

COARSE SPODUMENE CONCENTRATE PRODUCED AT SEYMOUR WITH LITHIUM RECOVERY EXCEEDING 72%

HIGHLIGHTS

- **Flagship Seymour concentrate recovery exceeds 72% using heavy liquid separation**
- **Confirms simple 2 stage DMS flowsheet potential**
- **Critical milestone for the PEA and future off-take partners**
- **Test work based on 1000kg of representative composites as per preliminary mine plan**
- **Very high recovery due to large spodumene crystal size**
- **Concentrates now being sent for conversion work to Lithium Hydroxide**

Green Technology Metals Limited (**ASX: GT1**) (**GT1** or the **Company**), a Canadian-focused multi-asset lithium business, is pleased to announce very high spodumene concentrate recoveries from its 100%-owned Seymour Project, located approximately 250km north of Thunder Bay in Ontario, Canada. The spodumene concentrate represents a critical milestone for the business and future off-take partners.



Figure 1: DMS test work being undertaken at SGS Canada's facilities in Lakefield, Ontario

"The next few months will be an exciting time for the team at GT1 as we focus on increasing the value of our Flagship Seymour Project through metallurgical test work. These results will culminate part of our Preliminary Economic Assessment (PEA) and marks another important step closer to building a vertically integrated lithium business in Ontario, Canada."

- GT1 Chief Executive Officer, Luke Cox



Four composite samples totalling approximately 1000kg of recent and historic diamond core, derived from the Seymour pegmatite was selected, and shipped for testing. The samples represent the different mineralisation phases seen in the pegmatite within, what GT1 expect to be, the final open pit mine shell at Seymour.

Heavy liquid separation (HLS) test work was undertaken at SGS Canada's facilities in Lakefield, Ontario. SGS compiled a master composite, based on the four samples selected, to provide a preliminary indication of the lithium beneficiation performance by utilising dense media separation (DMS).

Table 1 presents a summary table of the 8 mm HLS test results to achieve an HLS concentrate grade of 6.0% Li₂O and 1.2 % Fe₂O₃. The resulting global HLS recovery, post magnetic separation including losses to HLS fines bypass, is 72%. This recovery has been reduced to align the laboratory magnetic separator with the industrial-scale equipment. The HLS result shows a strong potential to use DMS as the primary recovery method. Further work will continue to be completed for an optimised flowsheet that will ensure both maximised lithium unit recovery, and mass yield will be achieved for the Seymour deposit.

HLS on master composite - crush size 8 mm			Grade		Distribution	
Description	HLS SG	Mass Distribution	% Li ₂ O	% Fe ₂ O ₃	Li ₂ O	Fe ₂ O ₃
HLS non-mag ¹ product	2.82	12.7	6.0	1.2	72.6	19.2
HLS mag ¹ product	2.82	3.6	1.3	9.1	4.5	34.7
HLS tailings	-2.65	46.7	0.0	0.2	2.0	9.3
Flotation feed (fines and middlings)	-	37.0	0.6	0.9	20.9	36.9
Total feed	-	100	1.1	0.9	100	100

¹ Dry magnetic separator used for HLS testwork

Table 1: HLS and assay test work on 8mm feed (12mm & 10mm samples being QAQC)

The Seymour bulk sample permit has been approved by the Ontario Mines Department (MNDM) and sampling is planned to begin in January 2023 where we will be extracting approximately ninety-nine tonnes from the North Aubry pegmatite outcrop within the Seymour project. The spodumene concentrate produced through the pilot work will be used as feed stock for the Lithium Hydroxide Conversion Program (LHCP), all forming part of the ongoing feasibility study.

This ASX release has been approved for release by the Board.

KEY CONTACTS

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Green Technology Metals (ASX:GT1)

GT1 is a North American focussed lithium exploration and development business. The Company's 100% owned Ontario Lithium Projects comprise high-grade, hard rock spodumene assets (Seymour, Root and Wisa) and lithium exploration claims (Allison and Solstice) located on highly prospective Archean Greenstone tenure in north-west Ontario, Canada.

All sites are proximate to excellent existing infrastructure (including hydro power generation and transmission facilities), readily accessible by road, and with nearby rail delivering transport optionality.

Seymour has an existing Mineral Resource estimate of 9.9 Mt @ 1.04% Li₂O (comprised of 5.2 Mt at 1.29% Li₂O Indicated and 4.7 Mt at 0.76% Li₂O Inferred).¹ Accelerated, targeted exploration across all three projects delivers outstanding potential to grow resources rapidly and substantially.



¹ For full details of the Seymour Mineral Resource estimate, see GT1 ASX release dated 23 June 2022, *Interim Seymour Mineral Resource Doubles to 9.9Mt*. The Company confirms that it is not aware of any new information or data that materially affects the information in that release and that the material assumptions and technical parameters underpinning this estimate continue to apply and have not materially changed.

APPENDIX A: IMPORTANT NOTICES

Competent Person's Statements

Information in this report relating to Exploration Results is based on information reviewed by Mr Luke Cox (Fellow AusIMM). Mr Cox has sufficient experience which is relevant to the style of mineralisation and type of deposit under consideration and to the activity which he is undertaking to qualify as a Competent Person as defined by the 2012 Edition of the Australasian Code for reporting of Exploration Results, Mineral Resources and Ore Reserves. Mr Cox consents to the inclusion of the data in the form and context in which it appears in this release. Mr Cox is the Chief Executive Officer of the Company and holds securities in the Company.

No new information

Except where explicitly stated, this announcement contains references to prior exploration results, all of which have been cross-referenced to previous market announcements made by the Company. The Company confirms that it is not aware of any new information or data that materially affects the information included in the relevant market announcements.

The information in this report relating to the Mineral Resource estimate for the Seymour Project is extracted from the Company's ASX announcement dated 23 June 2022. GT1 confirms that it is not aware of any new information or data that materially affects the information included in the original announcement and that all material assumptions and technical parameters underpinning the Mineral Resource estimate continue to apply.

Forward Looking Statements

Certain information in this document refers to the intentions of Green Technology Metals Limited (ASX: GT1), however these are not intended to be forecasts, forward looking statements or statements about the future matters for the purposes of the Corporations Act or any other applicable law. Statements regarding plans with respect to GT1's projects are forward looking statements and can generally be identified by the use of words such as 'project', 'foresee', 'plan', 'expect', 'aim', 'intend', 'anticipate', 'believe', 'estimate', 'may', 'should', 'will' or similar expressions. There can be no assurance that the GT1's plans for its projects will proceed as expected and there can be no assurance of future events which are subject to risk, uncertainties and other actions that may cause GT1's actual results, performance or achievements to differ from those referred to in this document. While the information contained in this document has been prepared in good faith, there can be given no assurance or guarantee that the occurrence of these events referred to in the document will occur as contemplated. Accordingly, to the maximum extent permitted by law, GT1 and any of its affiliates and their directors, officers, employees, agents and advisors disclaim any liability whether direct or indirect, express or limited, contractual, tortious, statutory or otherwise, in respect of, the accuracy, reliability or completeness of the information in this document, or likelihood of fulfilment of any forward-looking statement or any event or results expressed or implied in any forward-looking statement; and do not make any representation or warranty, express or implied, as to the accuracy, reliability or completeness of the information in this document, or likelihood of fulfilment of any forward-looking statement or any event or results expressed or implied in any forward-looking statement; and disclaim all responsibility and liability for these forward-looking statements (including, without limitation, liability for negligence).

