

# North Pilbara Exploration and Resource Development Success

## HIGHLIGHTS

- 1. The first drilling program at the favourably located Miralga Creek Project has returned excellent preliminary intercepts, including:
  - 64m @ 59.85%Fe (from 36m) in MRRC0012,
  - $\circ~$  48m @ 57.97%Fe (from surface) in MRRC0002, and
  - o 30m @ 59.20%Fe (from 2m) in MRRC0001.
- 2. New assays from Corunna Downs support existing interpretations, with further intercepts including:
  - 70m @ 57.78% Fe (from surface) in CDRC0238,
  - o 94m @ 58.92% Fe (from surface) in CDRC0239,
  - o 58m @ 60.52% Fe (from 112m) in CDRC0225, and
  - 58m @ 57.64% Fe (from 56m) in CDRC0197.

3. Indicated and Inferred Mineral Resource of 25.4Mt @ 57.1% Fe declared for the Corunna Downs Split Rock deposit, including 20Mt Indicated and 5.4Mt Inferred.

Atlas Iron Ltd is pleased to announce further exploration and resource development success from its North Pilbara projects.

Excellent first pass RC drilling results from Miralga Creek and a continued stream of intercepts at Corunna Downs demonstrate that the North Pilbara remains a favorable location for iron exploration. Atlas' significant land holding in this region is a key strategic advantage, supported by its proven on-highway haulage model. Continued exploration success is building a broad pipeline of opportunities for Atlas to maintain and grow its Reserve base.

Atlas Iron's Managing Director Ken Brinsden said "Miralga Creek has the potential to be a fantastic value-add to our existing Abydos mine."

"The location and quality of the Split Rock resource demonstrates the strong potential for Corunna Downs to become a substantial project in Atlas' portfolio that will likely complement our other North Pilbara operations. There are many more targets yet to be tested in the Corunna Downs Project," he added.

### **Miralga Creek Exploration**

First pass RC drilling at the 100% owned Miralga Creek Project (note 1% vendor royalty applies), has intersected iron mineralisation in banded iron formations along strike from the Abydos operation. The Miralga Creek project is located in close proximity to the Abydos Haul Rd (Figure 1, Figure 2). Follow up drilling is planned at Miralga Creek to further define the extent of mineralisation and progress the project given its favourable location.

Figures 3, 4 and 5 show details of the initial drill program undertaken. Significant intercepts are highlighted in Attachment 1.

### Corunna Downs Exploration

The Corunna Downs Project 160km south east of Port Hedland continues to impress with further significant intercepts returned recently from the Runway and Shark Gully prospects (see previous Atlas ASX release of 9 December 2013). These results support the down dip continuity of mineralisation previously identified and give further support to the previously released Exploration target of 100-150Mt @ 55-58% Fe for the northern portion of the Corunna Downs Project (see ASX release of 9 December, 2013). Estimation works are underway on these Prospect areas.

Figures 6, 7 and 8 show details of these new drill results and significant intercepts are tabulated in Attachment 2.

### Corunna Downs Resource Development

An updated Indicated and Inferred Mineral Resource estimate for the Split Rock deposit has been developed. Please refer to Table 1 below for details.

A detailed explanation of the mineral resource estimation and Competent Person attribution is provided in Attachment 3.

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$\square$	Resource	Tonnes	Fe	SiO <sub>2</sub>	$Al_2O_3$	Р	S	LOI	MnO	CaO	MgO	TiO <sub>2</sub>	K <sub>2</sub> O	C=F=9/
	Classification	(Mt)	(%)	(%)	(%)	(%)	(%)	(%)	(%)	(%)	(%)	(%)	(%)	CaFe%
	Indicated	20.0	57.3	6.5	1.3	0.12	0.01	8.9	0.43	0.03	0.04	0.03	0.03	62.9
	Inferred	5.4	56.2	7.1	2.1	0.12	0.01	9.1	0.31	0.15	0.12	0.06	0.03	61.9
	Total	25.4	57.1	6.6	1.5	0.12	0.01	9.0	0.40	0.06	0.06	0.04	0.03	62.7

#### Table 1: Split Rock Mineral Resource as at December 2013, reported above a 50% Fe cut-off

\*CaFe% is calcined Fe calculated by Atlas using the following formula (Fe%/(100-LOI%))\*100 \*Small discrepancies may occur due to rounding

#### Investor Enquiries:

Atlas Iron +61 8 6228 8000 Ken Brinsden, Managing Director

#### Media Enquiries:

Read Corporate Paul Armstrong +61 8 9388 1474 +61 421 619 084

#### **Competent Person's Statement – Exploration Results**

The information in this report that relates to Exploration Results is based on information compiled by Pip Darvall, a Competent Person who is a Member of the Australasian Institute of Mining and Metallurgy and the Australian Institute of Geoscientists. Pip Darvall is a full time employee of Atlas Iron Ltd and has sufficient experience that is relevant to the style of mineralisation and type of deposit under consideration and to the activity being undertaken to qualify as a Competent Person as defined in the 2012 Edition of the 'Australasian Code for Reporting of Exploration Results, Mineral Resources and Ore Reserves'. Pip Darvall consents to the inclusion in this report of the matters based on his information in the form and context in which it appears. The Exploration Results have been verified by Steven Warner, a Competent Person who is a Member of the Australasian Institute of Mining and Metallurgy. Steven Warner is a full time employee of Atlas Iron Ltd and has sufficient experience that is relevant to the style of the matters based on his information in the form and context in which it appears. The Exploration Results have been verified by Steven Warner, a Competent Person who is a Member of the Australasian Institute of Mining and Metallurgy. Steven Warner is a full time employee of Atlas Iron Ltd and has sufficient experience that is relevant to the style of mineralisation and type of deposit under consideration and to the activity being undertaken to qualify as a Competent Person as defined in the 2012 Edition of the 'Australasian Code for Reporting of Exploration Results, Mineral Resources and Ore Reserves'. Steven Warner consents to the inclusion in this report of the matters based on his information in the form and context in which it appears.

#### Competent Person's Statement – Mineral Resources

The information in this report that relates to Mineral Resources is based on information compiled by Steven Warner who is a member of the Australasian Institute of Mining and Metallurgy. Steven Warner is a permanent employee of Atlas Iron Ltd and has sufficient experience that is relevant to the style of mineralisation and type of deposit under consideration and to the activity which he is undertaking to qualify as Competent Person as defined in the 2012 Edition of the Australasian Code for Reporting of Exploration Results, Mineral Resources and Ore Reserves. Steven Warner consents to the inclusion in the report of the matters based on his information in the form and context in which it appears.

#### Exploration and Resource Targets

Any discussion in relation to the potential quantity and grade of Exploration Targets is only conceptual in nature. While Atlas is confident that it will report additional JORC compliant resources, there has been insufficient exploration to define mineral resources in addition the current JORC compliant Mineral Resource inventory and it is uncertain if further exploration will result in the determination of additional JORC compliant Mineral Resources.

#### FIGURES

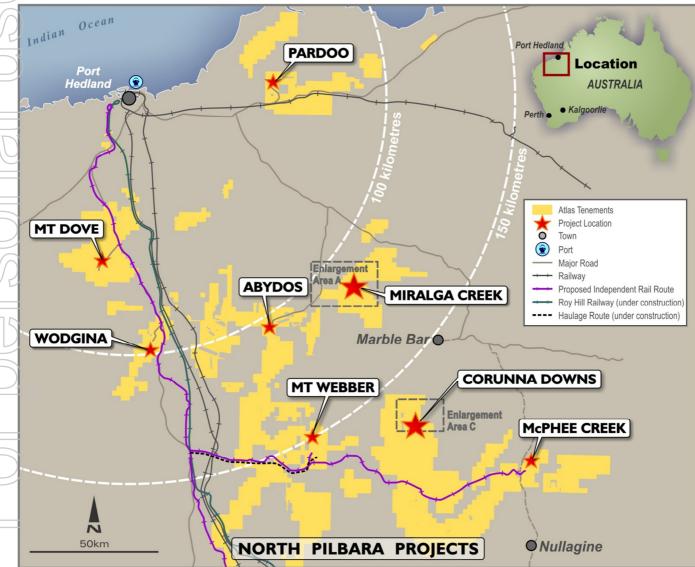
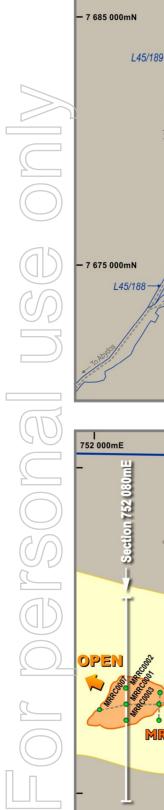


Figure 1 - North Pilbara Projects, Atlas Tenure, Existing and Proposed Infrastructure.



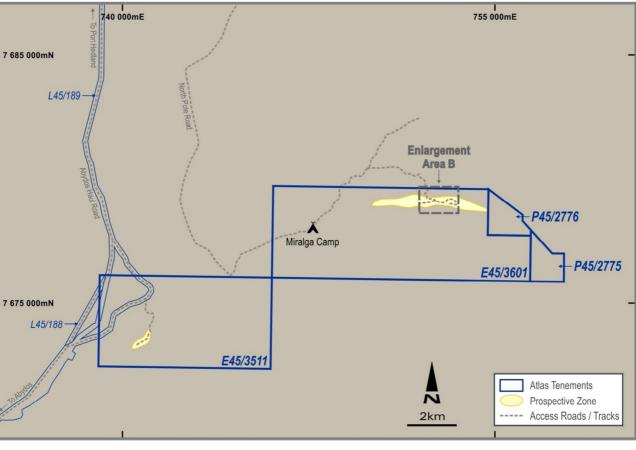


Figure 2 – Enlargement Area, Miralga Creek Prospect.

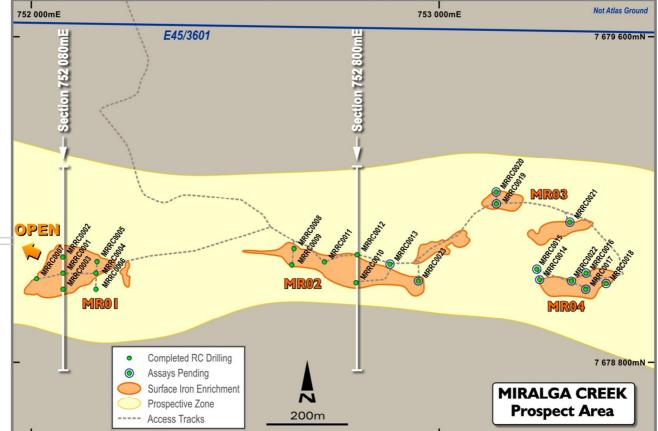
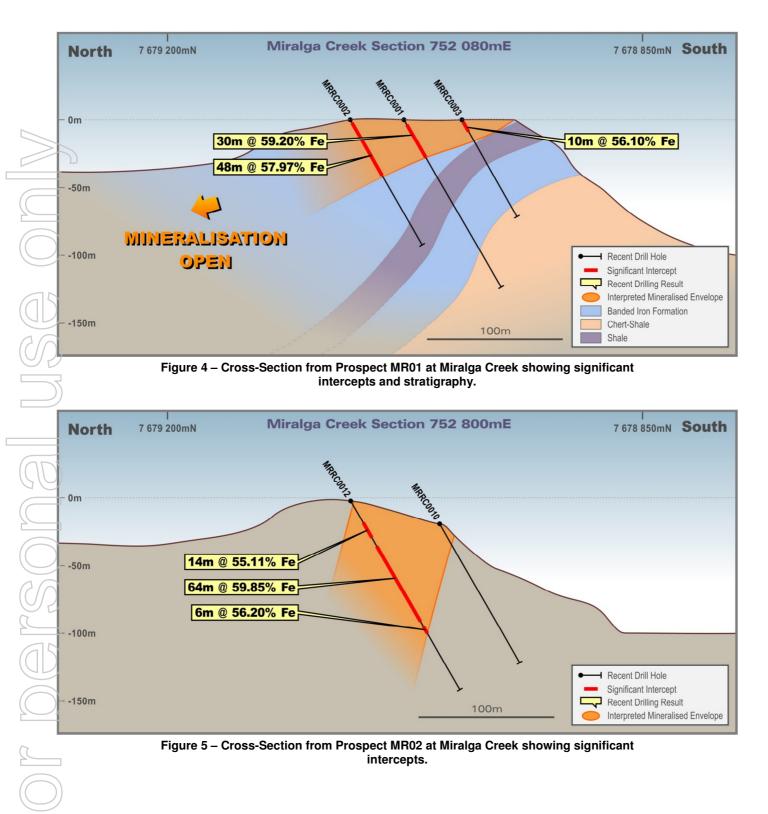


Figure 3 – Enlargement Area B, Collar Plan showing recently drilled RC holes at Miralga Creek and location of sections shown in Figure 4.



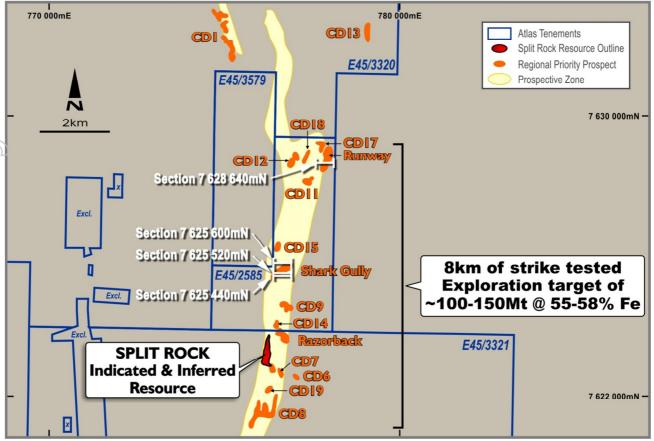


Figure 6 –Enlargement Area C - Plan showing Corunna Downs Prospects and location of sections shown in Figure 7 and Figure 8, and location of resource shown in Figure 9.

