report gol	d grade equ	uivalent (AuEq): A	u US\$ 1780) / oz Ag U	S\$24 /oz a	nd Zn US	5 2800 /t.
	 Where aggregate intercepts incorporate 	te No top cut	s have been a	pplied to t	he repor	ted grade	es.				
	short lengths of high-grade results and					B					
	longer lengths of low-grade results the										
	procedure used for such aggregation sl	hould									
	be stated and some typical examples o	f such									
	aggregations should be shown in detai	l.									
	- The assumptions used for any reporting										
	metal equivalent values should be clear	-									
	stated.	,									
Relationship	- These relationships are particularly	The mineralisati	on is moderat	elv or stee	olv dippii	ng and st	rikes NNE	and ENE. F	or some d	Irill holes.	there is
between	important in the reporting of	insufficient info				-					
	Exploration Results.	the exploration		•							Ö
mineralisation	- If the geometry of the mineralisation	•								., .	
widths and	with respect to the drill hole angle is	Apparent width						eralisation	changes a	and/or be	dding-parall
intercept lengths	known its nature should be reported.	mineralisation in	itersects NW	or ENE-strii	king cros	s faults a	na veins.				
	- If it is not known and only the down	Representative	cross section	nterpretati	ons have	been pr	ovided per	iodically w	ith release	es of signif	icant
	hole lengths are reported there should	intersections to								_	
	be a clear statement to this effect (eg							•			
	'down hole length true width not										
	known').										
Diagrams	- Appropriate maps and sections (with	Representative	mans and sec	ions are nr	ovided in	the boo	ly of report	s released	to the AS	Κ	
Diagianis	Appropriate maps and sections (with	Representative	naps and sec	ions are pr	o viucu ii	i tile bot	iy or report	J ICICUSCU	to the Asi	٠.	

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Criteria	JORC Code explanation	Commentary
	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.	
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 and plans of all drilling with results.
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.	Geological context and observations about the controls on mineralisation where these have been made are provided in the body of the report. 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. Two 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.
Further work	 The nature and scale of planned further work (eg tests for lateral extensions or depth extensions or large-scale stepout drilling). Diagrams clearly highlighting the areas of possible extensions including the main geological interpretations and future drilling areas provided this information is not commercially sensitive. 	 CEL Plans to undertake the following over the next 12 months Additional resource extension, infill and exploration drilling; Detailed interpretation of known mineralized zones; 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 and comminution test work. A preliminary economic assessment / scoping study for a mining project.

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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.
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 drill hole data is backed up and is updated periodically by the CEL GIS and data management team. The Competent Person has undertaken site visits during exploration. Site visits were undertaken from 3 to 16 October 2019 15 to 30 November 2019 and 1-19 February 2020 before COVID-19 closed international travel. Post COVID site visits were undertaken from 21 November – 4 December 2021 and 11 – 23 May 2022. 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 	30 separate domains were interpreted over a strike length of 2.2kms. The domains vary in width and orientation from 2m up to 100m in width. The deepest interpreted domain extends from the surface down approximately 550m below the surface.