Appendix A JORC Code, 2012 Edition – Table 1 report template

Criteria	JORC Code explanation	Commentary																																																																																																																																									
Sampling techniques	<ul style="list-style-type: none"> Nature and quality of sampling (eg cut channels, random chips, or specific specialised industry standard measurement tools appropriate to the minerals under investigation, such as down hole gamma sondes, or handheld XRF instruments, etc). These examples should not be taken as limiting the broad meaning of sampling. Include reference to measures taken to ensure sample representivity and the appropriate calibration of any measurement tools or systems used. Aspects of the determination of mineralisation that are Material to the Public Report. In cases where 'industry standard' work has been done this would be relatively simple (eg 'reverse circulation drilling was used to obtain 1 m samples from which 3 kg was pulverised to produce a 30 g charge for fire assay'). In other cases more explanation may be required, such as where there is coarse gold that has inherent sampling problems. Unusual commodities or mineralisation 	<p>Diamond Drilling</p> <ul style="list-style-type: none"> Available drill holes data were accumulated from multiple phases of drilling conducted by a number of operators from 2002 to the present. Diamond drilling was used to obtain nominally 1m downhole samples of core. Core samples were ½ cored using a diamond saw with ½ the core placed in numbered sample bags for assaying and the other half retained in sequence in the core tray. ½ core samples were approximately 2.5kg in weight with a minimum weight of 500grams. Core was cut down the apex of the core and the same downhole side of the core selected for assaying to reduce potential sampling bias. <table border="1"> <caption>Drilling Used in Mineral Resource</caption> <thead> <tr> <th rowspan="2">Year</th> <th colspan="4">Holes</th> <th colspan="4">Metres</th> <th colspan="4">All Companies</th> </tr> <tr> <th>Linear</th> <th>Ardide</th> <th>Green</th> <th>Total</th> <th>Linear</th> <th>Ardiden</th> <th>Green Tech</th> <th>Total</th> <th>BTW</th> <th>NQ</th> <th>HQ</th> <th>NR</th> </tr> </thead> <tbody> <tr> <td>2002</td> <td>30</td> <td>0</td> <td>0</td> <td>30</td> <td>1,677.45</td> <td>-</td> <td>-</td> <td>-</td> <td>-</td> <td>-</td> <td>1,677.45</td> <td>-</td> <td>-</td> </tr> <tr> <td>2009</td> <td>12</td> <td>0</td> <td>0</td> <td>12</td> <td>1,573.50</td> <td>-</td> <td>-</td> <td>1,573.50</td> <td>-</td> <td>-</td> <td>-</td> <td>-</td> <td>1,573.50</td> </tr> <tr> <td>2016</td> <td>0</td> <td>29</td> <td>0</td> <td>29</td> <td>-</td> <td>1,950.00</td> <td>-</td> <td>1,950.00</td> <td>-</td> <td>-</td> <td>-</td> <td>-</td> <td>1,950.00</td> </tr> <tr> <td>2017</td> <td>0</td> <td>69</td> <td>0</td> <td>69</td> <td>-</td> <td>7,864.29</td> <td>-</td> <td>7,864.29</td> <td>-</td> <td>-</td> <td>-</td> <td>-</td> <td>7,097.00</td> </tr> <tr> <td>2018</td> <td>0</td> <td>37</td> <td>0</td> <td>37</td> <td>-</td> <td>6,564.71</td> <td>-</td> <td>6,564.71</td> <td>4,659.71</td> <td>-</td> <td>-</td> <td>-</td> <td>1,905.00</td> </tr> <tr> <td>2021</td> <td>0</td> <td>0</td> <td>1</td> <td>1</td> <td>-</td> <td>-</td> <td>341.00</td> <td>341.00</td> <td>-</td> <td>341.00</td> <td>-</td> <td>-</td> <td>-</td> </tr> <tr> <td>2022</td> <td>0</td> <td>0</td> <td>21</td> <td>21</td> <td>-</td> <td>-</td> <td>7,950.69</td> <td>7,950.69</td> <td>-</td> <td>7,219.69</td> <td>-</td> <td>-</td> <td>731.00</td> </tr> <tr> <td>Grand</td> <td>42</td> <td>135</td> <td>22</td> <td>199</td> <td>3,250.95</td> <td>16,379.00</td> <td>8,291.69</td> <td>26,244.19</td> <td>4,659.71</td> <td>9,238.14</td> <td>2,636.00</td> <td>-</td> <td>10,620.50</td> </tr> </tbody> </table> <p>18 holes were excluded for the MRE Excluded holes were largely from 2009 and 2002 where assaying and/or geological data was missing from the record or the holes were re-drills.</p> <p>Historic Grab Samples</p> <ul style="list-style-type: none"> Grab samples were not used in the MRE <p>Historic Channel Samples</p> <ul style="list-style-type: none"> Preparation prior to obtaining the channel samples including grid and geo-references and marking of the pegmatite structures. Samples were cut across the pegmatite with a diamond saw perpendicular to strike. Average 1 metre samples are obtained, logged, removed and bagged and secured in accordance with QAQC procedures. 	Year	Holes				Metres				All Companies				Linear	Ardide	Green	Total	Linear	Ardiden	Green Tech	Total	BTW	NQ	HQ	NR	2002	30	0	0	30	1,677.45	-	-	-	-	-	1,677.45	-	-	2009	12	0	0	12	1,573.50	-	-	1,573.50	-	-	-	-	1,573.50	2016	0	29	0	29	-	1,950.00	-	1,950.00	-	-	-	-	1,950.00	2017	0	69	0	69	-	7,864.29	-	7,864.29	-	-	-	-	7,097.00	2018	0	37	0	37	-	6,564.71	-	6,564.71	4,659.71	-	-	-	1,905.00	2021	0	0	1	1	-	-	341.00	341.00	-	341.00	-	-	-	2022	0	0	21	21	-	-	7,950.69	7,950.69	-	7,219.69	-	-	731.00	Grand	42	135	22	199	3,250.95	16,379.00	8,291.69	26,244.19	4,659.71	9,238.14	2,636.00	-	10,620.50
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	<p>types (eg submarine nodules) may warrant disclosure of detailed information.</p>	<ul style="list-style-type: none"> • Sampling continued past the Spodumene -Pegmatite zone, even if it is truncated by Mafic Volcanic a later intrusion. • Samples were then transported directly to the laboratory for analysis accompanied with the log and instruction forms. • Bagging of the samples was supervised by a geologist to ensure there are no numbering mix-ups. • One tag from a triple tag book was inserted in the sample bag. <p>As recorded, procedures were consistent with normal industry practices</p> <p>Channel samples were used to aid the pegmatite interpretation but were not used in the estimate.</p>
Drilling techniques	<ul style="list-style-type: none"> • Drill type (eg core, reverse circulation, open-hole hammer, rotary air blast, auger, Bangka, sonic, etc) and details (eg core diameter, triple or standard tube, depth of diamond tails, face-sampling bit or other type, whether core is oriented and if so, by what method, etc). 	<ul style="list-style-type: none"> • Tri-cone drilling was undertaken through the thin overburden prior to NQ2 or BTW diamond drilling through the primary rock. 11 holes were drilled by Ardiden using HQ core. • 199 diamond core samples were used in the Mineral Resource estimate for 26,244.19 metres including 22 holes drilled by GT1 for 8,291.69m. • 18 holes were rejected from the estimate mainly from 2009 and 2002 due to missing lithology logging and assay data or re-drills or poor orientation to the pegmatite attitude. Some of the earlier North Aubry holes were drilled vertically until it was released the pegmatite strike 045. The vast majority of holes were drilled to the southwest approximately perpendicular to the pegmatite orientation.
Drill sample recovery	<ul style="list-style-type: none"> • Method of recording and assessing core and chip sample recoveries and results assessed. • Measures taken to maximise sample recovery and ensure representative nature of the samples. • Whether a relationship exists between sample recovery and grade and whether sample bias may have occurred due to preferential loss/gain of fine/coarse material. 	<ul style="list-style-type: none"> • No core was recovered through the overburden tri-coned section of the hole (top 5m of the hole) • Core recovery through the primary rock and mineralised pegmatite zones was over 95% and considered satisfactory. • Recovery was determined by measuring the recovered metres in the core trays against the drillers core block depths for each run.
Logging	<ul style="list-style-type: none"> • Whether core and chip samples have been geologically and geotechnically logged to a level of detail to support appropriate Mineral Resource estimation, mining studies and metallurgical studies. • Whether logging is qualitative or quantitative in nature. Core (or costean, channel, etc) photography. • The total length and percentage of the relevant intersections logged. 	<ul style="list-style-type: none"> • Each sample was logged for lithology, minerals, grain size and texture as well as alteration, sulphide content, and any structures. • Logging is qualitative in nature. • Samples are representative of an interval or length. • Sampling was undertaken for the entire cross strike length of the intersected pegmatite unit at nominal 1m intervals with breaks at geological contacts. Sampling extended into the country mafic rock. • Logging is qualitative in nature based on visual estimates of mineral species and geological features.