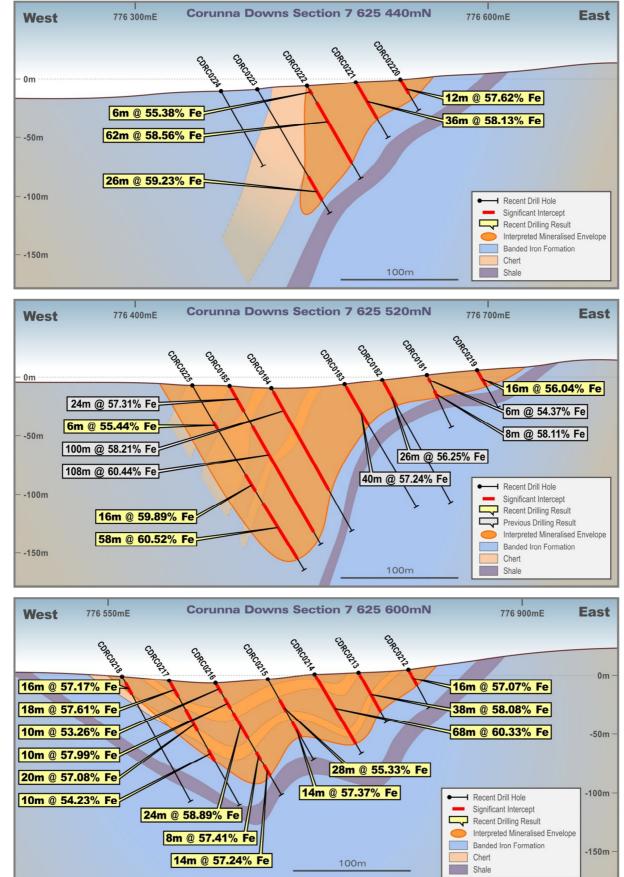


Figure 7 – Cross-Sections from the Corunna Downs Shark Gully Prospect showing significant intercepts and stratigraphy from recently received results. Note on section 7625520mN results shown in grey have previously been released.

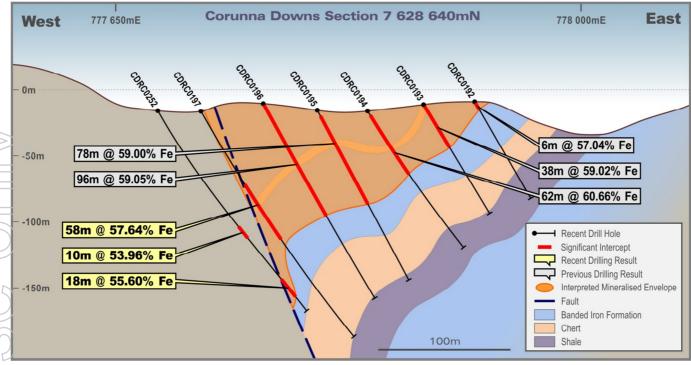


Figure 8 – Cross-Section from the Corunna Downs Runway Prospect showing significant intercepts and stratigraphy from recently received results. Note that results shown in grey have previously been released.

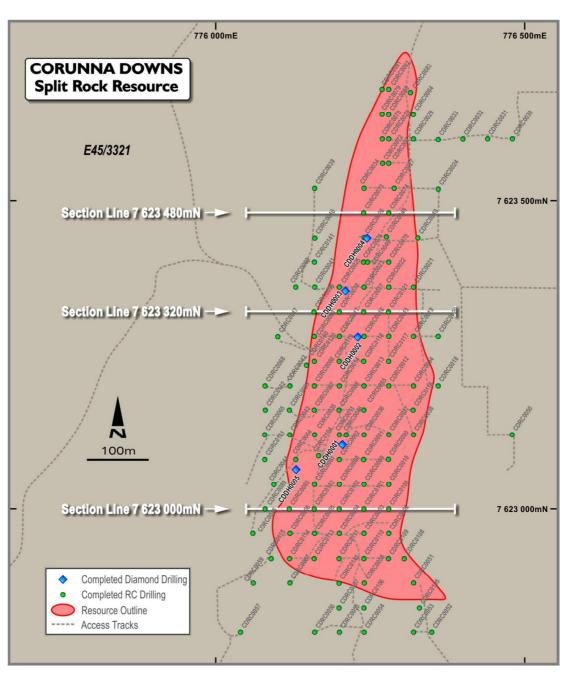


Figure 9 – Plan projection of the Split Rock Prospect at Corunna Downs. Showing updated Resource Outline, RC and Diamond Drill-Hole Collar Locations and the location of cross sections shown in Figure 10.

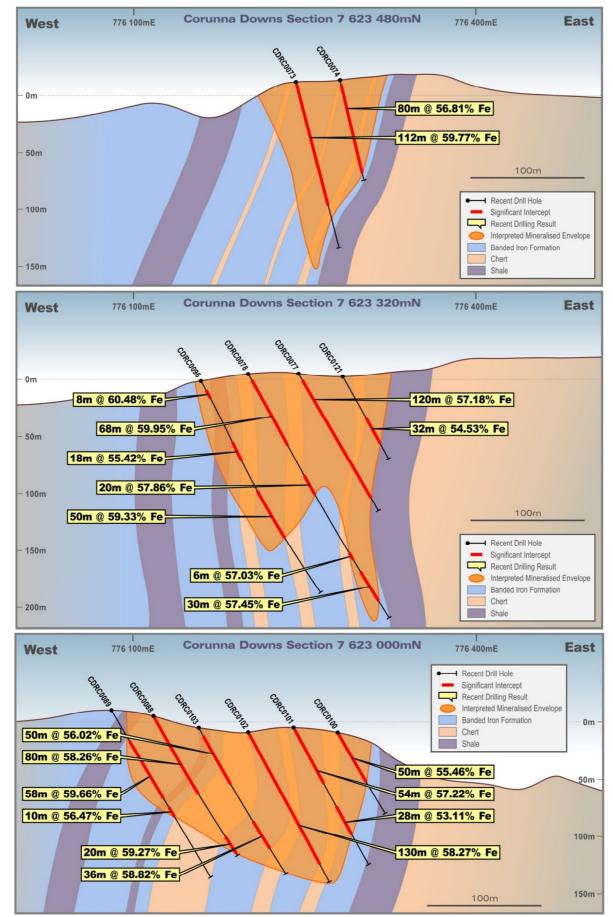


Figure 10 – Representative cross-sections at Split Rock showing new significant intercepts and stratigraphy.

## **ATTACHMENT 1**

## MIRALGA CREEK SIGNIFICANT INTERCEPTS AND JORC COMPLIANCE STATEMENTS

	HOLEID	EAST GDA_94_Z50	NORTH GDA_94_Z50	RL (m)	AZIMUTH	DIP	FROM (m)	TO (m)	WIDTH (m)	Fe%	SiO2%	AI2O3%	Р%	LOI1000%	S%
	MRRC0001	752080	7679020	267	180	-60	2	32	30	59.2	2.59	1.92	0.082	10.53	0.023
	MRRC0002	752080	7679060	276	180	-60	0	48	48	57.97	3.7	2.49	0.102	10.36	0.014
	MRRC0003	752079	7678978	300	180	-60	0	10	10	56.1	6.97	2.6	0.077	9.57	0.014
$\geq$	MRRC0004	752160	7679020	291	180	-60	0	8	8	55.78	8.45	1.56	0.107	9.74	0.022
	MRRC0007	752014	7679004	263	180	-60	8	24	16	55.6	8.46	1.83	0.194	9.6	0.013
1	MRRC0008	752645	7679080	261	180	-60	0	14	14	54.68	8.42	2.63	0.052	10.29	0.024
	MRRC0009	752640	7679040	269	180	-60	50	78	28	59.7	4.13	0.47	0.095	9.62	0.005
	MRRC0009	752640	7679040	269	180	-60	116	124	8	54.94	9.98	1.79	0.128	9.01	0.001
	MRRC0011	752720	7679047	264	180	-60	20	32	12	55.96	6.63	2.42	0.054	10.54	0.009
	MRRC0011	752720	7679047	264	180	-60	36	60	24	58.33	4.8	1.15	0.087	10.37	0.009
	MRRC0012	752800	7679065	290	180	-60	10	24	14	55.11	7.9	2.64	0.089	10.34	0.021
	MRRC0012	752800	7679065	290	180	-60	36	100	64	59.85	2.43	1.22	0.122	10.34	0.009
	MRRC0012	752800	7679065	290	180	-60	106	112	6	56.2	7.67	1.24	0.179	10.06	0.001

#### Significant Intercepts at Miralga Creek.

Notes to Miralga Creek Significant Intercepts: Assay results are based on 2 meter samples from cone split RC samples, analysis by XRF with total LOI by Thermo-Gravimetric Analysis. 10% of samples are subject to QAQC procedures (standards and duplicates). Laboratory check samples are routinely performed on each sample submission. Significant Intercepts are reported at a 53% Fe cut-off grade, and include a maximum of 6m internal dilution and 6m minimum width for intersection. Drill holes are spaced on a nominal 80m X 40m grid pattern, with collar locations surveyed by hand held GPS with an approximate error of +/- 3m horizontally (Northing and Easting) and +/-5m vertically (RL).

### JORC 2012 TABLE 1 – CHECKLIST OF ASSESSMENT AND REPORTING CRITERIA

Ð	MIRALGA CREEK PROSPECT – JANUARY 2014
CRITERIA	EXPLANATION
SECTION 1 - SAMPLING TECHN	
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Sampling techniques	Reverse Circulation (RC) chip samples collected via cone splitter.
$\mathbb{P}$	• One 6kg (average) sample taken for each two metre sample length and collected in pre- numbered calico sample bags.
	<ul> <li>6kg sample was dried, crushed and pulverised (total prep) to produce a sub sample for analysis for XRF and total LOI by TGA.</li> </ul>
0	<ul> <li>Quality of sampling continuously monitored by field geologist during drilling.</li> </ul>
2)	• To monitor the representivity of the sample, 5 duplicates are taken for every 100 samples (1:20).
	Sampling carried out under Atlas protocols and QAQC procedures as per industry best practice.
Drilling techniques	<ul> <li>Reverse Circulation (RC) drilling employing a 140mm diameter face sampling hammer.</li> <li>Nominal drill spacing of 80mN by 40mE</li> </ul>
Drill sample recovery	RC sample recovery is recorded by the geologist and is based on how much of the sample is
1	returned from the cone splitter. This is recorded as good, fair, poor or no sample.
$\mathbb{D}$	<ul> <li>To ensure maximum sample recovery and the representivity of the samples, the field geologist is present during drilling and monitors the sampling process. Any issues are immediately rectified.</li> </ul>
¥ l	<ul> <li>No significant sample recovery issues were encountered.</li> </ul>
	No twin RC or diamond drillholes have been completed to assess potential sample bias due to
<u>)</u>	preferential loss/gain of fine/coarse material or due to wet drilling.
Logging	<ul> <li>Geological logging is completed for every 2m interval is undertaken corresponding with the 2m sampled interval. This level of detail is sufficient to support future Mineral Resource estimation,</li> </ul>
	mining studies and metallurgical studies should they be undertaken.
	Geophysical data has not been collected for the RC drill holes at time of publication.
Sub-sample techniques and	Sampling technique:
sample preparation	RC Chip Samples:
	~6kg RC chip samples are collected via cone splitter for each 2m interval drilled in a pre-
	numbered calico bag. Samples are kept dry where possible.
	• The sample sizes are considered to be appropriate to correctly represent the mineralisation
	based on the style of mineralisation (massive goethite/hematite), the thickness and
	consistency of intersections, the sampling methodology and percent value assay ranges for the primary elements.
	Sample preparation:
	<ul> <li>Sample preparation.</li> <li>Sample dried at 105°C for 12-24 hrs</li> </ul>
	<ul> <li>Crushed to nominal -3mm</li> </ul>
	<ul> <li>Pulverised to 90% passing at 75µm</li> </ul>

		٠	Quality Control Procedures
			Duplicated sample: 5 every 100 samples (1:20).
			<ul> <li>Certified Reference Material assay standards inserted: 5 in every 100 samples (1:20).</li> </ul>
			Overall QAQC insertion rate of 1:10.
			<ul> <li>Sample weights recorded for all samples.</li> </ul>
			<ul> <li>Lab duplicates taken where large samples required splitting down by the lab.</li> </ul>
			Lab repeats taken and standards inserted at predetermined level specified by the lab.
	Quality of assay data and laboratory tests	•	All samples were submitted to SGS Laboratory in Perth and assayed for the full iron ore suite by XRF (24 elements) and 'loss on ignition' LOI.
$\geq$		•	Laboratory procedures are in line with industry standards and appropriate for iron ore deposits.
		•	Samples are dried at $105^{\circ}$ C in gas fired ovens for 18-24 hours before being crushed to a nominal -3mm size by Boyd crusher, then pulverised to 90% passing 75 micron using a LM2 mill. Sub-samples are collected to produce a 0.66g sample that is dried further, fused at $110^{\circ}$ C for 10 minutes poured into a platinum mould and placed in the XRF machine for analysing and reporting.
		•	LOI is measured by Thermogravimetric methods (TGA).
		•	Certified Reference Material assay standards, field duplicates and umpire laboratory analysis
			are used for quality control.
		•	There were no discernable issues with sample representivity and all duplicate samples were within 10% of the original sample value.
		•	Umpire laboratory campaigns with another laboratory (Ultratrace) have been carried out as independent checks of the assay results and these show good precision.
Л		•	Certified Reference Material assay standards having a good range of values, were inserted at predefined intervals by Atlas and randomly by the lab at set levels. Results highlight that sample assay values are accurate and precise.
		•	Analysis of field duplicate and lab pulp repeat samples reveals that greater than 90% of pairs
			have less than 10% difference and the precision of samples is within acceptable limits, which concurs with industry best practice.
	Verification of sampling and	•	Significant intersections have been independently verified by alternative company personnel.
	assaying	•	The Competent Person has inspected the sampling process in the field and also inspected the
			Laboratory.
	)T	•	Primary data are captured on field Toughbook laptops using acQuire <sup>tm</sup> software. The software
Ų	))		has validation routines to prevent data entry errors.
		•	All data is sent to Perth and stored in the secure, centralised acQuire SQL database which is
			managed by a full time database administrator.
		•	No adjustments or calibrations were made to any assay data used in the estimate, apart from
			resetting below detection values to half positive detection.
	Location of data points	٠	All Collars were surveyed by Altas field personnel using hand held GPS. Elevation values are in
			AHD RL. Expected accuracy is +/-3m for easting and northing and +/-5m for elevation
$\int$	)		coordinates. For cross section preparation drill collar locations were registered to the
7	)	_	topography.
		•	Downhole surveys have not been completed and there is therefore some uncertainty as to the
		•	orientation of the drill hole traces. The grid system for Miralga Creek is MGA_GDA94 Zone 50.
14		•	Landgate commercially available topography with a +/-10m resolution was utilised. Data was
	))	-	supplied in projection MGA_GDA94 Zone 50.
	Data spacing and distribution	•	Drill holes were spaced on an approximate 40m (N-S) by 80m (E-W) grid. But do not completely
			cover the prospect area. A broader coverage will be required to undertake Resource Estimation
_	/		works.
		•	This drill spacing would be sufficient to establish the degree of geological and grade continuity
			applied under the JORC Code (2012).
		٠	Samples were collected at 2m intervals.
	Orientation of data in relation	٠	The attitude of the lithological units is moderately to steeply north dipping and is drilled to the
	to geological structure		south with drillholes inclined between -60 and -90 degrees, slightly oblique to the orientation of
	2		the mineralisation. As such, due to the varying intersection angles all results are defined as downhole widths.
	Sample security	•	Samples are packed into sealed polyweave bags and then placed inside sealed Bulka bags.
	-		Samples are delivered to a despatch point in Port Hedland by Atlas staff.
		•	Chain of custody is managed by Atlas.
		٠	Samples are transported to the relevant Perth laboratory by courier (TOLL).
		•	Once received at the laboratory, samples are stored in a secure yard until analysis.
		•	The lab receipts received samples against the sample dispatch documents and issues a
			reconciliation report for every sample batch.