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Criteria	JORC Code explanation	Commentary							
	Mineral 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 parameters and maximum distance of extrapolation from data points. If a computer assisted estimation method was chosen include a description of computer software and parameters used. The availability of check estimates previous estimates and/or mine production records and whether the Mineral Resource estimate takes appropriate account of such data. The assumptions made regarding recovery of by-products.	Estimation was made for Au Ag, Zn and Pb being the elements of economic interest. Estimate was also made for Fe and S being the elements that for pyrite which is of economic and metallurgical interest and is also used to estimate the density for bocks in the Mineral Resource Estimate. No previous JORC Resource estimates or non-JORC Foreign Resource estimates were made with similar methods to compare to the current Resource estimate. No production records are available to provide comparisons. A 2m composite length was selected after reviewing the original sample lengths from the drilling which showed an average length of 1.54m for samples taken within the mineralised domains. A statistical analysis was undertaken on the sample composites Top cuts were applied to the Au, Ag, Zn and Pb composites on a domain-by-domain basis. The domains were then grouped by host rock and mineralisation style and group domain top cuts were applied in order to reduce the influence of extreme values on the resource estimates without downgrading the high-grade composites too severely. The top-cut values were chosen by assessing the high-end distribution of the grade population within each group and selecting the value above which the distribution became erratic. The following table shows the top cuts applied to each group and domain for Au, Ag, Zn and Pb.							
	 Estimation of deleterious elements or other non-grade variables of economic 		Group	Domain	Au (nnm)	Ag (nnm)	Zn (%)	Pb (%)	1
	significance (eg sulphur for acid mine drainage characterisation). - In the case of block model interpolation the block size in relation to the average sample spacing and the search employed.		Group Fault Zone hosted (Magnata and Sanchez)	Domain 101 102 103 104 201	Au (ppm) 80	Ag (ppm) 300	20 (%)	2.5	
	Any assumptions behind modelling of selective mining units.Any assumptions about correlation		LUT (siltstone) hosted	111 114 212	14	70	4.5	0.8	
	 between variables. Description of how the geological interpretation was used to control the resource estimates. Discussion of basis for using or not using grade cutting or capping. The process of validation the checking process used the comparison of model data to drill hole data and use of reconciliation 	1	DAC (intrusive) hosted	112 113 115 131 132 133 134 203 213	9	65	7	1.2	

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data	if available		301 302 303 304				
			303				
			304				
			305				
			202				
			121				
			211				
		CAL (limestone) hosted	221	80	300	20	2.5
			222				
			223				
			224				

Block modelling was undertaken in Surpac™ V6.6 software.

A block model was set up with a parent cell size of 10m (E) $\times 20m$ (N) $\times 10m$ (RL) with standard sub-celling to 2.5m (E) $\times 5.0m$ (N) $\times 2.5m$ (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 wireframes.

Variography was carried out using Leapfrog Edge software on the two metre composited data from each of the 28 domains for each variable.

All relevant variables; Au, Ag, 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, Zn, Fe and S that were not estimated were assigned the average values on a per-domain basis.

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Criteria	JORC Code explanation	Commentary
		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 0.25 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 0.25 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
Mining factors or assumptions	- Assumptions made regarding possible mining methods minimum mining dimensions and internal (or if applicable external) mining dilution. It is always necessary as part of the process of determining reasonable prospects for eventual economic extraction to consider potential mining methods but the assumptions made regarding mining methods and parameters when estimating Mineral Resources may not always be rigorous. Where this is the case this should	considered to have reasonable prospects of future mining by underground methods. The Resource estimate has assumed that near surface mineralisation would be amenable to open pit mining given that the mineralisation is exposed at surface and under relatively thin unconsolidated cover. A surface mine 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 following parameters with prices in USD: - 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 per tonne - Average metallurgical recoveries of 94.9% for Au, 90.9 % for Ag and 67 % for Zn and 57.8 % for Pb Ore and waste mining cost of \$2.00 per tonne - Unconsolidated cover removal cost of \$0.10 per tonne - Processing cost of \$10.00 per tonne - Transport and marketing of \$50 / oz of AuEq (road to Jan Juan then rail to Rosario Port) - Royalty of \$60 per oz Au, 3% for Ag, Zn and Pb.

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Criteria	JORC Code explanation	Commentary
	be reported with an explanation of the basis of the mining assumptions made.	 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 0.25 g/t AuEq 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.
Metallurgical factors or assumptions	- The basis for assumptions or predictions regarding metallurgical amenability. It is always necessary as part of the process of determining reasonable prospects for eventual economic extraction to consider potential metallurgical methods but the assumptions regarding metallurgical treatment processes and parameters made when reporting Mineral Resources may not always be rigorous. Where this is the case this should be reported with an explanation of the basis of the metallurgical assumptions made.	CEL has completed Stage 1 metallurgical test work on representative composite sample of mineralisation from: 1. Two separate composite samples of limestone-hosted massive sulphide (manto) Sample A has a weighted average grade of 10.4 g/t Ag, 4.0