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Sub-sampling techniques and sample preparation	<ul style="list-style-type: none"> If core, whether cut or sawn and whether quarter, half or all core taken. If non-core, whether riffled, tube sampled, rotary split, etc and whether sampled wet or dry. For all sample types, the nature, quality and appropriateness of the sample preparation technique. Quality control procedures adopted for all sub-sampling stages to maximise representivity of samples. Measures taken to ensure that the sampling is representative of the in situ material collected, including for instance results for field duplicate/second-half sampling. Whether sample sizes are appropriate to the grain size of the material being sampled. 	<ul style="list-style-type: none"> The bulk of the core is NQ diameter core with some BTK and HQ core drilled by Linear and Ardiden. All recent drilling has been NQ diameter core Each ½ core sample was dried, crushed to entirety to 90% -10 mesh, riffle split (up to 5 kg) and then pulverized with hardened steel (250 g sample to 95% -150 mesh)(includes cleaner sand). Blanks and Certified Reference samples were inserted in each batch submitted to the laboratory at a rate of approximately 1:20. Ardiden field duplicates were taken at a rate of 1:20 taken immediately adjacent to the original sample. The sample preparation process is considered representative of the whole core sample. 																																				
Quality of assay data and laboratory tests	<ul style="list-style-type: none"> The nature, quality and appropriateness of the assaying and laboratory procedures used and whether the technique is considered partial or total. For geophysical tools, spectrometers, handheld XRF instruments, etc, the parameters used in determining the analysis including instrument make and model, reading times, calibrations factors applied and their derivation, etc. Nature of quality control procedures adopted (eg standards, blanks, duplicates, external laboratory checks) and whether acceptable levels of accuracy (ie lack of bias) and precision have been established. 	<ul style="list-style-type: none"> Prior to 2016 little QAQC was performed other than some duplicate core sampling and verification laboratory internal standards. Whilst the results appear acceptable the lack of QAQC was a concern. A spatial sampling pairing review was undertaken comparing Ardiden and Linear samples located within 8m of each other within the pegmatite domains. The results were inconclusive but hinted at the Linear Li₂O results being biased slightly lower than Ardiden's results. It is unclear as to why this would be the case: <table border="1" data-bbox="709 948 1545 1101"> <thead> <tr> <th>Company</th> <th>Field Name</th> <th>Minimum</th> <th>Maximum</th> <th>No of Points</th> <th>Mean</th> <th>Variance</th> <th>Std Dev</th> <th>Coeff. of Variation</th> </tr> </thead> <tbody> <tr> <td>Ardiden</td> <td>Li2O_ppm</td> <td>105.4</td> <td>53609.7</td> <td>200</td> <td>14,776</td> <td>1.46E+08</td> <td>12085.76</td> <td>0.818</td> </tr> <tr> <td>Linear</td> <td>Li2O_ppm</td> <td>53.81</td> <td>59640</td> <td>200</td> <td>12,483</td> <td>1.64E+08</td> <td>12819.8</td> <td>1.027</td> </tr> <tr> <td colspan="3">Difference</td> <td>-11%</td> <td colspan="5"></td> </tr> </tbody> </table> <p>As the Linear drilling makes up only 12% of the meterage included in the mineral resource the bias is not considered material to the estimate.</p> <ul style="list-style-type: none"> In 2016 Ardiden employed a single Li₂O standard (CGL 128) certified by the Mongolian Central Geological Laboratory derived from the wolfram-lithium deposit located in the Arbayan area, Sukhbaatar province of Mongolia in April 2012. Ardiden used the standard from 2016 to 2018 until it was superseded by more reliable OREAS standards. The control charts produced over this time period for CGL 128 suggest occasional poor precision and a cluster of low grade assay returns. However, the OREAS standards, overlapping some of 2018 show no obvious bias and better precision from AGAT Laboratories. All the Ardiden drill samples were analysed by AGAT Laboratories who are accredited by The Standards Council of Canada (SCC), The Canadian Association for Laboratory Accreditation (CALA), SAI Global and have ISO/IEC 17025:2005 and ISO 9001:2015 accreditation. All Ardiden samples were analysed by AGAT for lithium and a suite of other elements, using Sodium Peroxide Fusion - ICP-OES/ICP-MS Finish (method# 201-378). Sodium Peroxide Fusion oxidizes samples at high temperatures effectively in 	Company	Field Name	Minimum	Maximum	No of Points	Mean	Variance	Std Dev	Coeff. of Variation	Ardiden	Li2O_ppm	105.4	53609.7	200	14,776	1.46E+08	12085.76	0.818	Linear	Li2O_ppm	53.81	59640	200	12,483	1.64E+08	12819.8	1.027	Difference			-11%					
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		<p>dissolving all the pegmatite minerals while the ICP-MS ionizes chemical species and sorts the ions based on their mass-to-charge ratio.</p> <div style="border: 1px solid black; padding: 5px;"> <p>Control Chart for Mean and Range for CGL 128 - Li_ppm</p> <p>Project Seymour CGL 128 From 01-Jan-16 To 31-Dec-16 Element Li_ppm</p> <p>Quality Characteristic Raw sample Standard Deviation vs Certified Tolerances Sample Size, n 93 k 3</p>  <table border="1" style="width: 100%; border-collapse: collapse;"> <thead> <tr> <th colspan="2">Statistics from Raw Data Table</th> <th>Certified Values</th> </tr> </thead> <tbody> <tr> <td>R-bar</td> <td>2.725</td> <td>See Below</td> </tr> <tr> <td>Process Mean, μ-hat</td> <td>2.725</td> <td></td> </tr> <tr> <td>Process St.Dev., σ-hat</td> <td>151</td> <td></td> </tr> <tr> <td>σ_{sbar}</td> <td>16</td> <td></td> </tr> </tbody> </table> <table border="1" style="width: 100%; border-collapse: collapse; margin-top: 10px;"> <thead> <tr> <th colspan="2">Control Limits for X-bar Chart</th> <th colspan="2">Control Limits for R Chart</th> </tr> </thead> <tbody> <tr> <td>CL_{xbar}</td> <td>2725</td> <td>CL_R</td> <td>2.685</td> </tr> <tr> <td>UCL_{xbar}</td> <td>2.772 CL+Kσ_{sbar}</td> <td>UCL_R</td> <td>2.894</td> </tr> <tr> <td>LCL_{xbar}</td> <td>2.678 CL-Kσ_{sbar}</td> <td>LCL_R</td> <td>2.476</td> </tr> <tr> <td>α</td> <td>0.0027</td> <td></td> <td></td> </tr> <tr> <td>ARL</td> <td>370.4 samples</td> <td></td> <td></td> </tr> </tbody> </table> </div> <p>charge ratio.</p> <ul style="list-style-type: none"> All GT1 drill samples were submitted to Actlabs Thunder Bay for analysis for sample preparation before forwarding the pulps to their Ancaster laboratory in Ontario Canada for analysis using Sodium Peroxide Fusion - ICP-OES/ICP-MS Finish. GT1 inserted certified lithium standards of varying grade and blanks into each batch submitted to Actlabs to monitor precision and bias performance at a rate of 1:20. Actlabs also inserted internal standards, blanks and pulp duplicates within each sample batch as part of their own internal monitoring of quality control. All GT1 results were within acceptable tolerances. No significant bias or precision issues were observed in the control samples. 	Statistics from Raw Data Table		Certified Values	R-bar	2.725	See Below	Process Mean, μ -hat	2.725		Process St.Dev., σ -hat	151		σ_{sbar}	16		Control Limits for X-bar Chart		Control Limits for R Chart		CL _{xbar}	2725	CL _R	2.685	UCL _{xbar}	2.772 CL+K σ_{sbar}	UCL _R	2.894	LCL _{xbar}	2.678 CL-K σ_{sbar}	LCL _R	2.476	α	0.0027			ARL	370.4 samples		
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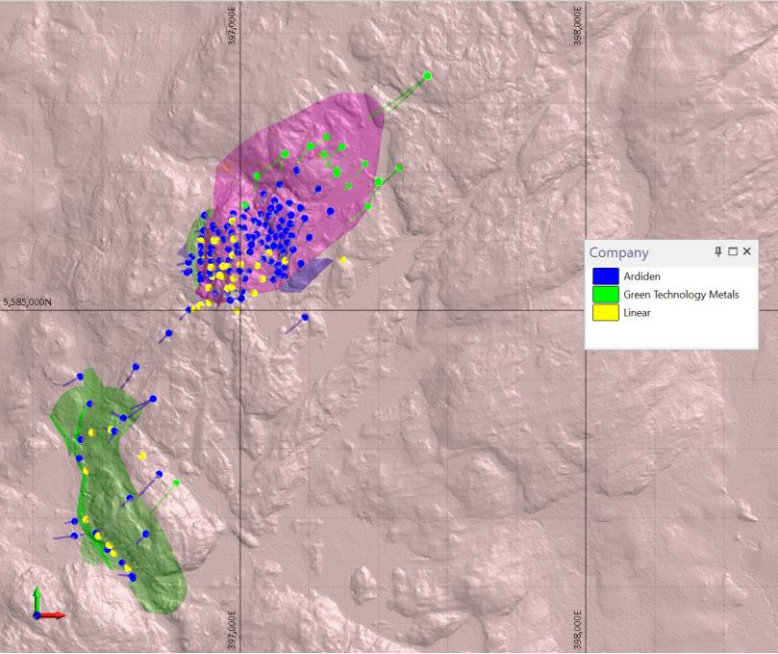
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		<div style="border: 1px solid #00a651; padding: 5px;"> <h3 style="margin: 0;">Control Chart for Mean and Range for OREAS 751 - Li_ppm</h3> <p>Project Seymour OREAS 751 From 01-Jan-16 To 31-Dec-22 Element Li_ppm</p> <p>Quality Characteristic <input type="text" value="Raw sample Standard Deviation vs Certified Tolerances"/> Sample Size, n 53 k <input type="text" value="3"/></p> <div style="display: flex; justify-content: space-around;"> <div style="width: 45%;"> <h4 style="margin: 0;">Control Chart for Mean and Range for OREAS 751 - Li_ppm</h4> </div> <div style="width: 45%;"> <h4 style="margin: 0;">Certified Control Limits</h4> </div> </div> <div style="display: flex; justify-content: space-between; margin-top: 10px;"> <table border="1" style="width: 45%; border-collapse: collapse;"> <thead> <tr> <th colspan="2" style="background-color: #e6f2ff;">Statistics from Raw Data Table</th> </tr> </thead> <tbody> <tr><td>R-bar</td><td>4,671</td></tr> <tr><td>Process Mean, μ-hat</td><td>4,671</td></tr> <tr><td>Process St.Dev., σ-hat</td><td>155</td></tr> <tr><td>σ_{xbar}</td><td>16</td></tr> </tbody> </table> <table border="1" style="width: 45%; border-collapse: collapse;"> <thead> <tr> <th colspan="2" style="background-color: #e6f2ff;">Certified Values</th> </tr> </thead> <tbody> <tr><td colspan="2">See Below</td></tr> </tbody> </table> </div> <div style="display: flex; justify-content: space-between; margin-top: 10px;"> <table border="1" style="width: 45%; border-collapse: collapse;"> <thead> <tr> <th colspan="2" style="background-color: #00a651; color: white;">Control Limits for X-bar Chart</th> </tr> </thead> <tbody> <tr><td>CL_{xbar}</td><td>4671</td></tr> <tr><td>UCL_{xbar}</td><td>4,720 CL+kσ_{xbar}</td></tr> <tr><td>LCL_{xbar}</td><td>4,623 CL-kσ_{xbar}</td></tr> <tr><td>α</td><td>0.0027</td></tr> <tr><td>ARL</td><td>370.4 samples</td></tr> </tbody> </table> <table border="1" style="width: 45%; border-collapse: collapse;"> <thead> <tr> <th colspan="2" style="background-color: #00a651; color: white;">Control Limits for R Chart</th> </tr> </thead> <tbody> <tr><td>CL_R</td><td>4,675</td></tr> <tr><td>UCL_R</td><td>5,185</td></tr> <tr><td>LCL_R</td><td>4,165</td></tr> </tbody> </table> </div> </div>	Statistics from Raw Data Table		R-bar	4,671	Process Mean, μ -hat	4,671	Process St.Dev., σ -hat	155	σ_{xbar}	16	Certified Values		See Below		Control Limits for X-bar Chart		CL _{xbar}	4671	UCL _{xbar}	4,720 CL+k σ_{xbar}	LCL _{xbar}	4,623 CL-k σ_{xbar}	α	0.0027	ARL	370.4 samples	Control Limits for R Chart		CL _R	4,675	UCL _R	5,185	LCL _R	4,165
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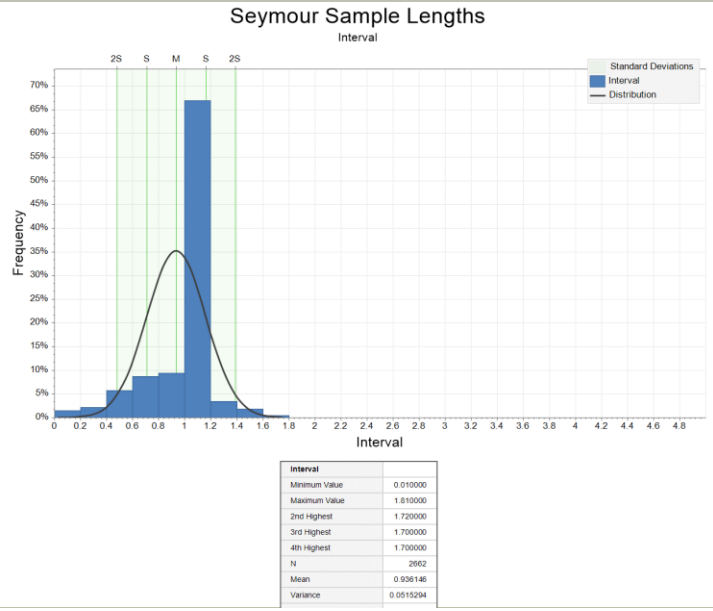
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		<ul style="list-style-type: none"> All independent certified reference data returns were within acceptable limits with no discernible bias, except one blank sample that appears to have been a field swap. The major element oxides and trace elements including Rb, Cs, Nb, Ta and Be were analyzed by FUS-ICP and FUS-MS (4Litho-Pegmatite Special) analytical codes which uses a lithium metaborate tetraborate fusion with analysis by ICP and ICPMS. Historic specific gravity testwork was determined for every 10th sample by RX17-GP analytical code measured on the pulp by a gas pycnometer. More recently GT1 submitted 226 samples for water immersion test work by Actlabs prior to samples preparation.
Verification of sampling and assaying	<ul style="list-style-type: none"> The verification of significant intersections by either independent or alternative company personnel. The use of twinned holes. Documentation of primary data, data entry procedures, data verification, data storage (physical and electronic) protocols. Discuss any adjustment to assay data. 	<ul style="list-style-type: none"> Ardiden drilled 17 diamond holes within 8m of hole drilled by the previous owner, Linear, in 2016 and 2017. The results were discussed in the previous section, <i>Quality of assay data and laboratory tests</i>. Whilst the result was erratic Ardidien were able to confirm the presence of high grade LCT pegmatites. Further drilling undertaken by GT1 has also confirmed the high grade nature of the main pegmatite (North Upper - HG). The majority of laboratory assay results have been sourced directly from the laboratory and the laboratory file directly imported into GT1's SQL database. All recent north seeking gyroscope surveys are uploaded directly from the survey tool output file and visually validated. Geological logs and supporting data are uploaded directly to the database using custom built importers to ensure no chance of typographical errors. No adjustment to laboratory assay data was made.
Location of data points	<ul style="list-style-type: none"> Accuracy and quality of surveys used to locate drill holes (collar and down-hole surveys), trenches, mine workings and other locations used in Mineral Resource estimation. Specification of the grid system used. Quality and adequacy of topographic control. 	<ul style="list-style-type: none"> A GPS reading was taken for each sample location using UTM NAD83 Zone16 (for Seymour); waypoint averaging or dGPS was performed when possible. Ardiden undertook a Lidar survey of the Seymour area in 2018 (+/- 0.15m) which underpins the local topographic surface. All drill collars have been draped onto the LIDAR surface to ensure accurate elevation data for the drillholes. GT1 employed a calibrated Reflex SprintIQ North Seeking Gyroscopic tool on all 2021 and 2022 drill holes and surveyed the holes in their entirety with readings downhole every 5m. North Seeking gyroscopes have a typical azimuth accuracy of +/- 0.75 degrees and +/-0.15 degrees for dip.