Audits or reviews	An audit of the Atlas acQuire drillhole database was completed in August 2012 by independent
Addits of Teviews	<ul> <li>An addit of the Atlas acquire dhinkie database was completed in August 2012 by independent database management company (Roredata Pty Ltd).</li> </ul>
	<ul> <li>The Atlas acQuire database is considered to be of sufficient quality.</li> </ul>
	A regular review of the data and sampling techniques is carried out internally.
SECTION 2 - REPORTING OF	EXPLORATION RESULTS
Mineral tenement and land	Exploration Prospects are located wholly within Exploration Leases 100% owned by Atlas
tenure status	<ul> <li>The tenements lie within the Njamal Native Title Claim (WC1999/008).</li> </ul>
	At the time of reporting, there are no known impediments to obtaining a licence to operate in the
	area and the tenement is in good standing.
Exploration done by other	None
parties	
Geology	The prospect is located on the northern margin of the Panorama Greenstone Belt within the East
T.	Pilbara terrane of Western Australia, approximately 100 km southeast of Port Hedland. The
	Miralga Creek BIF-hosted iron ore mineralisation is hosted by the ca. 3.02 Ga Cleaverville
2)	formation (Gorge Creek group, De Grey Supergroup) consisting of a package of banded iron
	formations, cherts and shales.
Drill hole information	Refer to Figure 3, Figure 4 and Attachment 2 – Significant Intercepts.
Data aggregation methods	A nominal 53% lower Fe cut is applied with 6m internal dilution and 6m minimum width for significant intercepts.
Relationship between	• The attitude of the lithological units is moderately to steeply north dipping and is drilled to the
mineralisation widths and	south with drillholes inclined between -60 and -90 degrees, slightly oblique to the orientation of
Intercept lengths	the mineralisation. As such, due to the varying intersection angles all results are defined as
	downhole widths.
Diagrams	A plan view of the collar locations can be seen in Figure 3.
	• Sections through one part of the deposit with stratigraphic and mineralisation interpretations can
	be seen in Figure 4.
Balanced reporting	All results are reported.
Other substantive exploration	Surface Geological (stratigraphic and structural) mapping of the Miralga Creek prospect was
data	completed by Atlas geologists.
	<ul> <li>Routine multi-element analysis of potential deleterious or contaminating substances such as Arsenic, Lead, Zinc and Sulphur is completed for all samples.</li> </ul>
Further work	• Downhole geophysical logging using a range of tools including gyro will be undertaken to
<u>+</u>	confirm drill hole orientations and additional parameters used for geological modelling.
D)	Geological mapping, rock chip sampling and follow up exploration RC drilling will be planned.
	<ul> <li>Infill drilling will be undertaken on the basis of successful results being received.</li> </ul>
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### **ATTACHMENT 2**

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## CORUNNA DOWNS NEW SIGNIFICANT INTERCEPTS AND JORC COMPLIANCE STATEMENTS

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			EAST	NORTH				FROM		WIDTH						
	HOLEID	PROSPECT	GDA94_Z50	GDA94_Z50	RL (m)	AZIMUTH	DIP	(m)	TO (m)	(m)	Fe%	SiO2%	AI2O3%	P%	LOI1000%	S96
ŀ	CDRC0197	RUNWAY	777714.27	7628637.58	391.83	90	-60	56	114	58	57.64	5.26	1.71	0.047	10.4	0.007
_ †	CDRC0199	RUNWAY	777839.92	7628478.05	404.44	90	-60	0	28	28	57.83	5.66	1.39	0.038	9.95	0.01
21	CDRC0200	RUNWAY	777802.56	7628479.65	403.8	90	-60	0	50	50	58.65	3.89	1.71	0.046	10.11	0.011
	CDRC0201	RUNWAY	777761.08	7628478.7	406.88	90	-60	0	38	38	59	3.44	1.9	0.032	10.06	0.014
Ī	CDRC0201	RUNWAY	777761.08	7628478.7	406.88	90	-60	58	98	40	55.12	10.59	0.69	0.028	9.51	0.001
[	CDRC0204	SHARKGULLY	776920.49	7625678.67	476.95	90	-60	0	24	24	59.82	3.38	1.9	0.058	9.04	0.025
[	CDRC0205	SHARKGULLY	776880.71	7625679.33	470.71	90	-60	0	36	36	58.38	4.61	1.59	0.048	10.13	0.025
[	CDRC0206	SHARKGULLY	776798.54	7625680.25	457.92	90	-60	0	46	46	57.04	6.23	2.71	0.21	8.67	0.018
	CDRC0207	SHARKGULLY	776762.18	7625682.69	456.86	90	-60	14	26	12	56.27	8.22	1.67	0.464	8.11	0.009
	CDRC0208	SHARKGULLY	776724.15	7625679.04	455.26	90	-60	8	16	8	53.82	9.54	1.67	0.181	10.96	0.017
	CDRC0208	SHARKGULLY	776724.15	7625679.04	455.26	90	-60	34	58	24	53.92	9.28	2.41	0.133	10.29	0.007
	CDRC0209	SHARKGULLY	776684.78	7625678.58	457.05	90	-60	12	20	8	54.51	11.05	0.86	0.14	9.44	0.027
	CDRC0209	SHARKGULLY	776684.78	7625678.58	457.05	90	-60	48	84	36	58.05	5.62	1.36	0.074	9.62	0.006
	CDRC0210	SHARKGULLY	776642.05	7625678.7	461.44	90	-60	0	14	14	59.09	5.75	1.19	0.071	8.27	0.016
-	CDRC0210	SHARKGULLY	776642.05	7625678.7	461.44	90	-60	40	90	50	56.63	7.41	1.04	0.07	10.16	0.007
ŀ	CDRC0211	SHARKGULLY	776845.79	7625678.53	466.26	90	-60	0	26	26	62.11	2.32	1.44	0.053	6.84	0.03
-	CDRC0212	SHARKGULLY	776799.32	7625598.75	463.66	90	-60	0	16	16	57.07	4.87	2.75	0.084	10.44	0.055
ŀ	CDRC0213	SHARKGULLY	776757.27	7625598.68	461.59	90	-60	0	38	38	58.08	3.64	3.03	0.12	9.79	0.018
ŀ	CDRC0214	SHARKGULLY	776720.31	7625600.02	459.45	90	-60	0	68	68	60.33	2.63	1.66	0.125	8.95	0.01
ŀ	CDRC0215 CDRC0215	SHARKGULLY	776680.75 776680.75	7625599.33 7625599.33	455.71 455.71	90 90	-60 -60	0 40	36 54	36	54.65 57.37	7.3	3.9 2.86	0.106	9.97 10.53	0.011
ŀ	CDRC0215 CDRC0216	SHARKGULLY	776637.22	7625500.73	455.71	90	-60	40	54	54	57.57	3.93 5.56	3.53	0.106	10.55	0.007
ŀ	CDRC0216 CDRC0216	SHARKGULLY	776637.22	7625600.73	451.47	90	-60	64	90	26	56.29	7.55	1.63	0.094	9.93	0.015
ŀ	CDRC0218 CDRC0217	SHARKGULLY	776597.53	7625599.49	454.24	90	-60	0	18	18	57.61	4.22	1.05	0.079	10.92	0.003
ŀ	CDRC0217	SHARKGULLY	776597.53	7625599.49	454.24	90	-60	26	46	20	57.08	4.77	2.58	0.092	10.47	0.005
ŀ	CDRC0217	SHARKGULLY	776597.53	7625599.49	454.24	90	-60	66	76	10	54.23	8.69	3.23	0.046	9.85	0.005
i i	CDRC0218	SHARKGULLY	776558.29	7625599.2	458.24	90	-60	2	18	16	57.17	4.36	2.24	0.162	11.02	0.014
i i	CDRC0219	SHARKGULLY	776682.94	7625516.13	464.49	90	-60	0	16	16	56.04	10.06	1.69	0.073	7.79	0.018
i i	CDRC0220	SHARKGULLY	776518.22	7625439.04	458.49	90	-60	0	12	12	57.62	4.31	2.69	0.076	10.2	0.03
Ī	CDRC0221	SHARKGULLY	776480.73	7625439.08	456.08	90	-60	0	36	36	58.13	4.72	1.87	0.068	10.04	0.017
Ī	CDRC0222	SHARKGULLY	776439.55	7625439.76	453.47	90	-60	2	76	74	57.52	4.16	2.75	0.099	10.17	0.013
Ī	CDRC0223	SHARKGULLY	776398.04	7625440.06	449.21	90	-60	82	108	26	59.23	3.66	0.86	0.046	10.59	0.007
[	CDRC0225	SHARKGULLY	776444.04	7625519.65	449.99	90	-60	36	42	6	55.44	7	1.38	0.174	10.61	0.012
[	CDRC0225	SHARKGULLY	776444.04	7625519.65	449.99	90	-60	86	102	16	59.89	4.13	0.86	0.035	9.1	0.01
	CDRC0225	SHARKGULLY	776444.04	7625519.65	449.99	90	-60	112	170	58	60.52	3.78	1.35	0.068	8.11	0.009
	CDRC0226	SHARKGULLY	776445.41	7625359.33	458.55	90	-60	0	12	12	57.13	6.38	2.11	0.033	9.52	0.017
	CDRC0227	SHARKGULLY	776401.67	7625359.91	457.16	90	-60	0	6	6	54.66	7.32	4.42	0.046	9.39	0.02
	CDRC0229	SHARKGULLY	776319.69	7625358.33	453.39	90	-60	0	6	6	57.05	5.59	1.9	0.052	10.62	0.027
	CDRC0230	SHARKGULLY	776844.18	7625755.07	472.73	90	-60	0	34	34	59.55	3.74	2	0.055	8.86	0.033
-	CDRC0231	SHARKGULLY	776802.21	7625754.5	470.25	90	-60	2	10	8	58.25	6.06	1.51	0.066	8.73	0.03
	CDRC0233	RUNWAY	777799.3	7628398.26	394.7	90	-60	12	20	8	59.22	4.54	3.52	0.037	5.55	0.011
-	CDRC0234	RUNWAY	777766.2	7628393.91	395.37	90	-60	12	24	12	55.58	6.42	4.85	0.033	7.45	0.016
-	CDRC0236	RUNWAY	777897.07	7628557.82	403.3	90	-60	0	14	14	56.84	7.57	2.29	0.045	8.48	0.038
-	CDRC0237	RUNWAY	777864.67	7628557.83	402.23	90	-60	0	30	30	58.53	3.47	1.92	0.031	10.6	0.012
ŀ	CDRC0238	RUNWAY	777798.68	7628558.39 7628554.99	399.88	90 90	-60	0	70	70	57.78	5.48	2	0.04	8.83	0.016
ŀ	CDRC0239		777761.03		405.19		-60 -60	0 38	94	94	58.92	3.68	1.75		9.46	0.016
ŀ	CDRC0240 CDRC0241	RUNWAY	777765.23	7628798.12	387.36	90 90	-60	96	82 126	44	56.83	2.88	2.72 3.03	0.027	11.29	0.014
ŀ	CDRC0241 CDRC0242	RUNWAY	777720.67 777948.63	7628799.4 7628718.03	396.41 399.58	90	-60	0	28	30 28	56.54 56.19	7.59	2.02	0.032	9.88 8.87	0.01
ŀ	CDRC0242 CDRC0243	RUNWAY	777878.9	7628718.05	390.39	90	-60	4	10	6	55.31	5.75	7.45	0.037	6.52	0.015
ŀ	CDRC0245	RUNWAY	777840.87	7628717.67	387.31	90	-60	14	54	40	56.88	6.22	1.78	0.04	10.22	0.006
ŀ	CDRC0245	RUNWAY	777798.61	7628716.34	388.46	90	-60	22	72	50	58.31	6.16	1.19	0.035	8.77	0.008
i i	CDRC0246	RUNWAY	777761.3	7628716.52	390.94	90	-60	20	76	56	57.63	5.59	1.58	0.031	9.9	0.012
	CDRC0246	RUNWAY	777761.3	7628716.52	390.94	90	-60	88	116	28	58.79	5.36	0.57	0.07	9.87	0.006
i i	CDRC0247	RUNWAY	777829.5	7628556.63	396.8	90	-60	0	40	40	59.42	2.8	1.51	0.032	10.23	0.01
ŀ	CDRC0247	RUNWAY	777829.5	7628556.63	396.8	90	-60	50	64	14	59.64	2.49	0.93	0.038	10.54	0.001
ŀ	CDRC0248	RUNWAY	777728.53	7628554.67	404.58	90	-60	20	34	14	55.35	8.36	2.46	0.052	9.81	0.016
i i	CDRC0248	RUNWAY	777728.53	7628554.67	404.58	90	-60	52	110	58	57.95	4.47	1.6	0.034	10.73	0.005
ľ	CDRC0249	RUNWAY	777920.72	7628719.04	395.58	90	-69	0	32	32	53.28	13.13	1.81	0.027	7.25	0.01
ľ	CDRC0250	RUNWAY	777719.6	7628719.15	388.8	90	-60	62	84	22	59.65	3.07	1.35	0.032	10.26	0.006
Ī	CDRC0250	RUNWAY	777719.6	7628719.15	388.8	90	-60	134	182	48	55.71	9.88	0.5	0.031	9.92	0.005
[	CDRC0252	RUNWAY	777682.24	7628642.86	392.17	90	-60	104	114	10	53.96	11.9	1.32	0.054	9.47	0.007
[	CDRC0252	RUNWAY	777682.24	7628642.86	392.17	90	-60	154	172	18	55.6	9.6	0.73	0.033	10.25	0.001

#### New Significant Intercepts at Corunna Downs.

Notes to Corunna Downs Significant Intercepts: Assay results are based on 2 meter samples from cone split RC samples, analysis by XRF with total LOI by Thermo-Gravimetric Analysis. 10% of samples are subject to QAQC procedures (standards and duplicates). Laboratory check samples are routinely performed on each sample submission. Significant Intercepts are reported at a 53% Fe cut-off grade, and include a maximum of 6m internal dilution and 6m minimum width for intersection. Drill holes are spaced on a nominal 80m X 40m grid pattern, with collar locations surveyed by DGPS\_RTK.