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West Perth WA 6005

Directors
Mr Kris Knauer, MD and CEO
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Mr Sergio Rotondo, Exec. Director

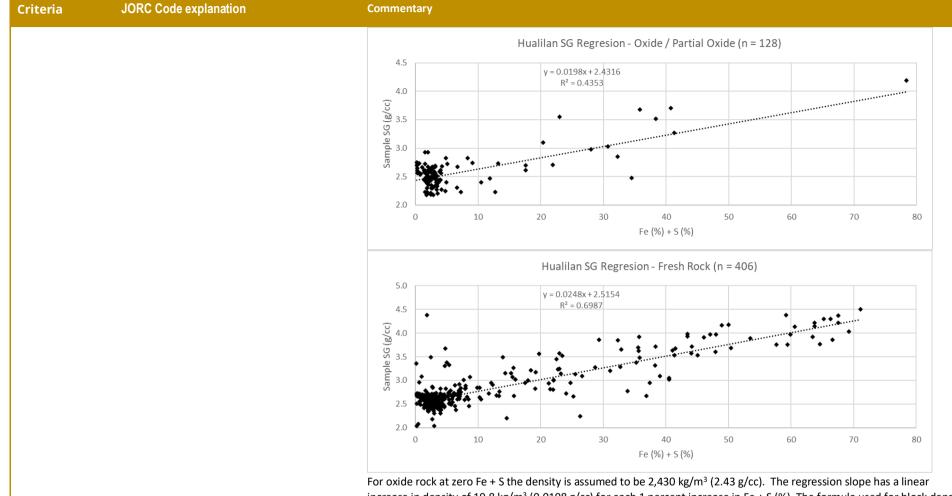
Criteria	JORC Code explanation	Commentary
		3. single stage rougher sulphide flotation,
		4. P80 = 20-30 micron regrind of the rougher concentrate (5-10% mass),
		5. one or two re-cleaning stages of the Au-Ag Rougher concentrate
		At primary grind of p80 = 76 micron and regrind of p80 = 51 micron an AuAg concentrate can be produced grading 5
		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 23
		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;
		 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
		Additional Stage 2 work involving comminution and variability testing, blended test work, and pilot plant testing is ongoing and planned.
Environmental	- Assumptions made regarding possible	It is considered that there are no significant environmental factors which would prevent the eventual extraction of gold
factors or	waste and process residue disposal options.	from the project. Environmental surveys and assessments have been completed in the past and will form a part of future
assumptions	It is always necessary as part of the process	pre-feasibility studies.
assamperons	of determining reasonable prospects for	

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Mr Kris Knauer, MD and CEO
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Mr Sergio Rotondo, Exec. Director

Contact T: +61 8 6380 9235 E: admin@challengerex.com

Criteria	JORC Code explanation	Commentary
Bulk density	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 environmental impacts should be reported. Where these aspects have not been considered this should be reported with an explanation of the environmental assumptions made. - Whether assumed or determined. If assumed the basis for the assumptions. If determined the method used whether wet or dry the frequency of the measurements the nature size and representativeness of the samples. - The bulk density for bulk material must have been measured by methods that adequately account for void spaces (vugs porosity etc) moisture and differences between rock and alteration zones within the deposit. - Discuss assumptions for bulk density estimates used in the evaluation process of	CEL has collected specific gravity measurements from drill core, which have been used to estimate block densities for the Resource estimate. Within the mineralised domains there are 534 specific gravity measurements made on drill core samples of 0.1 – 0.2 metres length. Measurements we determined on a dry basis by measuring the difference in sample weight in water and weight in air. For porous samples, the weight in water was measured after wrapping the sample so that no water enters the void space during weighing. A regression model for block density determination in oxide / partial oxide / fracture oxide (oxide) rock and a separate regression model for fresh rock samples has been made by plotting assay interval Fe (%) + S (%) from the interval where the SG measurement was made against the SG measurement. Fe and S are the two elements that form pyrite which is the mineral that is commonly associated with gold and base metal mineralisation at Hualilan. SG plotted against (Fe+S) follows a linear trend within the mineralised domains for oxide and fresh rock as shown in the graphs below.
	the different materials.	