Criteria	JORC Code explanation	Commentary
		 <p data-bbox="709 857 1860 906">All collars are picked up and stored in the database in North American Datum of 1983 (NAD83) Zone 16 horizontal and geometric control datum projection for the United States.</p>
<p data-bbox="121 971 296 1019">Data spacing and distribution</p>	<ul data-bbox="352 971 678 1269" style="list-style-type: none"> • <i>Data spacing for reporting of Exploration Results.</i> • <i>Whether the data spacing and distribution is sufficient to establish the degree of geological and grade continuity appropriate for the Mineral Resource and Ore Reserve estimation procedure(s) and classifications applied.</i> • <i>Whether sample compositing has been applied.</i> 	<ul data-bbox="709 971 1875 1101" style="list-style-type: none"> • The Seymour pegmatites in the North and South areas of the deposit have variable drill spacing from 20Ex20m in the shallower areas (<150m) of the deposit to 50mEx50mN at lower depths (150-250m) and greater than 80m spacing below this depth. • The drill spacing is sufficient to support the various levels of Mineral Resource classification applied to the estimate. • 1m compositing was applied to the Seymour Mineral Resource update based on a review of sample interval lengths.

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<p>Orientation of data in relation to geological structure</p>	<ul style="list-style-type: none"> Whether the orientation of sampling achieves unbiased sampling of possible structures and the extent to which this is known, considering the deposit type. If the relationship between the drilling orientation and the orientation of key mineralised structures is considered to have introduced a sampling bias, this should be assessed and reported if material. 	<ul style="list-style-type: none"> GT1 drill samples were drilled close to perpendicular to the strike of the pegmatite unit and sampled the entire length of the pegmatite as well including several metres into the mafic country rock either side of the pegmatite. Grab and trench samples were taken where outcrop was available. All attempts were made to ensure trench samples represented traverses across strike of the pegmatite. Older holes from Linear Metals and some of Ardidens earlier drilling were vertical and only approximated the true widths of the pegmatites. 																		
<p>Sample security</p>	<ul style="list-style-type: none"> The measures taken to ensure sample security. 	<ul style="list-style-type: none"> All core and samples were supervised and secured in a locked vehicle, warehouse, or container until delivered to Actlabs in Thunder Bay for cutting, preparation and analysis. 																		
<p>Audits or reviews</p>	<ul style="list-style-type: none"> The results of any audits or reviews of sampling techniques and data. 	<ul style="list-style-type: none"> No independent audits or reviews have been undertaken on this Mineral Resource estimate. 																		

Section 2 Reporting of Exploration Results

Criteria	JORC Code explanation	Commentary
Mineral tenement and land tenure status	<ul style="list-style-type: none"> Type, reference name/number, location and ownership including agreements or material issues with third parties such as joint ventures, partnerships, overriding royalties, native title interests, historical sites, wilderness or national park and environmental settings. The security of the tenure held at the time of reporting along with any known impediments to obtaining a licence to operate in the area. 	<ul style="list-style-type: none"> Seymour Lithium Asset consists of 744 Cell Claims (Exploration Licences) with a total claim area of 15,058 ha. All Cell Claims are in good standing An Active Exploration Permit exists over the Seymour Lithium Assets An Early Exploration Agreement is current with the Whitesand First Nation who are supportive of GT1 exploration activities.
Exploration done by other parties	<ul style="list-style-type: none"> Acknowledgment and appraisal of exploration by other parties. 	<ul style="list-style-type: none"> Regional exploration for lithium deposits commenced in the 1950's. In 1957, local prospector, Mr Nelson Aubry, discovered the North Aubry and the South Aubry pegmatites. Geological mapping by the Ontario Department of Mines commenced in 1959 and was completed in 1962 (Pye, 1968), with the publication of "Map 2100 Crescent Lake Area" in 1965. From the late 1950's to 2002, exploration by the Ontario Department of Mines was generally restricted to geological mapping and surface sampling, although some minor drilling was completed to test the North Aubry pegmatite in late 1957 (Rees, 2011). In 2001, Linear Resources Inc. ("Linear Resources") obtained the Seymour Lake Project with an initial focus on the project's tantalum potential. In 2002, a 23-diamond drill-hole campaign was completed at North Aubry, and a further 8 diamond drill-holes at South Aubry. In 2008, Linear Resources completed a regional soil-sampling program which resulted in the identification of a number soil geochemical anomalies. Based on these anomalies, another drilling campaign (completed in 2009), with 12 diamond drill-holes at North Aubry, 2 diamond drill-holes at South Aubry, and further 5 diamond drill-holes peripheral to the Aubry prospects designed to test the main 2008 soil geochemical anomalies. Little work was undertaken between 2010 and 2016 until Ardiden acquired the project from Linear Resources in 2016. Further drilling was carried out by Ardiden between 2017 and 2018 resulting in the completion of an updated mineral resource estimate of the Aubry pegmatites in 2018. Ground Penetrating Radar (GPR) was also undertaken by Ardiden in 2018 to test any further exploration potential beyond the current Aubry pegmatite delineating numerous targets.
Geology	<ul style="list-style-type: none"> Deposit type, geological setting and style of mineralisation. 	<ul style="list-style-type: none"> Regional Geology: The general geological setting of the Seymour Lithium Asset consists of the Precambrian Canadian Shield that underlies approximately 60% of Ontario. The Shield can be divided into three major geological and physiographic regions, from the oldest in the northwest to the youngest in the southeast. Local Geology: The Seymour Lithium Asset is located within the eastern part of the Wabigoon Subprovince, near the boundary with the English River Subprovince to the north. These subprovinces are part of the Superior Craton, comprised mainly of Archaean rocks but also containing some Mesoproterozoic rocks such as the Nipigon Diabase. Bedrock Geology: The bedrock is best exposed along the flanks of steep-sided valleys scoured by glaciers during the recent ice ages. The exposed bedrock is commonly metamorphosed basaltic rock, of which some varieties have well-preserved pillows that have been intensely flattened in areas of high tectonic strain. Intercalated between layers of basalt are lesser amounts of schists derived from sedimentary rocks and lesser rocks having felsic volcanic protoliths. These rocks are typical of the Wabigoon Subprovince, host to most of the pegmatites in the region. Ore Geology: Pegmatites are reasonably common in the region intruding the enclosing host rocks after metamorphism, evident from the manner in which the pegmatites cut across the well developed foliation within the metamorphosed host rocks. This post-dating relationship is supported by radiometric dating; an age of 2666 ± 6 Ma is given for the timing of