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	D12 TABLE 1 – CHECKLIST OF ASSESSMENT AND REPORTING CRITERIA
CRITERIA	CORUNNA DOWNS PROJECT – JANUARY 2014 EXPLANATION
SECTION 1 - SAMPLING TECH	NIQUES AND DATA
Sampling techniques	Reverse Circulation (RC) chip samples collected via cone splitter.
	<ul> <li>One 6kg (average) sample taken for each two metre sample length and collected in pre-</li> </ul>
	numbered calico sample bags.
	<ul> <li>6kg sample was dried, crushed and pulverised (total prep) to produce a sub sample for analysi</li> </ul>
	for XRF and total LOI by TGA.
	Quality of sampling continuously monitored by field geologist during drilling.
	• To monitor the representivity of the sample, 5 duplicates are taken for every 100 samples (1:20)
	Sampling carried out under Atlas protocols and QAQC procedures as per industry best practice
Drilling techniques	Reverse Circulation (RC) drilling employing a 140mm diameter face sampling hammer.
/	Nominal drill spacing of 80mN by 40mE
Drill sample recovery	• RC sample recovery is recorded by the geologist and is based on how much of the sample
	returned from the cone splitter. This is recorded as good, fair, poor or no sample.
))	To ensure maximum sample recovery and the representivity of the samples, the field geologist
9	present during drilling and monitors the sampling process. Any issues are immediately rectified
2	No significant sample recovery issues were encountered.
))	<ul> <li>No twin RC or diamond drillholes have been completed to assess potential sample bias due to profesential least and final assess potential sample bias due to use drilling.</li> </ul>
Logging	<ul> <li>preferential loss/gain of fine/coarse material or due to wet drilling.</li> <li>Geological logging is completed for every 2m interval is undertaken corresponding with the 2n</li> </ul>
rogging	sampled interval. This level of detail is sufficient to support future Mineral Resource estimation
2	mining studies and metallurgical studies should they be undertaken.
	<ul> <li>All holes were downhole geophysical logged (or attempted) for Natural Gamma, Resistivity</li> </ul>
	Gamma Density, Caliper and Magnetic Susceptibility. Not all holes were open at depth which
	precluded 100% coverage of measurements from all of the drillholes.
Sub-sample techniques and	Sampling technique:
sample preparation	RC Chip Samples:
	• ~6kg RC chip samples are collected via cone splitter for each 2m interval drilled in a pre-
	numbered calico bag. Samples are kept dry where possible.
	The sample sizes are considered to be appropriate to correctly represent the mineralisatio
	based on the style of mineralisation (massive goethite/hematite), the thickness an
	consistency of intersections, the sampling methodology and percent value assay ranges for
	the primary elements.
2	Sample preparation:
))	Sample dried at 105 °C for 12-24 hrs
	Crushed to nominal -3mm
	<ul> <li>Pulverised to 90% passing at 75μm</li> </ul>
	Quality Control Procedures
))	Duplicated sample: 5 every 100 samples (1:20).     Cartified Defenses Metarial essay standards inserted. 5 in every 100 samples (1:20)
	<ul> <li>Certified Reference Material assay standards inserted: 5 in every 100 samples (1:20).</li> <li>Overall OAOC insertion rate of 1:10</li> </ul>
2	<ul> <li>Overall QAQC insertion rate of 1:10.</li> <li>Sample weights recorded for all samples.</li> </ul>
)	<ul> <li>Sample weights recorded for all samples.</li> <li>Lab duplicates taken where large samples required splitting down by the lab.</li> </ul>
	<ul> <li>Lab duplicates taken where large samples required splitting down by the lab.</li> <li>Lab repeats taken and standards inserted at predetermined level specified by the lab.</li> </ul>
Quality of assay data and	<ul> <li>All samples were submitted to SGS Laboratory in Perth and assayed for the full iron ore suite b</li> </ul>
laboratory tests	<ul> <li>All samples were submitted to SGS Laboratory in Pertri and assayed for the full non-ore suite to XRF (24 elements) and 'loss on ignition' LOI.</li> </ul>
	<ul> <li>Laboratory procedures are in line with industry standards and appropriate for iron ore deposits.</li> </ul>
0	<ul> <li>Samples are dried at 105°C in gas fired ovens for 18-24 hours before being crushed to</li> </ul>
リー・・・	nominal -3mm size by Boyd crusher, then pulverised to 90% passing 75 micron using a LM
~	mill. Sub-samples are collected to produce a 0.66g sample that is dried further, fused at $110^{\circ}$
	for 10 minutes poured into a platinum mould and placed in the XRF machine for analysing an
	reporting.
	LOI is measured by Thermogravimetric methods (TGA).
	· Certified Reference Material assay standards, field duplicates and umpire laboratory analysi
	are used for quality control.
	• There were no discernable issues with sample representivity and all duplicate samples were
	within 10% of the original sample value.
	• Umpire laboratory campaigns with another laboratory (Ultratrace) have been carried out a

Verification of sam assaying
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Orientation of data i to geological structu
Sample security
Audits or reviews
SECTION 2 - REPOR
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Geology

	<ul> <li>independent checks of the assay results and these show good precision.</li> <li>Certified Reference Material assay standards having a good range of values, were inserted at predefined intervals by Atlas and randomly by the lab at set levels. Results highlight that sample approximately and exercise.</li> </ul>
	<ul> <li>assay values are accurate and precise.</li> <li>Analysis of field duplicate and lab pulp repeat samples reveals that greater than 90% of pairs have less than 10% difference and the precision of samples is within acceptable limits, which</li> </ul>
	concurs with industry best practice.
Verification of sampling and	Significant intersections have been independently verified by alternative company personnel.
assaying	The Competent Person has inspected the sampling process in the field and also inspected the
	Laboratory.
	• Primary data are captured on field Toughbook laptops using acQuire <sup>tm</sup> software. The software
	has validation routines to prevent data entry errors.
	• All data is sent to Perth and stored in the secure, centralised acQuire SQL database which is
	managed by a full time database administrator.
	• No adjustments or calibrations were made to any assay data used in the estimate, apart from
	resetting below detection values to half positive detection.
Location of data points	All collars were surveyed by licensed surveyors (MRH Surveyors, Perth) utilising a RTK GPS
· · · · · · · · · · · · · · · · · · ·	system tied into the state survey mark (SSM) network with the expected relative accuracy of 0.05m E, N & RL. Elevation values are in AHD RL coordinates.
	<ul> <li>The grid system for Corunna Downs is MGA_GDA94 Zone 50.</li> </ul>
D)	• LiDAR topographic data and imagery collected by Outline Global Pty Ltd based on 10cm
U	resolution RGB imagery. 2m vertical contour interval resolution derived from stereoscopic
	imagery DTM. Aerial survey flown on the 16th March 2013. Data supplied in projection
))	MGA_GDA94 Zone 50. The quality and resolution of the topographic data is considered to be
P	adequate for resource estimation purposes
Data spacing and distribution	• Drill holes were spaced on an approximate 80m (N-S) by 40m (E-W) grid. But do not completely
))	cover the prospect area.
P	• This drill spacing would be sufficient to establish the degree of geological and grade continuity
	applied under the JORC Code (2012).
	Samples were collected at 2m intervals.
Orientation of data in relation	• The attitude of the lithological units is moderately to steeply west dipping and is drilled to the
to geological structure	east with drillholes inclined between -60 and -90 degrees, slightly oblique to the orientation of
D)	the mineralisation. As such, due to the varying intersection angles all results are defined as
Ľ	downhole widths.
Sample security	• Samples are packed into sealed polyweave bags and then placed inside sealed Bulka bags.
	Samples are delivered to a despatch point in Port Hedland by Atlas staff.
	Chain of custody is managed by Atlas.
D)	<ul> <li>Samples are transported to the relevant Perth laboratory by courier (TOLL).</li> </ul>
$\mathcal{O}$	Once received at the laboratory, samples are stored in a secure yard until analysis.
	• The lab receipts received samples against the sample dispatch documents and issues a
))	reconciliation report for every sample batch.
Audits or reviews	An audit of the Atlas acQuire drillhole database was completed in August 2012 by independent
	database management company (Roredata Pty Ltd).
	The Atlas acQuire database is considered to be of sufficient quality.
	• A regular review of the data and sampling techniques is carried out internally.
SECTION 2 - REPORTING OF E	
Mineral tenement and land	Exploration Prospects are located wholly within Exploration Leases 100% owned by Atlas
tenure status	The tenements lie within the Njamal Native Title Claim (WC1999/008).
	<ul> <li>At the time of reporting, there are no known impediments to obtaining a licence to operate in the</li> </ul>
	area and the tenement is in good standing.
Exploration done by other	<ul> <li>7 open hole percussion drill holes completed by Geotechnics Australia Ltd (1972), no</li> </ul>
parties	intersections of DSO grade mineralisation were reported, area determined to not be prospective.
D)	<ul> <li>Rock chip sampling, geological mapping and geophysical surveys completed by Gondwana</li> </ul>
U	Resources Pty Ltd (2010), recognized presence of near surface zones of DSO grade iron
	mineralisation.
Geology	• The Corunna Downs Split Rock BIF-hosted iron ore resource is hosted by the ca. 3.02 Ga
	Cleaverville formation (Gorge Creek group, De Grey Supergroup). The prospect is located in
	the Kelly greenstone belt within the East Pilbara terrane of Western Australia, approximately
	170km southwest of Port Hedland. The N-S trending Kelly greenstone belt is bound by the
	Corunna Downs and Shaw granitoid complexes. The Split Rock resource features successive
	macrobands of goethite-hematite rich, high grade (>55 wt% Fe) ore zones associated with
	neighbouring jaspilitic BIF units and banded chert and shale.

Γ	Drill hole information	• Refer to Figures 6, 7 and 8 and Attachment 3 – Significant Intercepts at Corunna Downs.
-	Data aggregation methods	<ul> <li>A nominal 53% lower Fe cut is applied with 6m internal dilution and 6m minimum width for significant intercepts.</li> </ul>
-	Relationship between mineralisation widths and intercept lengths	<ul> <li>The attitude of the lithological units is moderately to steeply west dipping and is drilled to the east with drillholes inclined between -60 and -90 degrees, slightly oblique to the orientation of the mineralisation. As such, due to the varying intersection angles all results are defined as downhole widths.</li> </ul>
	Diagrams	<ul> <li>Sections through the deposits with stratigraphic and mineralisation interpretations can be seen in Figures 7 and 8.</li> </ul>
	Balanced reporting	All results are reported.
	Other substantive exploration data	<ul> <li>Atlas previously reported deposit information for Split Rock including a Mineral Resource Estimate (see Atlas ASX release, Maiden Resource at Corunna Downs, 24 July 2013).</li> <li>Surface Geological mapping (stratigraphy, mineralisation and structure) of the Split Rock prospect was performed by Atlas Geological personnel and Digirock consultants.</li> <li>Routine multi-element analysis of potential deleterious or contaminating substances such as Arsenic, Lead, Zinc and Sulphur is completed for all samples.</li> <li>Geologists from the Centre for Exploration Targeting (CET), University of Western Australia</li> </ul>
	) ) V <sup>c</sup> uther work	<ul> <li>(UWA) are completing research studies on the Corunna Downs Project with focus on the controls on mineralisation. The nature and timing of mineralisation events is also being evaluated through isotopic and geochemical analysis.</li> <li>Preliminary Metallurgical test work based on RC composite samples from a selection of holes has been performed by SGS Lakefield Oretest Pty Ltd. The aim of this test work was to determine preliminary characteristics of the deposite such as particle size distribution, abrasion index, bulk density, moisture and asbestiform mineral analysis.</li> </ul>
	Further work	<ul> <li>5 Geotechnical PQ3 diamond drill holes were recently completed to determine pit design parameters. All diamond core has been geotechnically logged and the holes scanned by televiewer. Results of this analysis are pending at the time of this release.</li> <li>4 of the HQ3 diamond hole sample bulk residues are to be used for bulk materials flow testing, transportable moisture limit and dust extinction level tests. Additional diamond drilling is planned to provide more definitive metallurgical physical properties data such as Cwi, UCS, Ai, bulk density and moisture.</li> <li>Hydrogeology studies to determine dewatering requirements are currently being scoped.</li> <li>Waste classification samples have been collected to assess the nature of potentially acid forming (PAF) sulphidic carbonaceous shale material.</li> <li>A selection of drillholes will be left open for use in subterranean fauna studies.</li> <li>No further RC infill or extensional drilling is planned to be completed on Split Rock as the mineralisation is effectively closed off in all directions except for at depth in a few locations, but this is felt to be too deep and problematic to drill and would realistically be beyond the maximum depth limit of most optimal pits based on the lateral extents of the resource and ore body orientation.</li> <li>Work related to any potential mining development of the Split Rock deposit is dependent on outcomes of scoping level mining studies.</li> </ul>