Issued Capital 1,044.9m shares 10m options 120m perf shares 16m perf rights Australian Registered Office Level 1 1205 Hay Street West Perth WA 6005 Directors
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Mr Sergio Rotondo, Exec. Director



For oxide rock at zero Fe + S the density is assumed to be 2,430 kg/m 3 (2.43 g/cc). The regression slope has a linear increase in density of 19.8 kg/m 3 (0.0198 g/cc) for each 1 percent increase in Fe + S (%). The formula used for block density (kg/m 3) determination in oxide rock is 2,430 + 19.8 x (Fe % + S%).

For fresh rock at zero Fe + S the density is assumed to be 2,520 kg/m³ (2.52 g/cc). The regression slope has a linear increase in density of 24.8 kg/m³ (0.0248 g/cc) for each 1 percent increase in Fe + S (%). The formula used for block density (kg/m³) determination in oxide rock is 2,520 + 24.8 x (Fe % + S%).

Challenger Exploration Limited ACN 123 591 382 ASX: CEL Issued Capital 1,044.9m shares 10m options 120m perf shares 16m perf rights Australian Registered Office Level 1 1205 Hay Street West Perth WA 6005 Directors
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Contact T: +61 8 6380 9235 E: admin@challengerex.com

Criteria	JORC Code explanation	Commentary
Classification	 The basis for the classification of the Mineral Resources into varying confidence categories. Whether appropriate account has been taken of all relevant factors (ie relative confidence in tonnage/grade estimations reliability of input data confidence in continuity of geology and metal values quality quantity and distribution of the data). Whether the result appropriately reflects the Competent Person's view of the deposit. 	The Mineral Resource has been classified based on the guidelines specified in the JORC Code. The classification level is based upon semi-qualitative assessment of the geological understanding of the deposit, geological and mineralisation continuity, drill hole spacing, QC results, search and interpolation parameters and an analysis of available density information. The estimation search strategy was undertaken in three separate passes with different search distances, and the minimum number of samples used to estimate a block which were then used as a guide for the classification of the resource into Indicated, Inferred and Unclassified. The classification was then further modified to restrict the Indicated Resource to the domains with closer spaced drilling. The potential open pit resource was constrained within an optimised pit shell run using a gold price of \$1,800 per ounce. Resources reported inside the pit shell were reported above a AuEq cut-off grade of 0.25 ppm and Resources outside the pit shell were reported above a AuEq cut-off grade of 1.0 ppm. Resource reported outside the pit shell above a 1.0 g/t AuEq cut-off is considered 100% Inferred. The Competent Person has reviewed the result and determined that these classifications are appropriate given the confidence in the data and results from drilling.
Audits or reviews	 The results of any audits or reviews of Mineral Resource estimates. 	The Mineral Resource estimate has not been independently audited or reviewed.
Discussion of relative accuracy/confidence	 Where appropriate a statement of the relative accuracy and confidence level in the Mineral Resource estimate using an approach or procedure deemed appropriate by the Competent Person. For example the application of statistical or geostatistical procedures to quantify the relative accuracy of the resource within stated confidence limits or if such an approach is not deemed appropriate a qualitative discussion of the factors that could affect the relative accuracy and confidence of the estimate. The statement should specify whether it relates 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. 	There is sufficient confidence in the data quality drilling methods and analytical results that they can be relied upon. The available geology and assay data correlate well. The approach and procedure is deemed appropriate given the confidence limits. The main factors which could affect relative accuracy are: - domain boundary assumptions - orientation - grade continuity - top cut. Grade continuity is variable in nature in this style of deposit and has not been demonstrated to date and closer spaced drilling is required to improve the understanding of the grade continuity in both strike and dip directions. It is noted that the results from the twinning of three holes by La Mancha are encouraging in terms of grade repeatability. The deposit contains very high grades and there is need for the use of top cuts. No production data is available for comparison.

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Criteria	JORC Code explanation	Commentary
	 These statements of relative accuracy and confidence of the estimate should be compared with production data where 	
	available.	

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