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		<p>intrusion of the pegmatites (Breaks, et al., 2006).</p> <ul style="list-style-type: none"> The pegmatites in North Aubry have a northeast plunge direction varying from 10 to 35 degrees from horizontal some 800m downdip extent and 250-300m strike. The North Upper and North Upper high grade component within, appears to wedge towards the south east and is still open down dip and to the north west. Southern pegmatites are thinner and less well developed with higher muscovite content and appear to have a more north to north-westerly trend and dip more shallowly to the east. These pegmatites are also hosted in pillow basalts. The pegmatites are zoned with better developed spodumene crystal appearing as bands, often at an acute angle to the general trend of the pegmatite. The dominant economic minerals are spodumene with varying proportions of muscovite, microcline, and minor petalite and lepidolite. The adjacent pillow basalts contain minor disseminated pyrite and pyrrhotite. 																																																																																																																																																											
Drill hole Information	<ul style="list-style-type: none"> A summary of all information material to the understanding of the exploration results including a tabulation of the following information for all Material drill holes: <ul style="list-style-type: none"> easting and northing of the drill hole collar elevation or RL (Reduced Level - elevation above sea level in metres) of the drill hole collar dip and azimuth of the hole down hole length and interception depth hole length. If the exclusion of this information is justified on the basis that the information is not Material and this exclusion does not detract from the understanding of the report, the Competent Person should clearly explain why this is the case. 	<ul style="list-style-type: none"> A total of 199 diamond holes, on a nominal 20m x 20m grid, have been used in the resource modelling at North Aubry and South Aubry. A total of 130 holes were drilled by Ardiden, with the previous owners Linear drilling 42 holes, some of which were excluded from this estimate due to missing logging, assay reliability or re-drills. The 2018 Ardiden drilling was completed by Rugged Aviation Inc. using BTW coring equipment producing 4.20 cm diameter core. <table border="1"> <caption>Drilling Used in Mineral Resource</caption> <thead> <tr> <th rowspan="2">Company</th> <th colspan="4">Holes</th> <th colspan="4">Metres</th> <th colspan="4">All Companies</th> </tr> <tr> <th>Linear</th> <th>Ardiden</th> <th>Green</th> <th>Total</th> <th>Linear</th> <th>Ardiden</th> <th>Green Tech</th> <th>Total</th> <th>BTW</th> <th>NQ</th> <th>HQ</th> <th>NR</th> </tr> </thead> <tbody> <tr> <td>Year</td> <td>DDH</td> <td>DDH</td> <td>DDH</td> <td>DDH</td> <td>DDH</td> <td>DDH</td> <td>DDH</td> <td>DDH</td> <td>BTW</td> <td>NQ</td> <td>HQ</td> <td>NR</td> </tr> <tr> <td>2002</td> <td>30</td> <td>0</td> <td>0</td> <td>30</td> <td>1,677.45</td> <td>-</td> <td>-</td> <td>-</td> <td>-</td> <td>-</td> <td>1,677.45</td> <td>-</td> </tr> <tr> <td>2009</td> <td>12</td> <td>0</td> <td>0</td> <td>12</td> <td>1,573.50</td> <td>-</td> <td>-</td> <td>1,573.50</td> <td>-</td> <td>-</td> <td>-</td> <td>1,573.50</td> </tr> <tr> <td>2016</td> <td>0</td> <td>29</td> <td>0</td> <td>29</td> <td>-</td> <td>1,950.00</td> <td>-</td> <td>1,950.00</td> <td>-</td> <td>-</td> <td>-</td> <td>1,950.00</td> </tr> <tr> <td>2017</td> <td>0</td> <td>69</td> <td>0</td> <td>69</td> <td>-</td> <td>7,864.29</td> <td>-</td> <td>7,864.29</td> <td>-</td> <td>-</td> <td>-</td> <td>7,864.29</td> </tr> <tr> <td>2018</td> <td>0</td> <td>37</td> <td>0</td> <td>37</td> <td>-</td> <td>6,564.71</td> <td>-</td> <td>6,564.71</td> <td>4,659.71</td> <td>-</td> <td>-</td> <td>1,905.00</td> </tr> <tr> <td>2021</td> <td>0</td> <td>0</td> <td>1</td> <td>1</td> <td>-</td> <td>-</td> <td>341.00</td> <td>341.00</td> <td>-</td> <td>-</td> <td>341.00</td> <td>-</td> </tr> <tr> <td>2022</td> <td>0</td> <td>0</td> <td>21</td> <td>21</td> <td>-</td> <td>-</td> <td>7,950.69</td> <td>7,950.69</td> <td>-</td> <td>-</td> <td>7,219.69</td> <td>731.00</td> </tr> <tr> <td>Grand Total</td> <td>42</td> <td>135</td> <td>22</td> <td>199</td> <td>3,250.95</td> <td>16,379.00</td> <td>8,291.69</td> <td>26,244.19</td> <td>4,659.71</td> <td>9,238.14</td> <td>2,636.00</td> <td>10,620.50</td> </tr> <tr> <td>Proportion</td> <td>21%</td> <td>68%</td> <td>11%</td> <td></td> <td>12%</td> <td>62%</td> <td>32%</td> <td></td> <td>18%</td> <td>35%</td> <td>10%</td> <td>40%</td> </tr> </tbody> </table> <p>18 holes were excluded for the MRE Excluded holes were largely from 2009 and 2010 where logging and/or geological data was missing from the record or the holes were re-drills.</p> <ul style="list-style-type: none"> The earlier drill holes were either vertical or inclined towards the west. Once the pegmatite was determined to be dipping towards the north-east, the later drill holes were inclined towards the south-west Green Technology Metals Ltd has completed 34 NQ diamond holes since December 2021, of which 22 holes are included in the current Mineral Resource estimate, with the following collar coordinates: 	Company	Holes				Metres				All Companies				Linear	Ardiden	Green	Total	Linear	Ardiden	Green Tech	Total	BTW	NQ	HQ	NR	Year	DDH	DDH	DDH	DDH	DDH	DDH	DDH	DDH	BTW	NQ	HQ	NR	2002	30	0	0	30	1,677.45	-	-	-	-	-	1,677.45	-	2009	12	0	0	12	1,573.50	-	-	1,573.50	-	-	-	1,573.50	2016	0	29	0	29	-	1,950.00	-	1,950.00	-	-	-	1,950.00	2017	0	69	0	69	-	7,864.29	-	7,864.29	-	-	-	7,864.29	2018	0	37	0	37	-	6,564.71	-	6,564.71	4,659.71	-	-	1,905.00	2021	0	0	1	1	-	-	341.00	341.00	-	-	341.00	-	2022	0	0	21	21	-	-	7,950.69	7,950.69	-	-	7,219.69	731.00	Grand Total	42	135	22	199	3,250.95	16,379.00	8,291.69	26,244.19	4,659.71	9,238.14	2,636.00	10,620.50	Proportion	21%	68%	11%		12%	62%	32%		18%	35%	10%	40%
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2018	0	37	0	37	-	6,564.71	-	6,564.71	4,659.71	-	-	1,905.00																																																																																																																																																	
2021	0	0	1	1	-	-	341.00	341.00	-	-	341.00	-																																																																																																																																																	
2022	0	0	21	21	-	-	7,950.69	7,950.69	-	-	7,219.69	731.00																																																																																																																																																	
Grand Total	42	135	22	199	3,250.95	16,379.00	8,291.69	26,244.19	4,659.71	9,238.14	2,636.00	10,620.50																																																																																																																																																	
Proportion	21%	68%	11%		12%	62%	32%		18%	35%	10%	40%																																																																																																																																																	

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Data aggregation methods	<ul style="list-style-type: none"> In reporting Exploration Results, weighting averaging techniques, maximum and/or minimum grade truncations (eg cutting of high grades) and cut-off grades are usually Material and should be stated. Where aggregate intercepts incorporate short lengths of high grade results and longer lengths of low grade results, the procedure used for such aggregation should be stated and some typical examples of such aggregations should be shown in detail. The assumptions used for any reporting of metal equivalent values should be clearly stated. 	<ul style="list-style-type: none"> length weighted averages and all resource estimates are tonnage weighted averages Grade cut-offs have not been incorporated. No metal equivalent values are quoted. 																																																																																																																																										
Relationship between mineralisation widths and intercept lengths	<ul style="list-style-type: none"> These relationships are particularly important in the reporting of Exploration Results. If the geometry of the mineralisation with respect to the drill hole angle is known, its nature should be reported. 	<ul style="list-style-type: none"> The historic reported results are stated as down hole lengths. The historic pierce angle of the drilling with the pegmatite varies hole by hole so all intersection widths are longer than true widths. The resource modelling considers the intersections in 3D and adjusts accordingly. Holes drilled by GT1 attempt to pierce the mineralised pegmatite approximately perpendicular to strike, and therefore, the downhole intercepts reported are approximately equivalent to the true width of the mineralisation. Trenches are representative widths of the exposed pegmatite outcrop. Some exposure may not be a complete 																																																																																																																																										

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	<ul style="list-style-type: none"> If it is not known and only the down hole lengths are reported, there should be a clear statement to this effect (eg 'down hole length, true width not known'). 	<p>representation of the total pegmatite width due to recent glacial deposit cover limiting the available material to be sampled.</p>
Diagrams	<ul style="list-style-type: none"> Appropriate maps and sections (with scales) and tabulations of intercepts should be included for any significant discovery being reported These should include, but not be limited to a plan view of drill hole collar locations and appropriate sectional views. 	<ul style="list-style-type: none"> The appropriate maps are included in the announcement.