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## **ATTACHMENT 3**

## CORUNNA DOWNS SPLIT ROCK RESOURCE SIGNIFICANT INTERCEPTS AND JORC COMPLIANCE STATEMENTS

	CDRC0022	776291.7	7623362.26	412.61	90	-60	28	52	24	57.09	4.94	1.92	0.131	10.03	0.018	Previously Reported
	CDRC0023	776241.96	7623365.13	418.25	90	-60	4	94	90	58.74	2.96	1.14	0.118	10.13	0.01	Previously Reported
ŀ	CDRC0023	776241.96	7623365.13	418.25	90	-60	102	120	18	58.93	3.02	1.16	0.086	11.08	0.008	Previously Reported No Significant Intercept
	CDRC0024 CDRC0025	776368.02 776211.92	7623522.32 7623360.79	428.66 417.75	90 90	-60 -60	8	48	40	59.22	4.02	1.26	0.121	8.7	0.008	Previously Reported
	CDRC0025	776211.92	7623360.79	417.75	90	-60	72	170	98	57.77	6.18	1.14	0.084	9.35	0.01	Previously Reported
9	CDRC0026 CDRC0027	776244.57 776298.13	7623440.35 7623520.99	423.5 429.06	90 90	-60 -60	6	126	120	56.86	4.17	2.46	0.127	10.66	0.011	Previously Reported No Significant Intercept
Ē	CDRC0028	776283.71	7623599.5	428.42	90	-60	26	32	6	53.64	6.55	4.73	0.188	11.33	0.018	Previously Reported
-	CDRC0029 CDRC0030	776321.67 776496.67	7623599.97 7623601.4	428.39 420.16	90 90	-60 -60										No Significant Intercept No Significant Intercept
	CDRC0031	776443.41	7623613.51	422.05	90	-60										No Significant Intercept
7	CDRC0032 CDRC0033	776407.77 776364.01	7623606.19 7623601.75	424.84 427.86	90 90	-60 -60										No Significant Intercept No Significant Intercept
J	CDRC0034	776246.24	7623520.12	420.91	90	-60	8	104	96	56.66	4.94	2.63	0.131	10.03	0.009	Previously Reported
$\overline{}$	CDRC0035 CDRC0035	776163.54 776163.54	7623122.3 7623122.3	406.21 406.21	90 90	-85 -85	8 108	72 114	64 6	57.19 54.37	4.2 13.66	2.79 0.64	0.107	10.09 7.71	0.01	Previously Reported Previously Reported
$(\cap$	CDRC0035	776163.54	7623122.3	406.21	90	-85	144	158	14	53.99	13.95	0.41	0.093	8.21	0.007	Previously Reported
リビ	CDRC0035 CDRC0036	776163.54 776242.24	7623122.3 7623123.42	406.21 411.3	90 90	-85 -60	168 0	234 66	66 66	57.31 54.38	12.2 10.76	0.3	0.07	5.48 9.43	0.008	Previously Reported Previously Reported
	CDRC0036	776242.24	7623123.42	411.3	90	-60	82	92	10	54.44	10.69	0.71	0.115	10.18	0.011	Previously Reported
-	CDRC0036 CDRC0037	776242.24 776282.77	7623123.42 7623117.8	411.3 404	90 90	-60 -60	104 4	124 16	20 12	56.02 54.93	8.58 8.53	0.72	0.111 0.199	10.12 10.36	0.006	Previously Reported Previously Reported
15	CDRC0037	776282.77	7623117.8	404	90	-60	26	76	50	58.33	4.19	0.69	0.094	10.30	0.005	Previously Reported
$\mathbb{D}$	CDRC0038 CDRC0039	776319.82 776171.68	7623119.57 7623519.84	401.52 387.52	90 90	-75 -60	0	42	42	55.53	8.79	0.86	0.103	10.32	0.008	Previously Reported No Significant Intercept
$\leq$	CDRC0040	776167.46	7623443.45	393.98	90	-60										No Significant Intercept
7	CDRC0041 CDRC0041	776156.72 776156.72	7623359.67 7623359.67	407.37 407.37	90 90	-60 -60	114 190	126 228	12 38	56.41 58.34	12.06 5.21	0.68	0.077	6.1 9.51	0.005	Previously Reported Previously Reported
$\square$	CDRC0042	776123.42	7623206.07	407.66	90	-70	54	96	42	56.95	9.98	0.59	0.075	7.44	0.005	Previously Reported
ŀ	CDRC0043 CDRC0043	776115.92 776115.92	7623120.24 7623120.24	403.42 403.42	90 90	-85 -85	48 102	74 168	26 66	53.45 56.07	12.47 7.31	2.73 2.62	0.084	7.87 9.52	0.015	Previously Reported Previously Reported
Ē	CDRC0044	776094.59	7623045.48	420.01	90	-80	102	100	00	50.07	7.51	2.02	0.050	5.52	0.015	No Significant Intercept
	CDRC0045 CDRC0046	776062.51 776207.03	7622959.18 7623113.53	415.99 412.34	90 90	-60 -75	0	92	92	60.46	3.47	1.17	0.114	8.44	0.007	No Significant Intercept Previously Reported
_	CDRC0047	776101.76	7623279.29	407.94	90	-60	120	162	42	55.51	12.24	0.76	0.133	7.24	0.005	Previously Reported
P	CDRC0048 CDRC0048	776276.33 776276.33	7623442.29 7623442.29	421.17 421.17	90 90	-60 -60	2 44	34 84	32 40	55.79 54.2	4.51 7.85	3.31 3.09	0.174	11.16 10.07	0.008	Previously Reported Previously Reported
$\square$	CDRC0049	776334.41	7623448.56	426.18	90	-60				57.2	7.00	5.03	0.007	20.07	0.023	No Significant Intercept
-	CDRC0050 CDRC0051	776490.49 776334.3	7623121.57 7622877.53	429.52 412.2	90 <b>90</b>	-60 -75	8	54	46	57.66	4.07	1.69	0.177	10.64	0.006	No Significant Intercept New Infill Hole
ļ	CDRC0052	776346.96	7622803.93	419.29	90	-60	0	14	14	53.28	8.96	4.28	0.042	8.92	0.007	New Infill Hole
-	CDRC0053 CDRC0054	776333.53 776249.8	7622799.84 7622803.56	418.9 415.6	90 90	-75 -60										No Significant Intercept No Significant Intercept
Ļ	CDRC0055	776198.05	7622801.91	421.49	80	-60				_					_	No Significant Intercept
ŀ	CDRC0056 CDRC0057	776162.13 776042.99	7622799.83 7622800.02	424.07 418.85	90 0	-60 -60	0	8	8	57.45	5.46	1.98	0.057	10.17	0.047	New Infill Hole No Significant Intercept
	CDRC0057	776251.83	7622883.96	397.5	90	-60	2	30	28		-			9.51	0.006	

- [	CDRC0060	776283.98	7623042.05	401.6	90	-62.5	0	48	48	54.22	9.4	1.99	0.176	10.1	0.008	New Infill Hole
[	CDRC0061	776246.28	7623441.77	423.48	90	-80	8	220	212	59.97	3.56	1.2	0.132	8.54	0.008	New Infill Hole
ſ	CDRC0062	776239.08	7623518.94	420.29	0	-90	4	66	62	58.27	5.01	1.86	0.121	8.96	0.007	New Infill Hole
Ī	CDRC0063	776163.23	7623279.17	410.59	90	-60	6	12	6	57.03	6.69	2.35	0.052	6.79	0.028	New Infill Hole
ľ	CDRC0063	776163.23	7623279.17	410.59	90	-60	50	56	6	56.01	9.21	0.62	0.164	8.21	0.001	New Infill Hole
ŀ	CDRC0063	776163.23	7623279.17	410.59	90	-60	66	82	16	58.51	8.98	0.51	0.098	6.1	0.008	New Infill Hole
ŀ	CDRC0064	776129.6		413.16	90	-60	10	88		58.45	4.43	2.47	0.083	8.8	0.008	
ŀ			7623079.81						78							New Infill Hole
ŀ	CDRC0064	776129.6	7623079.81	413.16	90	-60	108	186	78	60.25	3.99	0.72	0.105	8.8	0.009	New Infill Hole
	CDRC0065	776078.03	7623121.91	403.71	90	-80										No Significant Intercept
	CDRC0066	776158.2	7623201.15	402.75	90	-60	6	60	54	57.76	5.68	1.71	0.098	8.66	0.009	New Infill Hole
	CDRC0067	776240.9	7623201.75	407.24	90	-60										No Significant Intercept
	CDRC0068	776087.68	7623198.9	404.1	90	-60										No Significant Intercept
	CDRC0069	776141.5	7623365.87	404.96	90	-70										No Significant Intercept
	CDRC0070	776273.34	7623595.76	428.66	0	-90	10	82	72	56.1	4.68	3.32	0.116	11.04	0.009	New Infill Hole
	CDRC0071	776278.83	7623559.37	429.81	90	-60	16	40	24	54.71	6.38	3.71	0.075	11.05	0.017	New Infill Hole
	CDRC0072	776269.75	7623558.28	429.7	0	-90	10	94	84	56.23	5.95	3.75	0.085	8.66	0.011	New Infill Hole
-	CDRC0073	776240.63	7623479.47	421.35	90	-75	0	112	112	59.77	4.46	1.16	0.119	7.87	0.008	New Infill Hole
	CDRC0074	776278.85	7623480.16	421.33	90	-75	4	84	80	56.81	4.40	3.02	0.136	10.55	0.005	New Infill Hole
	-															
ŀ	CDRC0075	776288.55	7623399.72	414.83	90	-60	46	60	14	57.57	4.49	1.61	0.104	10.67	0.011	New Infill Hole
	CDRC0076	776237.41	7623399.13	421.94	90	-80	2	66	64	59.24	3.8	1.65	0.178	8.74	0.011	New Infill Hole
į.	CDRC0076	776237.41	7623399.13	421.94	90	-80	90	112	22	60.15	6.21	0.56	0.136	6.67	0.015	New Infill Hole
_	CDRC0076	776237.41	7623399.13	421.94	90	-80	138	232	94	58.38	3.65	1.3	0.109	10.75	0.011	New Infill Hole
	CDRC0077	776243.66	7623318.63	414.32	90	-60	4	124	120	57.18	6.13	0.92	0.133	9.62	0.009	New Infill Hole
	CDRC0078	776199.73	7623316.37	413.44	90	-60	4	72	68	59.95	3.7	0.86	0.112	7.96	0.007	New Infill Hole
	CDRC0078	776199.73	7623316.37	413.44	90	-60	100	120	20	57.86	6.48	0.82	0.071	9.58	0.007	New Infill Hole
$\neg$	CDRC0078	776199.73	7623316.37	413.44	90	-60	180	186	6	57.03	9.69	0.4	0.052	7.76	0.041	New Infill Hole
ļ	CDRC0078	776199.73	7623316.37	413.44	90	-60	196	226	30	57.45	5.91	1.24	0.101	10.06	0.02	New Infill Hole
ţ	CDRC0079	776268.98	7623636.04	424.78	0	-90	12	24	12	53.61	6.4	4.45	0.156	11.21	0.012	New Infill Hole
$\sim$	CDRC0079	776268.98	7623636.04	424.78	0	-90	32	40	8	53.79	7.37	3.73	0.163	10.9	0.008	New Infill Hole
$\leq$	CDRC0080	776279.91	7623632.92	424.72	90	-60	18	40	24	56.01	6.11	1.89	0.198	10.53	0.000	New Infill Hole
	CDRC0081	776269.01	7623677.9	420.79	0	-90	-0	~		50.01		2.35	0.150	20.00	0.011	No Significant Intercept
	CDRC0081 CDRC0082	776279.83	7623677.91	420.79	90	-90	4	20	16	54.53	7.62	2.58	0.191	10.76	0.008	No Significant Intercept New Infill Hole
$\leq$							2	20								,
	CDRC0083	776314.57	7623673.67	425.21	90	-60	2	10	8	56.61	4.45	2.82	0.088	11.15	0.012	New Infill Hole
	CDRC0084	776321.36	7623638.93	427.28	90	-60			-		0.17		0.455	0.55		No Significant Intercept
$ \land$	CDRC0085	776240.82	7623159.03	406.86	90	-60	6	14	8	55.84	8.17	1.17	0.198	9.69	0.006	New Infill Hole
Ч	CDRC0085	776240.82	7623159.03	406.86	90	-60	142	150	8	58.13	4.08	1.32	0.187	10.82	0.007	New Infill Hole
	CDRC0086	776198.33	7623159.23	408.35	90	-60	2	22	20	53.73	15.12	1.47	0.105	5.02	0.01	New Infill Hole
	CDRC0086	776198.33	7623159.23	408.35	90	-60	180	194	14	56.26	7.39	1.14	0.075	10.64	0.006	New Infill Hole
	CDRC0087	776158.89	7623159.16	402.55	90	-60	0	46	46	57.64	7.76	1.27	0.096	7.42	0.009	New Infill Hole
_	CDRC0087	776158.89	7623159.16	402.55	90	-60	88	110	22	55.99	11.9	0.4	0.095	7.37	0.006	New Infill Hole
	CDRC0088	776115.9	7623010.99	416.33	90	-60	0	80	80	58.26	6.73	2.41	0.055	6.56	0.01	New Infill Hole
	CDRC0088	776115.9	7623010.99	416.33	90	-60	122	142	20	59.27	6.01	0.76	0.113	8.17	0.006	New Infill Hole
	CDRC0089	776081.04	7622999.57	418.6	90	-60	30	88	58	59.66	6.61	1.25	0.064	5.6	0.007	New Infill Hole
_	CDRC0089	776081.04	7622999.57	418.6	90	-60	96	106	10	56.47	8.42	0.61	0.066	10.03	0.007	New Infill Hole
_	CDRC0090	776282.04	7623079.83	403.46	90	-65	0	44	44	59.1	2.54	1.12	0.197	11.04	0.008	New Infill Hole
	CDRC0091	776239.29	7623079.3	410.65	90	-60	0	68	68	57.38	4.76	1.79	0.213	10.21	0.01	New Infill Hole
1	CDRC0091	776239.29	7623079.3	410.65	90	-60	90	100	10	58.81	3.87	0.54	0.173	10.98	0.009	New Infill Hole
$\neg$	CDRC0091	776239.29	7623079.3	410.65	90	-60	110	134	24	56.39	7.89	0.61	0.134	10.37	0.008	New Infill Hole
-	CDRC0092	776203.68	7623081.15	413.04	90	-60	0	94	94	58.61	5.57	1.38	0.138	8.4	0.007	New Infill Hole
- j	CDRC0092 CDRC0092	776203.68	7623081.15	413.04	90	-60	128	94 146	94 18	57.57	6.34	0.72	0.158	10.09	0.007	New Infill Hole
ŀ							32		72		17.18	0.64				
-	CDRC0093 CDRC0094	776113.8 776321.03	7623159.55 7623198.44	393.59 416.62	90 90	-60 -60	0	104 52	52	53.84 54.54	7.61	3.23	0.066	4.56 10.01	0.007	New Infill Hole New Infill Hole
	-		7623198.44				0	52	52	34.34	7.01	3.23	0.175	10.01	0.01	No Significant Intercept
	CDRC0095	776289.39		410.39	90	-60	~			<b>CO 40</b>	<i></i>	2.42	0.040	2.24	0.001	° .
	CDRC0096	776160.29	7623315.09	404.7	90	-60	6	14	8	60.48	6.44	3.42	0.018	3.34	0.001	New Infill Hole
	CDRC0096	776160.29	7623315.09	404.7	90	-60	58	76	18	55.42	9.93	1.04	0.121	6.8	0.005	New Infill Hole
	CDRC0096	776160.29	7623315.09	404.7	90	-60	102	152	50	59.33	7.04	0.51	0.087	7.3	0.006	New Infill Hole
$\rightarrow$	CDRC0100	776277.12	7623010.19	400.37	90	-60	0	50	50	55.46	7.23	2	0.18	10.3	0.008	New Infill Hole
$\neg$	CDRC0101	776238.98	7622999.32	404.5	90	-60	2	56	54	57.22	5.28	1.6	0.159	10.18	0.005	New Infill Hole
	CDRC0101	776238.98	7622999.32	404.5	90	-60	64	92	28	53.11	13.1	1.13	0.119	9.37	0.005	New Infill Hole
r l	CDRC0102	776199.73	7622998.82	399.86	90	-60	0	130	130	58.27	5.59	1.27	0.12	9.14	0.009	New Infill Hole
_	CDRC0103	776156.29	7622999.35	404	90	-60	6	56	50	56.02	10.49	2.16	0.103	5.85	0.031	New Infill Hole
ļ	CDRC0103	776156.29	7622999.35	404	90	-60	88	124	36	58.82	7.98	0.6	0.12	6.96	0.007	New Infill Hole
- [	CDRC0104	776166.99	7623079.05	409.61	90	-60	32	98	66	60.59	3.43	0.81	0.105	8.64	0.007	New Infill Hole
Ч	CDRC0104	776166.99	7623079.05	409.61	90	-60	118	140	22	54.2	12.21	0.51	0.078	9.45	0.006	New Infill Hole
$\overline{}$	CDRC0105	776336.58	7622838.19	415.89	90	-75										No Significant Intercept
Л	CDRC0106	776243.14	7622836.48	410.72	90	-60										No Significant Intercept
$\supset$	CDRC0107	776201.58	7622833.85	416.45	90	-60										No Significant Intercept
_ [	CDRC0108	776306.15	7622919.77	395.37	90	-60	6	26	20	57.47	3.81	2.08	0.22	10.53	0.01	New Infill Hole
1	CDRC0109	776282.72	7622918.44	394.38	90	-60	10	52	42	57.79	4.73	1.21	0.185	10.17	0.008	New Infill Hole
	CDRC0110	776243.87	7622919.12	395.74	90	-60	8	40	32	55.75	5.58	3.03	0.126	9.1	0.009	New Infill Hole
1	CDRC0111	776202.03	7622917.24	396.23	90	-60	4	90	86	56.94	6.28	1.45	0.135	9.85	0.006	New Infill Hole
t	CDRC0112	776188.36	7622884.14	401.44	90	-60										No Significant Intercept
ţ	CDRC0113	776160.18	7622917.05	401.46	90	-60	4	90	86	59.64	3.79	1.11	0.082	9.1	0.008	New Infill Hole
ţ	CDRC0114	776120.82	7622917.09	401.21	90	-60	30	76	46	59.66	5.09	0.85	0.071	7.99	0.007	New Infill Hole
_ †	CDRC0115	776089.91	7622918.16	402.86	90	-60		-	-							No Significant Intercept
ļ	CDRC0116	776316.02	7623158.95	405.93	90	-80	0	62	62	56.75	6.03	1.37	0.15	10.59	0.006	New Infill Hole
_	CDRC0117	776277.22	7623239.27	410.4	90	-60	•	~-	~-		5.00	/	5.20	_0.00		No Significant Intercept
	CDRC0118	776239.29	7623239.29	408.09	90	-60	10	20	10	58.41	7.37	0.82	0.173	7.15	0.001	New Infill Hole
	CDRC0118 CDRC0119	776199.42	7623238.81	408.09	90	-60	2	38	36	57.78	4.81	1.86	0.173	8.64	0.001	New Infill Hole
	CDRC0119 CDRC0120	776158.39	7623238.74	409.32	90	-60	8	68	60	57.68	7.27	1.00	0.137	7.69	0.005	New Infill Hole
$\neg$	CDRC0120 CDRC0121	776281.9	7623318.91	407.8	90	-60	34	66	32	57.68	9.37	0.93	0.125	10.08	0.007	New Infill Hole
ŀ	CDRC0121 CDRC0141	776157.15	7623318.91	398.51	90	-60	JH	00	32	54.33	3.31	0.93	0.123	10.00	0.009	No Significant Intercept
ŀ	CDRC0141 CDRC0143															
_ }		776410.7	7622719.28	434.32	90	-60		24	20	EE 20	0.30	1	0.040	10.45	0.01	No Significant Intercept
ļ	CDRC0144	776375.64	7622723.94	434.19	90	-60	4	34	30	55.38	8.29	1.57	0.049	10.45	0.01	New Infill Hole
ŀ	CDRC0145	776324.34	7622729.3	424.06	90	-60	-	67		F0		4.55		40.55		No Significant Intercept
ŀ	CDRC0149	776245	7623400	422	90	-60	2	92	90	58.29	2.72	1.55	0.131	10.93	0.008	New Infill Hole
ŀ	CDRC0149	776245	7623400	422	90	-60	108	136	28	58.42	4.37	0.93	0.071	10.84	0.008	New Infill Hole
ļ	CDRC0150	776230	7623400	422	0	-90	6	48	42	60.34	3.89	1.35	0.131	7.42	0.013	New Infill Hole
ļ	CDRC0150	776230	7623400	422	0	-90	58	72	14	57.88	4.52	0.81	0.136	10.32	0.009	New Infill Hole
ļ	CDRC0160	776146.06	7623233.53	408.08	90	-80	48	98	50	58.74	6.83	0.83	0.074	7.42	0.009	New Infill Hole
L	CDRC0161	776079.17	7623079.42	409.26	90	-60	74	86	12	59.32	5.2	1.33	0.077	8.24	0.011	New Infill Hole
- 1	CDRC0161	776079.17	7623079.42	409.26	90	-60	116	186	70	59.85	5	1.19	0.095	7.85	0.016	New Infill Hole
ŀ																
t	CDRC0162	776082.09	7623165.07	392.1	90	-60	102	120	18	55.29	13.85	0.18	0.056	6.72	0.012	New Infill Hole

CDRC0165	776310	7623240	417	90	-60	14	58	44	55.95	6.29	2.32	0.146	9.16	0.015	New Infill Hole
CDRC0166	776305	7623720	415	0	-90										No Significant Intercept
CDRC0167	776320	7623720	422	90	-60										No Significant Intercept
CDRC0168	776205	7623358	419	90	-75	12	72	60	57.02	7.76	0.65	0.124	8.22	0.009	New Infill Hole
CDRC0168	776205	7623358	419	90	-75	156	180	24	58.39	4.89	0.68	0.106	10.14	0.006	New Infill Hole
CDRC0174	776240	7623120	411	90	-60	0	76	76	57.45	5.36	1.61	0.167	9.97	0.007	New Infill Hole

<u>Notes to Significant Intercepts</u>: Assay results are based on 2 meter samples from cone split RC samples, analysis by XRF with total LOI by Thermo-Gravimetric Analysis. 10% of samples are subject to QAQC procedures (standards and duplicates). Laboratory check samples are routinely performed on each sample submission. Significant Intercepts are reported at a 53% Fe cut-off grade, and include a maximum of 6m internal dilution and 6m minimum width for intersection. Drill holes are spaced on a nominal 40m X 40m grid pattern, with collar locations surveyed by licensed surveyor using RTK\_GPS. New Intercepts from the recently completed infill drilling are presented in bold font, intercepts for holes CDRC0001 through CDRC0050 have been previously reported by Atlas and holes without significant intercepts are reported as 'No Significant Intercept'.

The Mineral Resource estimate has been completed in accordance with the guidelines of the JORC Code (2012 edition). The geological model and estimation was completed by Atlas and internally reviewed, as described in the JORC (2012 edition) "Table 1 Checklist of Assessment and Reporting Criteria". The Split Rock resource is classified in the Indicated and Inferred Mineral Resource category. The resource at a range of cut-off grades is shown in Attachment 4.

JORC CODE 2012 EDITION – TABLE 1						
CORUNNA DOWNS SPLIT ROCK RESOURCE - JANUARY 2014						
EXPLANATION						
IQUES AND DATA (Criteria in this section apply to all succeeding sections)						
<ul> <li>Reverse circulation (RC) drilling was used to obtain 2.0m down hole interval samples. The samples were passed through a cone splitter to collect a nominal 4.0-6.0kg sample (approximately 10% split ratio) into pre-numbered calico bags.</li> <li>3 RC holes subjected to sample weight and split analysis to ensure the minimum 10% split ratio is being consistently achieved plus these holes were also duplicate sampled to check sampling representivity over the entire length of the holes.</li> <li>4 HQ3 diamond twin holes were sampled at 1m intervals, with the whole core submitted to the laboratory for comparison back to RC samples.</li> <li>Duplicate samples taken at a set frequency of one every twenty samples (5% of total samples) from the cone splitter to monitor sampling representivity.</li> <li>Geophysical gamma density measurements collected downhole by ABIMS geophysical contractor using a Geovista Dual Density logging tool (Cesium source, density range 1-3.5g/cc) to ascertain approximate in-situ density values. Tool is regularly calibrated every 2 weeks using a range of known media and a calibration hole.</li> </ul>						
<ul> <li>Reverse Circulation drilling employing a 140mm diameter face sampling hammer. A nominal drillhole spacing of 40mN x 40mE has been completed for this resource update. A total of 134 RC holes for 19,360m have been drilled.</li> <li>5 HQ3 diamond drillholes for 1,187m have been drilled. HQ3 diamond core runs are orientated by Reflex orientation tool.</li> </ul>						
<ul> <li>RC sample recovery is logged at the drill site by the geologist based on the volume of sample returned from the cone splitter. This is recorded as either good, fair, poor or no sample recovered. Of the total 9,680 RC samples collected, 9,513 (98.3%)were recorded as Good, 70 (0.7%) were recorded as fair, 91 (0.9%) were recorded as poor and 6 (0.1%) were recorded as No Sample return</li> <li>All samples are weighed at the laboratory to continually monitor and record sample size. 3 RC holes were duplicate sampled for every interval down hole and also had the entire sample volume presenting to the splitter weighed to ensure appropriate sample split ratio was achieved through the splitter and the samples were of a representative size.</li> <li>To ensure maximum sample recovery and representivity of the samples, the field geologist was present during drilling, continuously monitoring the sampling process. Any issues were immediately rectified.</li> <li>4 HQ3 diamond twin holes have been used for comparison to RC holes to check for any bias introduced by the drilling technique. The diamond core and RC results compare closely for the top 80m of the holes, however poor recovery was experienced in the diamond holes below this depth due to the friable nature of the material and the sample was deemed to not be representative of the interval and therefore a valid comparison could not be made. Below 80m depth, the RC holes consistently show slightly lower Fe grade and higher contaminant grades than the diamond holes indicating that the diamond drilling may be washing out fines during the drilling process and preferentially upgrading the sample.</li> <li>Atlas is satisfied that the RC holes have taken a sufficiently representative sample of the mineralisation and minimal loss of fines has occurred in the RC drilling resulting in minimal</li> </ul>						