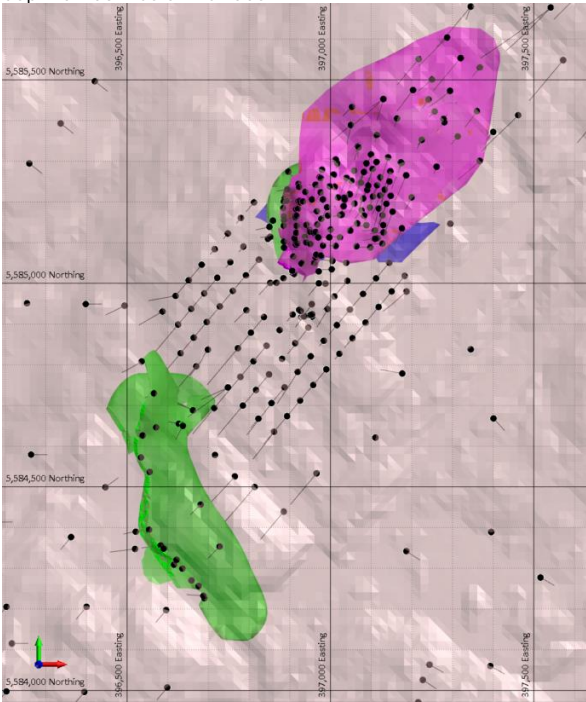
Balanced reporting

- Where comprehensive reporting of all Exploration Results is not practicable, representative reporting of both low and high grades and/or widths should be practiced to avoid misleading reporting of Exploration Results.
- Pegmatite downhole interval summary with associated assay results are listed in Appendix A

Criteria	JORC Code explanation	Commentary
Other substantive exploration data	<ul style="list-style-type: none"> Other exploration data, if meaningful and material, should be reported including (but not limited to): geological observations; geophysical survey results; geochemical survey results; bulk samples – size and method of treatment; metallurgical test results; bulk density, groundwater, geotechnical and rock characteristics; potential deleterious or contaminating substances. 	<ul style="list-style-type: none"> GT1 completed a fixed wing single sensor magnetic/radiometric/VLF airborne geophysical survey. Survey details, 1191 line-km, 75m line spacing, direction 90 degrees to cross cut pegmatite strike, 70m altitude. Preliminary images have been received for Total Count Radiometric, Total Magnetics and VLF. Raw data currently being processed by MPX Geophysics. Interpretation will be completed by Southern Geoscience
Further work	<ul style="list-style-type: none"> The nature and scale of planned further work (eg tests for lateral extensions or depth extensions or large-scale step-out drilling). Diagrams clearly highlighting the areas of possible extensions, including the main geological interpretations and future drilling areas, provided this information is not commercially sensitive. 	<ul style="list-style-type: none"> Test further potential downdip extensions and pegmatite stacking at North Aubry. Drilling program commencement at Root and Morssion prospects. Geological field mapping of anomalies and associated pegmatites at Seymour and regional claims. Sampling pegmatites for spodumene Drill targeting and followed by diamond drilling over the next 24 months. Commencement of detailed mining studies

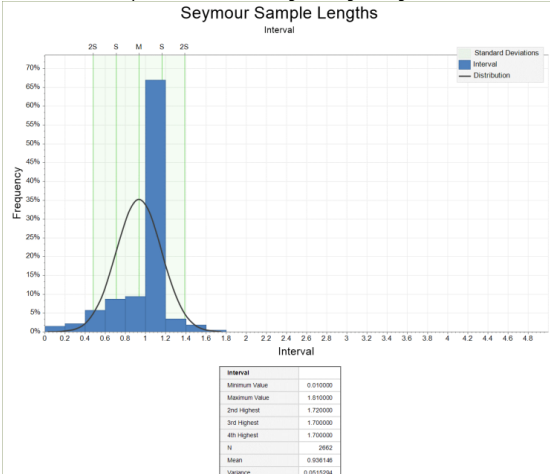
Section 3 Estimation and Reporting of Mineral Resources

Criteria	JORC Code explanation	Commentary
Database integrity	<ul style="list-style-type: none"> Measures taken to ensure that data has not been corrupted by, for example, transcription or keying errors, between its initial collection and its use for Mineral Resource estimation purposes. Data validation procedures used. 	<ul style="list-style-type: none"> Data was imported into the database directly from source geology logs and laboratory csv files. Was then passed through a series of validation checks before final acceptance of the data for downstream use.
Site visits	<ul style="list-style-type: none"> Comment on any site visits undertaken by the Competent Person and the outcome of those visits. If no site visits have been undertaken indicate why this is the case. 	<ul style="list-style-type: none"> A site visit was undertaken by the Competent Person (John Winterbottom) between 8th and 9th June 2022; general site layout, drilling sites, diamond drilling operations were viewed, plus diamond core in the storage facility Thunder Bay.

Criteria	JORC Code explanation	Commentary
<p>Geological interpretation</p>	<ul style="list-style-type: none"> Confidence in (or conversely, the uncertainty of) the geological interpretation of the mineral deposit. Nature of the data used and of any assumptions made. The effect, if any, of alternative interpretations on Mineral Resource estimation. The use of geology in guiding and controlling Mineral Resource estimation. The factors affecting continuity both of grade and geology. 	<ul style="list-style-type: none"> There is good confidence in the geological interpretation of the deposit in most areas; there are some areas of uncertainty at the outer limits of the deposit where drill spacing is sparse. Interpretation was made directly from pegmatites noted in geological logs and confirmation through core photographs. Alternative geological interpretation would have a minimal effect on the resource estimate. Pegmatite intrusions were used to constrain the mineral resource estimation. Continuity of grade and geology is strongly tied to pegmatite thickness that varies considerably throughout the deposit due to structural elongation and dilation dynamics.
<p>Dimensions</p>	<ul style="list-style-type: none"> The extent and variability of the Mineral Resource expressed as length (along strike or otherwise), plan width, and depth below surface to the upper and lower limits of the Mineral Resource. 	<ul style="list-style-type: none"> The deposit consists of a number of stacked pegmatite units of varying thicknesses. The deposit consists of two principal areas North and South The Northern area of the deposit has a maximum horizontal extent of 800m, 390m wide and varies from 2m up to 43m in thickness. 5 mineralised pegmatites that have been interpreted down to a depth of 350m below surface and is still open at depth. Pegmatites dip approximately 30-35 degrees to the northeast. The Southern area consists of an Upper and a Lower pegmatite. The Upper pegmatite is continuous over the entire extent of the Southern deposit whilst the Lower pegmatite is broken into a northern and southern half. The Southern area extends upto 740m along a 330 strike direction, upto 170m across with thickness varying from 0 to 22m, with a maximum depth of 130m below surface. 

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<p>Estimation and modelling techniques</p>	<ul style="list-style-type: none"> The nature and appropriateness of the estimation technique(s) applied and key assumptions, including treatment of extreme grade values, domaining, interpolation parameters and maximum distance of extrapolation from data points. If a computer assisted estimation method was chosen include a description of computer software and parameters used. The availability of check estimates, previous estimates and/or mine production records and whether the Mineral Resource estimate takes appropriate account of such data. The assumptions made regarding recovery of by-products. Estimation of deleterious elements or other non-grade variables of economic significance (eg sulphur for acid mine drainage characterisation). In the case of block model interpolation, the block size in relation to the average sample spacing and the search employed. Any assumptions behind modelling of selective mining units. Any assumptions about correlation between variables. Description of how the geological interpretation was used to control the resource estimates. Discussion of basis for using or not using grade cutting or capping. The process of validation, the checking process used, the comparison of model data to drill hole data, and use of 	<ul style="list-style-type: none"> An Ordinary Kriging (OK) grade estimation methodology has been used for Li₂O in the Mineral Resource Estimate which is considered appropriate for the style of mineralisation under review. OK was also applied to important potential bi-product or deleterious elements (Ta₂O₅, Rb₂O, Cs, K, Fe, Mg, Nb). Secondary elements were not exhaustively assayed for in the historic areas of the resource and therefore are only approximations at this stage and have not been included in the Mineral Resource figures. Micromine 2022.4 software was used for estimation, statistical and geostatistical data analysis. A previous estimate of the deposit was made by Phillip Jones, an independent consultant employed by Ardiden Ltd in April 2019. <table border="1"> <thead> <tr> <th>Area</th> <th>Category</th> <th>Mt</th> <th>Li₂O (%)</th> </tr> </thead> <tbody> <tr> <td>North Aubry</td> <td>Indicated</td> <td>2.1</td> <td>1.29</td> </tr> <tr> <td>North Aubry</td> <td>Inferred</td> <td>1.7</td> <td>1.50</td> </tr> <tr> <td>South Aubry</td> <td>Inferred</td> <td>1.0</td> <td>0.80</td> </tr> <tr> <td>TOTAL</td> <td></td> <td>4.8</td> <td>1.25</td> </tr> </tbody> </table> <ul style="list-style-type: none"> The previous mineral resource was constrained within the pegmatite units and reported above a zero cut-off Li₂O grade. <table border="1"> <thead> <tr> <th>Pegmatite</th> <th>Volume</th> </tr> </thead> <tbody> <tr> <td>South Upper:</td> <td>742,546</td> </tr> <tr> <td>South Lower:</td> <td>150,664</td> </tr> <tr> <td>North Upper (incl HG):</td> <td>2,330,000</td> </tr> <tr> <td>North Lower:</td> <td>565,460</td> </tr> <tr> <td>North HW:</td> <td>73,568</td> </tr> <tr> <td>North Minor:</td> <td>5,362</td> </tr> <tr> <td>North Upper spur:</td> <td>106,210</td> </tr> <tr> <td>North FW:</td> <td>91,656</td> </tr> <tr> <td>North HW Minor:</td> <td>10,955</td> </tr> <tr> <td>Total</td> <td>4,076,420</td> </tr> </tbody> </table> <ul style="list-style-type: none"> Geological units were first interpreted in Leapfrog 2021.2 software from geological logs and core photography references. Pegmatite and overburden wireframes were exported from Leapfrog and then imported into Micromine for estimation. Data was composited to 1m length to geological contacts.  <p style="text-align: center;">Seymour Sample Lengths</p> <p>Interval</p> <p>Frequency</p> <p>Standard Deviations</p> <p>Interval</p> <p>Distribution</p> <table border="1"> <thead> <tr> <th>Interval</th> <th>Value</th> </tr> </thead> <tbody> <tr> <td>Minimum Value</td> <td>0.010000</td> </tr> <tr> <td>Maximum Value</td> <td>1.010000</td> </tr> <tr> <td>1st Highest</td> <td>1.700000</td> </tr> <tr> <td>3rd Highest</td> <td>1.700000</td> </tr> <tr> <td>4th Highest</td> <td>1.700000</td> </tr> <tr> <td>N</td> <td>2962</td> </tr> <tr> <td>Mean</td> <td>0.208148</td> </tr> <tr> <td>Variance</td> <td>0.0515294</td> </tr> </tbody> </table>	Area	Category	Mt	Li ₂ O (%)	North Aubry	Indicated	2.1	1.29	North Aubry	Inferred	1.7	1.50	South Aubry	Inferred	1.0	0.80	TOTAL		4.8	1.25	Pegmatite	Volume	South Upper:	742,546	South Lower:	150,664	North Upper (incl HG):	2,330,000	North Lower:	565,460	North HW:	73,568	North Minor:	5,362	North Upper spur:	106,210	North FW:	91,656	North HW Minor:	10,955	Total	4,076,420	Interval	Value	Minimum Value	0.010000	Maximum Value	1.010000	1st Highest	1.700000	3rd Highest	1.700000	4th Highest	1.700000	N	2962	Mean	0.208148	Variance	0.0515294
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	reconciliation data if available.	<ul style="list-style-type: none"> Variography was carried out to define the variogram models for the Ordinary Kriging (OK) interpolation. <p>North Upper Li20</p> <p>North Upper Ta205</p>

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		<p>blocks upto 1/5 of the parent blocks to preserve pegmatite volumes.</p> <ul style="list-style-type: none"> Model dimensions are shown below: <table border="1"> <thead> <tr> <th>Model</th> <th>minX</th> <th>maxX</th> <th>minY</th> <th>MaxY</th> <th>minZ</th> <th>maxZ</th> <th>xdim</th> <th>ydim</th> <th>zdim</th> <th>Rotation</th> </tr> </thead> <tbody> <tr> <td>bm_seymour_north_May22</td> <td>396555.00</td> <td>397805.00</td> <td>5584902.50</td> <td>5586002.50</td> <td>- 61.25</td> <td>418.75</td> <td>10</td> <td>5</td> <td>2.5</td> <td>45</td> </tr> <tr> <td>bm_seymour_south_May22</td> <td>396295.00</td> <td>397005.00</td> <td>5584005.00</td> <td>5584895.00</td> <td>151.25</td> <td>400.00</td> <td>10</td> <td>10</td> <td>2.5</td> <td>0</td> </tr> </tbody> </table> <ul style="list-style-type: none"> Recovery of by-products will be determined following detailed metallurgical testwork. Estimated averages for bi product and deleterious elements for North Aubry are tabulated below but are not available for South Aubry as testwork was limited to Li₂O and Ta₂O₅. <p style="text-align: center;">Bi-product and Deleterious elements</p> <p style="text-align: center;">Reported within \$US4000 pit shell above 0.2% Li₂O</p> <table border="1"> <tbody> <tr> <td>Tonnes (mt)</td> <td>7.8</td> </tr> <tr> <td>Li₂O%</td> <td>1.17</td> </tr> <tr> <td>Ta₂O₅ppm</td> <td>148</td> </tr> <tr> <td>Rb₂O ppm</td> <td>2,550</td> </tr> <tr> <td>K ppm</td> <td>17,800</td> </tr> <tr> <td>Fe ppm</td> <td>8,170</td> </tr> <tr> <td>Mg ppm</td> <td>2,120</td> </tr> <tr> <td>Nb ppm</td> <td>62</td> </tr> <tr> <td>Cs ppm</td> <td>400</td> </tr> </tbody> </table> <ul style="list-style-type: none"> Acid mine drainage estimates will be made on the return of waste rock samples for multi-elemental analysis, including sulphur. Locally Varying Trend (LVA) models were produced for each pegmatite from Leapfrog footwall surfaces. Multiple passes were used to ensure blocks are filled in areas with sparser drilling. Searches of 50m, 100 and 150m with applied anisotropy and orientation to the search ellipsoid based on the trend model were made. Sample data was composited to 1m down-hole composites, while honouring geological contacts. Top cut analysis was carried out to identify extreme outliers, using a combination plots, and histograms and the effect of top cuts on cut mean and coefficient of variation. Variable top cuts have been applied by domain and element, as follows:. 	Model	minX	maxX	minY	MaxY	minZ	maxZ	xdim	ydim	zdim	Rotation	bm_seymour_north_May22	396555.00	397805.00	5584902.50	5586002.50	- 61.25	418.75	10	5	2.5	45	bm_seymour_south_May22	396295.00	397005.00	5584005.00	5584895.00	151.25	400.00	10	10	2.5	0	Tonnes (mt)	7.8	Li₂O%	1.17	Ta₂O₅ppm	148	Rb₂O ppm	2,550	K ppm	17,800	Fe ppm	8,170	Mg ppm	2,120	Nb ppm	62	Cs ppm	400
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		<p style="text-align: center;">North Upper HG Li2O_ppm</p> <p>The figure displays six statistical plots for Li₂O_ppm data from the North Upper HG site:</p> <ul style="list-style-type: none"> Histogram: Shows the frequency distribution of Li₂O_ppm values, with a peak frequency of approximately 5.5% at low concentrations (below 10,000 ppm). Probability Plot: Plots Li₂O_ppm against cumulative probability, showing a non-linear relationship characteristic of a skewed distribution. Mean vs. Top Cut: Shows the relationship between the mean value and the top cut value, with the mean increasing and leveling off as the top cut increases. COV vs. Top Cut: Shows the Coefficient of Variation (COV) decreasing as the top cut value increases, indicating reduced relative variability at higher concentrations. Relative Nugget vs. Top Cut: Plots the relative nugget value against the top cut value, showing a relatively stable trend around 0.4. Decile Analysis: A bar chart showing the percentage of total Li₂O_ppm contained within various deciles, with the highest percentage (around 27%) in the 90-100 decile.