	No relationship between sample recovery and grade has been demonstrated.
Logging	<ul> <li>Logging of every 2m interval corresponding with 2m sampled interval. This level of detail supportive and appropriate for Mineral Resource estimation, mining and metallurgical studies a bulk commodity such as iron ore.</li> </ul>
	<ul> <li>Core and RC logging is qualitative and quantitative in nature.</li> <li>BC Logging records the abundance/protection of aposition minorele/material types and lithelesis</li> </ul>
	<ul> <li>RC Logging records the abundance/proportion of specific minerals/material types and lithologic hardness recorded by physical chip percent measurement, weathering and colour. Additional diamond core was logged for density (dimensional tray method), geotechnical conditions, RC</li> </ul>
	and structure and each tray was photographed both wet and dry after meter marking a orientation.
	<ul> <li>The entire lengths of RC holes were logged on a 2m interval basis, 100% of the drilling w logged. Where no sample was returned due to voids/cavities it is recorded as such. Drill co was also logged over its entire length and core recovery recorded.</li> </ul>
	· All holes were downhole geophysical logged (or attempted) for Natural Gamma, Resistivi
	Gamma Density, Caliper and Magnetic Susceptibility. Not all holes were open at depth whi precluded 100% coverage of measurements from all of the drillholes.
Sub-sampling techniques and sample preparation	<ul> <li>HQ3 diamond core - whole core was sampled at 1m intervals and despatched to the lab where was dried for 12 hours at 105°C, primary crushed down to 8mm fraction and secondary crush</li> </ul>
	to 4mm before being further split down using a rotary splitter to produce a sub-sample approximately 3.5kg before pulverizing in a LM2 mill to a nominal 90% passing 75 micron. A 7 pulp sample is obtained for XRF analysis.
$\bigcirc$	<ul> <li>1:10 of the coarse crushed samples were duplicate sampled by the lab to ensure samp homogeneity and monitor the additional splitting stage performed by the lab and approximate</li> </ul>
6	<ul><li>1:20 pulp samples are duplicated by the lab.</li><li>All RC samples were collected on two meter down hole intervals passed through a cone split</li></ul>
2	to collect a nominal 4.0kg-6.0kg sample. The majority of samples are reported as dry, however proportion of below water table samples are reported as being moist or wet. Of the 9,680 F samples collected 5,175 (53%) reported as dry, 1,043 (11%) moist and 3,456 (36%) as wet and
	<ul> <li>Where RC samples were considered to be large (&gt;6kg), they were crushed down to 3mm fracti and rotary split down to produce a smaller sample suitable for pulverizing. Coarse duplicates a table be the last entitie of 410 k member that the suitable for pulverizing.</li> </ul>
	<ul><li>taken by the lab at a ratio of 1:10 to monitor this process.</li><li>Sample weight/split analysis shows that on average at least 10% split ratio is being achiev</li></ul>
	<ul><li>consistently through the cone splitter primary and duplicate sampling ports.</li><li>Duplicate sample analysis show the data has acceptable precision, indicating that the sampli</li></ul>
۲.	<ul><li>technique is appropriate for the deposit</li><li>Diamond twin analysis also shows good precision where core recovery has been sufficient</li></ul>
	provide a representative sample of the interval.
	<ul> <li>The sample sizes were considered to be appropriate to correctly represent the mineralisati (massive goethite/hematite), the thickness and consistency of intersections, the sampli methodology and percent values assay ranges for the primary elements.</li> </ul>
Quality of assay data and laboratory tests	<ul> <li>All samples submitted to SGS Laboratory in Perth and assayed for the extended iron ore su (24 elements) by XRF and a total LOI by thermogravimetric technique. The method used designed to measure the total amount of each element in the sample.</li> </ul>
9 2	<ul> <li>Samples were subjected to routine particle sizing analysis by the lab to ensure the pulverizi stage is achieving appropriate particle size for XRF analysis showed acceptable results. The analysis shows that 95% of samples tested returned greater than the 90% passing 75 micr</li> </ul>
	<ul> <li>Atlas inserts commercially available certified reference material (standards) at a set frequency 1:20 (5% of total samples) within its sample batches. A number of different standards at a ran of grades are used to monitor analytical precision of the assay results.</li> </ul>
	· Blanks are not used by Atlas due to the nature of the analysis being a complete multi-eleme
	<ul> <li>Acceptable levels of precision have been achieved with all standard assays reporting within Acceptable levels of precision have been achieved with all standard assays reporting within</li> </ul>
-	<ul> <li>standard deviations of the certified mean grade for the 12 main elements of interest.</li> <li>The lab also inserts its own standards at set frequencies and monitors the precision of the XI analysis. These results also reported well within the specified 2 standard deviations of the mean standards for all 40 main elements of interest.</li> </ul>
5	<ul><li>grades for all 12 main elements of interest.</li><li>The Laboratory performs repeat analyses of sample pulps at a rate of 1:20 (5% of all sample</li></ul>
U Contraction of the second se	<ul> <li>these compare very closely with the original analysis for all elements.</li> <li>Analysis of field duplicate and lab pulp duplicates and repeats reveals that greater than 90% pairs have less than 10% difference and the precisions of samples is within acceptable limits a</li> </ul>
	<ul> <li>Atlas sent a selection of pulps to an umpire laboratory (Bureau Veritas, Perth) for verification</li> </ul>
	an independent laboratory. Comparison of results between laboratories did not reveal any issu and analytical precision was considered acceptable.
	<ul> <li>Laboratory procedures are in line with industry standards and are appropriate for iron c analysis.</li> </ul>
Verification of sampling and assaying	<ul> <li>Significant intersections have been independently verified by alternative company personn Drill core and RC chips have been inspected in the field to verify the correlation of mineralis zones with assay results. The Competent Person for this report has visited site and inspected</li> </ul>

	<ul> <li>sampling processes in the field and also inspected the laboratory on a regular basis.</li> <li>4 HQ3 diamond twin holes have been drilled for comparison with RC drillholes and quantitatively analysed with no issues identified.</li> <li>All primary data is captured electronically on field Toughbook laptops using acQuire<sup>Im</sup> software. The software has built in validation routines to prevent data entry errors at the point of entry. Data is also validated prior to export from the Toughbook and again on import into the main corporate acQuire database.</li> <li>All data is sent to Perth and stored in a secure, centralised acQuire SQL database which is administered by a full database administrator.</li> <li>Documentation related to data custody, validation and storage are maintained on the company's server.</li> <li>No adjustments or calibrations were made to any assay data used in the estimate, apart from resetting below detection level values to half positive detection.</li> </ul>
Location of data points	<ul> <li>All collars except 2 were surveyed by licensed surveyors (MRH Surveyors, Perth) utilising a RTK GPS system tied into the state survey mark (SSM) network with the expected relative accuracy of 0.05m E, N &amp; RL. Elevation values are in AHD RL.</li> <li>2 collars were surveyed with handheld GPS with expected accuracy of +/-5m.</li> <li>The grid system for the Corunna Downs Project and the Split Rock resource is MGA_GDA94_Z50.</li> <li>Downhole gyroscopic surveys are attempted on all RC and diamond holes by ABIMS geophysical contractors. Readings are taken at 5m intervals downhole using a SPT north seeking gyroscopic survey tool with a stated accuracy of +/-1° in azimuth and +/-0.1° in inclination. QC of the gyro tool involved field calibration using a test stand and also a calibration hole.</li> <li>LiDAR topographic data and imagery collected by Outline Global Pty Ltd based on 10cm resolution RGB imagery. 2m vertical contour interval resolution derived from stereoscopic imagery DTM. Aerial survey flown on the 16<sup>th</sup> March 2013. Data supplied in projection MGA_GDA94 Zone 50. The quality and resolution of the topographic data is considered to be adequate for resource estimation purposes</li> </ul>
Data spacing and distribution	<ul> <li>RC Drill spacing is on an approximate 40m (N-S) by 40m (E-W) grid, however due to topographic constraints this is sometimes not achievable.</li> <li>This drill spacing and distribution is sufficient to establish the degree of geological and grade continuity appropriate to support an Inferred/Indicated resource classification under the 2012 JORC code and is suitable for this style of deposit.</li> <li>Sample compositing has not been applied to the RC samples used in the resource estimate; all RC samples are collected at 2m intervals. Diamond samples were composited to 2m length to match the RC sample length and maintain equal weighting for comparison purposes, no diamond sample/assays were used in this estimate or for reporting of significant intercepts.</li> <li>Geophysical density measurements collected at 10cm increments were composited up to 2m intervals to correspond with the sample length. The compositing process was checked to ensure that no changes to the statistical population had been incurred due to the compositing process.</li> </ul>
Orientation of data in relation to geological structure Sample security	<ul> <li>The attitude of the Split Rock resource is dominantly steeply west dipping from 70-80 degrees and is drilled to grid east with drillholes inclined between -60 and -90 degrees which is slightly oblique to the orientation of the mineralisation. Structural logging of orientated drill core and surface mapping supports the drilling direction and sampling orientation. Due to the varying intersection angles all intercept results are reported as downhole widths and not true widths.</li> <li>No drilling orientation and sampling bias has been recognized at this time and is not considered to have introduced a sampling bias.</li> <li>Chain of custody is managed by Atlas. Pre-numbered calico sample bags are packed into sealed and the balled balled.</li> </ul>
Audits or reviews	<ul> <li>and labelled polyweave bags on site and then placed inside sealed and labelled bulka bags. Samples are delivered to a dispatch point in Port Hedland by Atlas Staff and a consignment number issued by the transport company (TOLL). Samples are transported to the relevant laboratory in Perth by courier. Once received at the laboratory, the consignment of samples is receipted against the sample dispatch documents and a reconciliation report is issued to Atlas for every sample batch. Samples are stored in a secure yard at the lab until analysis.</li> <li>Sample security was not considered a significant risk to the project.</li> </ul>
	<ul> <li>A detailed audit of the Atlas acQuire drillhole database is performed regularly by independent database management consultants (rOREdata Pty Ltd). The last audit was completed in August 2012 and the database is considered to be of a high standard and acceptable for JORC compliant resource estimation activities.</li> <li>A review of all the resource drillhole data and sampling techniques is carried out internally as part of the resource estimation process.</li> </ul>
SECTION 2 - REPORTING OF E Mineral tenement and land tenure status	<ul> <li><b>EXPLORATION RESULTS (Criteria listed in the preceding section also apply to this section)</b></li> <li>The Split Rock resource is located wholly within Exploration Lease E45/3321. The tenement is 100% Atlas owned.</li> <li>The tenement sits within the Njamal Native Title Claim (WC1999/088).</li> <li>At the time of reporting, there are no known impediments to obtaining a licence to operate in the area and the tenement is in good standing.</li> </ul>

Explanation dama by athen	
Exploration done by other parties	<ul> <li>7 open hole percussion drill holes completed by Geotechnics Australia Ltd (1972), not intersections of DSO grade mineralisation were reported, area determined to not be prospective.</li> <li>Rock chip sampling, geological mapping and geophysical surveys completed by Gondwana Resources Pty Ltd (2010), recognized presence of near surface zones of DSO grade iror mineralisation.</li> </ul>
Geology	The Corunna Downs Split Rock BIF-hosted iron ore resource is hosted by the ca. 3.02 Ga Cleaverville formation (Gorge Creek group, De Grey Supergroup). The prospect is located in the Kelly greenstone belt within the East Pilbara terrane of Western Australia, approximately 170km southwest of Port Hedland. The N-S trending Kelly greenstone belt is bound by the Corunna Downs and Shaw granitoid complexes. The Split Rock resource features successive macrobands of goethite-hematite rich, high grade (>55 wt% Fe) ore zones associated with neighbouring jaspilitic BIF units and banded chert and shale.
Drill hole Information	<ul> <li>Refer to Attachment 2 for information on all drillhole intercepts used in the resource estimation Also refer to Figure 3 which shows the drillhole collar plan and location of cross sections and Figure 4 which show 3 sections through the resource that were drilled as part of the infil program.</li> </ul>
Data aggregation methods	All reported assays have been length weighted; no top cuts have been applied. A nominal 53% Fe lower cut-off is applied with a maximum of 6m width of internal dilution and a 6m minimum intercept width. These criteria have been selected to most appropriately represent the mineralisation, taking into account overall deposit grade and geological continuity.
Relationship between mineralisation widths and intercept lengths	The attitude of the Split Rock resource is dominantly westerly dipping from 70-80 degrees and is drilled to grid east with drillholes inclined between -60 and -90 degrees which is slightly oblique to the orientation of the mineralisation. As such, due to the varying intersection angles all results are defined as down hole widths and not true widths of mineralisation.
Diagrams	<ul> <li>A plan view of the collar locations for the Split Rock resource can be seen in Figure 3.</li> <li>3 sections through the deposit with significant intercepts, stratigraphic and mineralisation interpretations can be seen in Figure 4.</li> </ul>
Balanced reporting	All Exploration drill hole results are reported in Attachment 2. Where results do not meet the criteria of significant interval these are reported in Attachment 2 as "no significant intercept".
Other substantive exploration data	<ul> <li>Estimate (see Atlas ASX release, Maiden Resource at Corunna Downs, 24 July 2013).</li> <li>Surface Geological mapping (stratigraphy, mineralisation and structure) of the Split Rock prospect was performed by Atlas Geological personnel and Digirock consultants.</li> <li>Routine multi-element analysis of potential deleterious or contaminating substances such as Arsenic, Lead, Zinc and Sulphur is completed for all samples.</li> <li>Geologists from the Centre for Exploration Targeting (CET), University of Western Australia (UWA) are completing research studies on the Corunna Downs Project with focus on the controls on mineralisation. The nature and timing of mineralisation events is also being evaluated through isotopic and geochemical analysis.</li> <li>Preliminary Metallurgical test work based on RC composite samples from a selection of holes has been performed by SGS Lakefield Oretest Pty Ltd. The aim of this test work was to determine preliminary characteristics of the deposit such as particle size distribution, abrasior index, bulk density, moisture and asbestiform mineral analysis.</li> </ul>
	<ul> <li>5 Geotechnical PQ3 diamond drill holes were recently completed to determine pit design parameters. All diamond core has been geotechnically logged and the holes scanned by televiewer. Results of this analysis are pending at the time of this release.</li> <li>4 of the HQ3 diamond hole sample bulk residues are to be used for bulk materials flow testing transportable moisture limit and dust extinction level tests. Additional diamond drilling is planned to provide more definitive metallurgical physical properties data such as Cwi, UCS, Ai, bulk density and moisture.</li> <li>Hydrogeology studies to determine dewatering requirements are currently being scoped.</li> <li>Waste classification samples have been collected to assess the nature of potentially acid forming (PAF) sulphidic carbonaceous shale material.</li> <li>A selection of drillholes will be left open for use in subterranean fauna studies.</li> <li>No further RC infill or extensional drilling is planned to be completed on Split Rock as the mineralisation is effectively closed off in all directions except for at depth in a few locations, but this is felt to be too deep and problematic to drill and would realistically be beyond the maximum depth limit of most optimal pits based on the lateral extents of the resource and ore body orientation.</li> <li>Work related to any potential mining development of the Split Rock deposit is dependent or</li> </ul>
	outcomes of scoping level mining studies.

SECTION 3 - Estimation and Reporting of Mineral Resources (Criteria listed in section 1, and where relevant in section 2, also apply to this section)

Database integrity	• All data is entered digitally in the field into acquire logging software on a Toughbook computer via								
	templates and lookup tables with enforced data validation rules. The data files are then								
	electronically transferred to the Perth office via email where they are loaded into the centralised								

Site visits	<ul> <li>SQL acQuire drillhole database and undergo further validation routines before being finally accepted. Validation reports are produced for each drillhole and sent back out to the site Geologists for final checking.</li> <li>Assay files sent electronically from the lab in a secure file format and also in hard copy reports. The assay data undergo numerous checks before being accepted into the database on passing all QAQC rules.</li> <li>The Atlas acQuire drillhole database is administered by a full-time Geological Database Administrator. Data validation checks are run routinely by the database administrator and database consultancy 'rOREdata' using acQuire software validation routines.</li> <li>The Competent Person for this report is a full time employee of Atlas Iron and undertakes regular</li> </ul>
	site visits ensuring that industry acceptable standards of the entire process from sampling through the final block model estimate are maintained. Site visits were carried out in June and October 2013 to inspect the deposit area, RC and diamond logging and sampling practices. Discussions were held with site personnel regarding procedures and a number of minor recommendations were made but nothing was noted that was of a material nature.
Geological interpretation	<ul> <li>There is good confidence in the geological interpretation of the mineral deposit and demonstrated good consistency both on section and between sections.</li> <li>The stratigraphical, structural and mineralisation interpretation has been based on a combination of geophysical, geochemical and lithological data obtained from drillholes plus surface mapping information.</li> <li>Wireframes of the stratigraphic and mineralisation surfaces are used to generate an empty geological block model.</li> <li>The overlying hardcap/hydrated zone displays higher variability and lower continuity and as such there is less confidence of the estimation of this zone.</li> <li>The mineralisation is noted to pinch down in a few isolated locations and lack continuity; there is less confidence in the estimation of these zones.</li> </ul>
Dimensions	<ul> <li>The Split Rock resource has dimensions of approximately 900m (N-S) along strike and 150m (E-W) across strike and extends from surface to a maximum depth of 230m, with an average depth of approximately 150m. A thin, 10-15m thick hydrated layer blankets the entire resource at surface. Thin bands (5-10m thick) of unmineralised to weakly mineralised jaspilite and shale are seen internal to the mineralisation and have been domained out where thick and continuous enough.</li> </ul>
Estimation and modelling techniques	<ul> <li>Mineralisation was domained according to stratigraphy and mineralisation style (hydrated or primary). Each geological unit was domained and estimated separately using hard boundaries. Drillhole sample data was flagged using domain codes generated from three dimensional stratigraphical and mineralisation surfaces.</li> <li>Interpretation does not extend mineralisation more than half drill hole spacing and surface mapping has been used to constrain the extents of mineralisation at surface.</li> <li>Univariate statistical analysis and variogram modeling completed with Snowden Supervisor software and used to define the spatial continuity of all elements within the mineralised domains.</li> <li>Quantitative kriging neighbourhood analysis (QKNA) undertaken to optimize estimation parameters, including search parameters, number of samples (minimum and maximum) and block discretization.</li> <li>No assumptions have been made regarding the modelling of selective mining units apart from the use of 5m parent cell heights to correspond with current mining bench heights used by Atlas at other projects.</li> <li>No assumptions regarding correlation between variables has been made, however it has been noted during statistical analysis that Fe and Phosphorous show some correlation and SiO<sub>2</sub> and Al<sub>2</sub>O<sub>3</sub> are correlated in most mineralised domains.</li> <li>Block model extends from 775880mE to 776680mE and 7622760mN to 7623960mN and elevation from 100mRL to 500mRL.</li> <li>A single block model to encompass the Split Rock Mineral Resource was constructed using a 20mN by 20mE by 5mRL parent block size with sub-celling to ensure the mineralisation is well represented by the blocks and appropriate sample support is maintained.</li> <li>The block model has been assigned unique mineralisation codes that correspond with the geological domain as defined by the wireframes. These domains are used to control the resource estimates.</li> <li>Ordinary Kriging was used to estimate the standar</li></ul>
	<ul> <li>Search directions and ranges determined from variogram modelling were used to constrain the block interpolation. Estimation search strategies have sought to ensure robust estimates whilst minimising conditional bias.</li> <li>Three search estimation runs are used with initial short search runs. The search ellipses typically cover 2 drill spacing's for run 1, 3 drill spacing's for run 2 and 4 drill spacing's for run 3.</li> <li>A minimum of 12 samples and a maximum of 30 samples are required for an estimate in run 1, the minimum number of samples reducing to 10 for run 2 and 8 for run 3. A maximum of 4</li> </ul>

	<ul> <li>samples from any one drill hole is allowed per estimate.</li> <li>A block discretisation of 5, 5, 2 was applied to align with the parent cell block size.</li> </ul>
	<ul> <li>A block discretisation of 5, 5, 2 was applied to align with the parent cell block size.</li> <li>Generally a high proportion of blocks (&gt;90%) were estimated in run 1.</li> </ul>
	• Grade restriction search routines were applied to some of the minor deleterious elements in
	<ul> <li>some domains to limit the influence of extreme/outlier grades from smearing distant blocks.</li> <li>All block estimates are based on interpolation into parent block volumes.</li> </ul>
	<ul> <li>Mineral resource estimate does not include any form of dilution, apart from where small intervals</li> </ul>
	of internal waste could not be adequately domained out.
	<ul> <li>Maptek Vulcan software was used to complete the block estimation.</li> <li>Standard model and estimation validation has been completed using visual and numerical</li> </ul>
	methods and formal peer review by appropriately qualified internal staff.
	• Kriging efficiency and slope of regression statistics were used to quantify the estimation results
	<ul> <li>were to the desired level of quality.</li> <li>Block model validation methods used were visual checks comparing composite grades to block</li> </ul>
	grades, global statistical comparisons for each domain, swath plot comparisons produced along
	easting's, northings and elevations and a change of support analysis was completed.
	<ul> <li>This resource estimate was compared to the previous estimate completed in July 2013 to understand changes between the models due to the infill drilling. The two models compared well</li> </ul>
	with the updated estimate reporting similar volume, tones and grade, demonstrating the robust
$\mathcal{O}$	nature of the resource.
Moisture	Tonnages are estimated on a dry basis.
	• The water table sits approximately 60m below the ground surface; approximately 40% of the resource is located below water table.
))	
Cut-off parameters	• The criteria used for domaining mineralised material is >50% Fe, which appears to be a natural grade boundary for this deposit between mineralised and unmineralised BIF.
D)	<ul> <li>Based on the current Atlas shipped product grade specification, a 50% Fe lower cut-off grade is</li> </ul>
	deemed a suitable cut-off to report resources for Split Rock.
Mining factors or	• Mining is assumed to be similar to the process used at other nearby Atlas deposits by open pit
Jassumptions	using conventional backhoe excavator methods with ore being mined in 5m benches on 2.5m flitches.
	No other assumptions on mining methodology have been assumed at this stage as no detailed
	mine planning or production scenarios have been reviewed and are subject to a scoping level
9	<ul> <li>study.</li> <li>It is a reasonable assumption that this resource will eventually be economically extracted based</li> </ul>
D)	on its proximal location to existing Atlas projects and infrastructure and also due to its favourable
P	size and grade characteristics which will fit the Atlas product specification.
Metallurgical factors or assumptions	Preliminary Metallurgical test work based on RC composite samples from a selection of holes     have been performed by SCS I alofield Oretest Dry Itd. The sim of this test work was te
	has been performed by SGS Lakefield Oretest Pty Ltd. The aim of this test work was to determine preliminary characteristics of the deposit such as particle size distribution, abrasion
	index, bulk density, moisture and asbestiform mineral analysis.
Environmental factors or	• A thick (20-30m) carbonaceous and sulphidic (pyrite) shale unit has been identified along the
assumptions	entire footwall position of the deposit below the depth of oxidation. The net acid producing
))	potential of this shale has not been determined to date, however samples have been collected and the test work is anticipated to commence shortly by Graeme Campbell and Associates.
	• The volume of this sulphidic shale within any potential pit is expected to be comfortably
	encapsulated by inert waste within any waste dump volume based on high level studies completed by Atlas. Mitigation of acid drainage within the pit will need further analysis.
	Other detailed waste characterisation studies have not been undertaken but are anticipated to be
	completed during 2014.
Bulk density	• Dry bulk density has been estimated into the model with the use of geophysical density
))	measurements collected in RC holes and regressed back to dry core dimensional density measurements.
	• All RC holes are attempted to be downhole surveyed for gamma density however some holes
	were open to end of hole depth resulting in incomplete data coverage over the deposit. Not all core intervals had 100% complete core recovery and these density measurements were
	excluded from the regression analysis as they are not representative.
	Geophysical density measures the in-situ density inclusive of moisture and porosity. Filtered and     algoridate the medal in
))	cleaned Geophysical density was composited to 2m length and then estimated into the model in a similar fashion to grades and then a regression has been applied to account for the moisture,
	porosity and hole rugosity present in the readings to derive a dry density.
	<ul> <li>The regression has been calculated by comparing geophysical measurements in a diamond hole with dry, diamond core dimensional density measurements over the same intervals. Geophysical</li> </ul>
	measurements taken in RC and Diamond Twin holes are also directly compared to account for
	differences due to hole effect (rugosity).
	<ul> <li>The use of dimensional tray density techniques is generally believed to be unbiased as it accounts for all material types and avoids material handling and selectivity issues commonly</li> </ul>
	encountered by using more traditional Archimedes style density measurements.
	<ul> <li>1,007 tray dimensional density measurements were determined from 5 HQ3 diamond holes (1,187m core) for the analysis.</li> </ul>
	<ul> <li>A density regression of 4.7% reduction to geophysical density to derive the dry bulk density has</li> </ul>

	<ul> <li>been applied globally to this resource.</li> <li>The resulting dry bulk density of 2.76t/m<sup>3</sup> for the mineralisation compares consistently with Atlas's other nearby deposits such as Abydos and is felt to be a realistic determination of the density.</li> <li>This is a bulk commodity project.</li> </ul>
Classification	<ul> <li>Mineral resources have been classified by the Competent Person into the Inferred and Indicated categories based on RC drillhole spacing (40m x 40m), geological interpretation confidence, diamond core vs RC comparison, QAQC and overall data quality and confidence, grade continuity and resultant estimation statistical quality.</li> <li>Mineral resource classification has appropriately taken into account the data spacing, distribution continuity, reliability, quality and quantity of data.</li> <li>The input data is comprehensive in its coverage of the mineralisation and does not misrepresen in-situ mineralisation.</li> <li>The definition of mineralised zones is based on a high level of geological understanding producing a robust model of mineralised domains.</li> <li>The results of the validation of the block model show good correlation of the input data to the estimated grades.</li> <li>The geological model and mineral resource estimation appropriately reflect the Competen Persons view of the deposit and appropriate account has been taken of all relevant factors.</li> <li>All near surface hydrated mineralisation has been given an Inferred classification due to limited RC drilling coverage, sparse geophysical density measurements and generally wet drilling conditions. Where the mineralisation pinches down and lacks continuity and shows increased complexity has also been given an Inferred classification.</li> <li>An Indicated classification has been applied to areas of consistent RC drilling density, sufficient coverage of geophysical and core density data, confidence in QAQC of input data, strong geological and mineralisation continuity, mostly above water table (above 150m depth) or where RC drilling has been kept relatively dry and have confidence simation results.</li> <li>The results of this updated resource compare well with the previous Split Rock resource estimate and show consistency of grade and tonnages.</li> </ul>
Audits or reviews	<ul> <li>Atlas have undertaken an internal review of the mineral resource estimate and is satisfied the estimation is valid and of sufficient confidence to support an Indicated/Inferred classification.</li> <li>The review consisted of numerous checks made throughout the data collection and estimation process. A final peer review including visual checks of blocks versus drillhole grades, global means comparisons, histogram distribution comparisons, total assay closure checks, swath plots in Easting, Northing and elevation and a change of support analysis was completed.</li> <li>This mineral resource has not been audited externally.</li> <li>Internal peer reviews are conducted throughout the estimation process and on completion by the Competent Person.</li> </ul>
Discussion of relative accuracy/ confidence	<ul> <li>The confidence in this resource estimate has been deemed appropriate as a basis for long term planning and mine design and is not necessarily sufficient for shorter term planning and scheduling.</li> <li>A change of support analysis was undertaken to assess the sensitivity to the grade-tonnage curve in going from sample to block sized support at a range of cut-off grades. This analysis shows that some misclassification of material around the specified cut-off grades can be expected and is attributed to an expected amount of smoothing incurred by the ordinary kriging process.</li> <li>The Split Rock Resource Estimate is sufficient for scoping level study purposes commensurate with the classification of the resource.</li> <li>This statement relates to global estimates of tonnes and grade.</li> <li>There has been no production from the Split Rock deposit to provide comparison of relative</li> </ul>

## ATTACHMENT 4 – CORUNNA DOWNS SPLIT ROCK RESOURCE GRADE TONNAGE INFORMATION

Fe Cut-	Tonnes	Fe%	SiO2%	AI2O3%	<b>P%</b>	LOI%	S%	MnO%
Off								
50	25,440,000	57.1	6.6	1.5	0.12	9.0	0.01	0.41
51	24,900,000	57.2	6.4	1.5	0.12	9.0	0.01	0.41
52	24,310,000	57.4	6.3	1.5	0.12	9.0	0.01	0.41
53	23,350,000	57.6	6.1	1.4	0.12	8.9	0.01	0.41
54	21,980,000	57.8	5.8	1.4	0.12	8.9	0.01	0.41
55	20,220,000	58.1	5.6	1.3	0.12	8.9	0.01	0.41
56	17,690,000	58.5	5.3	1.2	0.12	8.8	0.01	0.42
57	14,550,000	58.9	4.9	1.2	0.12	8.7	0.01	0.41
58	10,640,000	59.4	4.5	1.1	0.12	8.5	0.01	0.40
59	6,200,000	60.1	4.2	1.0	0.12	8.1	0.01	0.35
60	2,870,000	60.8	3.9	0.9	0.12	7.7	0.01	0.30
61	930,000	61.6	3.6	0.9	0.12	7.1	0.01	0.25
62	190,000	62.5	3.5	0.8	0.12	6.4	0.00	0.22
63	20,000	63.4	4.2	0.4	0.09	5.3	0.00	0.09
64	160	64.7	3.4	0.3	0.08	4.6	0.00	0.03

Grade Tonnage curve for Corunna Downs Split Rock Indicated &Inferred Resource at various Fe cut-offs.