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<tr><td>Cs_ppm</td><td>North FW</td><td>1,352</td><td>254</td><td>1.4</td><td>230</td><td>0.99</td></tr> <tr><td>Li2O_ppm</td><td>North HW</td><td>4,895</td><td>736</td><td>1.9</td><td>695</td><td>1.76</td></tr> <tr><td>Ta2O5_ppm</td><td>North HW</td><td>551</td><td>187</td><td>0.8</td><td>185</td><td>0.73</td></tr> <tr><td>Rb2O_ppm</td><td>North HW</td><td>5,238</td><td>1,705</td><td>0.9</td><td>1,697</td><td>0.85</td></tr> <tr><td>K_ppm</td><td>North HW</td><td>33,566</td><td>13,372</td><td>0.7</td><td>13,332</td><td>0.73</td></tr> <tr><td>Fe_ppm</td><td>North HW</td><td>79,473</td><td>11,162</td><td>1.7</td><td>10,942</td><td>1.67</td></tr> <tr><td>Mg_ppm</td><td>North HW</td><td>34,420</td><td>3,620</td><td>2.6</td><td>3,177</td><td>2.24</td></tr> <tr><td>Nb_ppm</td><td>North HW</td><td>143</td><td>58</td><td>0.7</td><td>56</td><td>0.52</td></tr> <tr><td>Cs_ppm</td><td>North HW</td><td>535</td><td>137</td><td>1.1</td><td>135</td><td>1.07</td></tr> <tr><td>Li2O_ppm</td><td>North 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<tr><td>Rb2O_ppm</td><td>North Upper</td><td>11,621</td><td>3,021</td><td>1.0</td><td>3,006</td><td>1.00</td></tr> <tr><td>K_ppm</td><td>North Upper</td><td>87,187</td><td>25,278</td><td>1.0</td><td>25,201</td><td>1.04</td></tr> <tr><td>Fe_ppm</td><td>North Upper</td><td>91,781</td><td>16,587</td><td>1.6</td><td>16,491</td><td>1.61</td></tr> <tr><td>Mg_ppm</td><td>North Upper</td><td>43,706</td><td>6,073</td><td>2.0</td><td>6,041</td><td>1.98</td></tr> <tr><td>Nb_ppm</td><td>North Upper</td><td>294</td><td>64</td><td>1.0</td><td>63</td><td>0.93</td></tr> <tr><td>Cs_ppm</td><td>North Upper</td><td>2,984</td><td>453</td><td>1.6</td><td>411</td><td>1.17</td></tr> <tr><td>Li2O_ppm</td><td>North Upper HG</td><td>44,654</td><td>13,208</td><td>0.9</td><td>13,102</td><td>0.84</td></tr> <tr><td>Ta2O5_ppm</td><td>North Upper HG</td><td>747</td><td>175</td><td>2.5</td><td>148</td><td>0.90</td></tr> <tr><td>Rb2O_ppm</td><td>North Upper 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Upper</td><td>331</td><td>118</td><td>0.68</td><td>118</td><td>0.68</td></tr> </tbody> </table>	Field Name	Key	Top Cut	Mean	Coeff. of Variation	Mean	Coeff. of Variation	Li2O_ppm	North FW	17,749	4,224	1.2	4,213	1.19	Ta2O5_ppm	North FW	817	156	1.1	156	1.11	Rb2O_ppm	North FW	4,980	2,113	0.5	2,105	0.50	K_ppm	North FW	44,230	21,587	0.5	21,383	0.47	Fe_ppm	North FW	20,091	5,876	0.7	5,765	0.62	Mg_ppm	North FW	4,984	1,091	1.1	1,058	0.98	Nb_ppm	North FW	122	57	0.5	56	0.50	Cs_ppm	North FW	1,352	254	1.4	230	0.99	Li2O_ppm	North HW	4,895	736	1.9	695	1.76	Ta2O5_ppm	North HW	551	187	0.8	185	0.73	Rb2O_ppm	North HW	5,238	1,705	0.9	1,697	0.85	K_ppm	North HW	33,566	13,372	0.7	13,332	0.73	Fe_ppm	North HW	79,473	11,162	1.7	10,942	1.67	Mg_ppm	North HW	34,420	3,620	2.6	3,177	2.24	Nb_ppm	North HW	143	58	0.7	56	0.52	Cs_ppm	North HW	535	137	1.1	135	1.07	Li2O_ppm	North Lower	36,955	8,423	1.2	8,358	1.17	Ta2O5_ppm	North Lower	710	195	1.1	184	0.76	Rb2O_ppm	North Lower	8,137	2,390	0.8	2,360	0.74	K_ppm	North Lower	60,479	19,204	0.7	19,017	0.68	Fe_ppm	North Lower	82,207	12,216	1.5	12,104	1.46	Mg_ppm	North Lower	39,912	4,179	2.1	4,116	2.10	Nb_ppm	North Lower	159	59	0.7	57	0.56	Cs_ppm	North Lower	1,250	355	0.9	340	0.70	Li2O_ppm	North Minor	14,246	4,260	1.2	4,254	1.24	Ta2O5_ppm	North Minor	241	132	0.4	132	0.44	Rb2O_ppm	North Minor	5,693	2,211	0.7	2,194	0.65	K_ppm	North Minor	57,145	19,282	0.8	19,215	0.77	Fe_ppm	North Minor	27,068	8,547	0.7	8,547	0.74	Mg_ppm	North Minor	15,018	2,644	1.5	2,644	1.53	Nb_ppm	North Minor	104	60	0.3	60	0.27	Cs_ppm	North Minor	612	266	0.7	266	0.68	Li2O_ppm	North Upper	6,572	1,284	1.2	1,241	1.09	Ta2O5_ppm	North Upper	553	168	0.9	167	0.88	Rb2O_ppm	North Upper	11,621	3,021	1.0	3,006	1.00	K_ppm	North Upper	87,187	25,278	1.0	25,201	1.04	Fe_ppm	North Upper	91,781	16,587	1.6	16,491	1.61	Mg_ppm	North Upper	43,706	6,073	2.0	6,041	1.98	Nb_ppm	North Upper	294	64	1.0	63	0.93	Cs_ppm	North Upper	2,984	453	1.6	411	1.17	Li2O_ppm	North Upper HG	44,654	13,208	0.9	13,102	0.84	Ta2O5_ppm	North Upper HG	747	175	2.5	148	0.90	Rb2O_ppm	North Upper HG	8,862	3,057	0.7	3,030	0.69	K_ppm	North Upper HG	79,228	23,400	0.7	23,244	0.72	Fe_ppm	North Upper HG	46,838	8,990	1.1	8,526	0.77	Mg_ppm	North Upper HG	17,630	1,575	2.9	1,322	1.91	Nb_ppm	North Upper HG	243	63	0.9	61	0.74	Cs_ppm	North Upper HG	1,535	432	0.8	420	0.62	Li2O_ppm	North Upper Spur	15,609	3,374	1.2	3,272	1.13	Ta2O5_ppm	North Upper Spur	256	112	0.7	111	0.67	Rb2O_ppm	North Upper Spur	2,162	875	0.6	864	0.62	K_ppm	North Upper Spur	24,100	7,390	0.7	7,270	0.68	Fe_ppm	North Upper Spur	87,991	24,697	1.4	24,657	1.40	Mg_ppm	North Upper Spur	40,577	10,319	1.5	10,292	1.54	Nb_ppm	North Upper Spur	112	49	0.7	49	0.65	Cs_ppm	North Upper Spur	422	163	0.7	163	0.64	Li2O_ppm	South Lower	20,000	9,709	0.92	9,641	0.91	Ta2O5_ppm	South Lower	863	123	1.41	100	0.67	Li2O_ppm	South Upper	18,000	5,932	0.96	5,930	0.96	Ta2O5_ppm	South Upper	331	118	0.68	118	0.68
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K_ppm	North HW	33,566	13,372	0.7	13,332	0.73																																																																																																																																																																																																																																																																																																																																																																																																																																							
Fe_ppm	North HW	79,473	11,162	1.7	10,942	1.67																																																																																																																																																																																																																																																																																																																																																																																																																																							
Mg_ppm	North HW	34,420	3,620	2.6	3,177	2.24																																																																																																																																																																																																																																																																																																																																																																																																																																							
Nb_ppm	North HW	143	58	0.7	56	0.52																																																																																																																																																																																																																																																																																																																																																																																																																																							
Cs_ppm	North HW	535	137	1.1	135	1.07																																																																																																																																																																																																																																																																																																																																																																																																																																							
Li2O_ppm	North Lower	36,955	8,423	1.2	8,358	1.17																																																																																																																																																																																																																																																																																																																																																																																																																																							
Ta2O5_ppm	North Lower	710	195	1.1	184	0.76																																																																																																																																																																																																																																																																																																																																																																																																																																							
Rb2O_ppm	North Lower	8,137	2,390	0.8	2,360	0.74																																																																																																																																																																																																																																																																																																																																																																																																																																							
K_ppm	North Lower	60,479	19,204	0.7	19,017	0.68																																																																																																																																																																																																																																																																																																																																																																																																																																							
Fe_ppm	North Lower	82,207	12,216	1.5	12,104	1.46																																																																																																																																																																																																																																																																																																																																																																																																																																							
Mg_ppm	North Lower	39,912	4,179	2.1	4,116	2.10																																																																																																																																																																																																																																																																																																																																																																																																																																							
Nb_ppm	North Lower	159	59	0.7	57	0.56																																																																																																																																																																																																																																																																																																																																																																																																																																							
Cs_ppm	North Lower	1,250	355	0.9	340	0.70																																																																																																																																																																																																																																																																																																																																																																																																																																							
Li2O_ppm	North Minor	14,246	4,260	1.2	4,254	1.24																																																																																																																																																																																																																																																																																																																																																																																																																																							
Ta2O5_ppm	North Minor	241	132	0.4	132	0.44																																																																																																																																																																																																																																																																																																																																																																																																																																							
Rb2O_ppm	North Minor	5,693	2,211	0.7	2,194	0.65																																																																																																																																																																																																																																																																																																																																																																																																																																							
K_ppm	North Minor	57,145	19,282	0.8	19,215	0.77																																																																																																																																																																																																																																																																																																																																																																																																																																							
Fe_ppm	North Minor	27,068	8,547	0.7	8,547	0.74																																																																																																																																																																																																																																																																																																																																																																																																																																							
Mg_ppm	North Minor	15,018	2,644	1.5	2,644	1.53																																																																																																																																																																																																																																																																																																																																																																																																																																							
Nb_ppm	North Minor	104	60	0.3	60	0.27																																																																																																																																																																																																																																																																																																																																																																																																																																							
Cs_ppm	North Minor	612	266	0.7	266	0.68																																																																																																																																																																																																																																																																																																																																																																																																																																							
Li2O_ppm	North Upper	6,572	1,284	1.2	1,241	1.09																																																																																																																																																																																																																																																																																																																																																																																																																																							
Ta2O5_ppm	North Upper	553	168	0.9	167	0.88																																																																																																																																																																																																																																																																																																																																																																																																																																							
Rb2O_ppm	North Upper	11,621	3,021	1.0	3,006	1.00																																																																																																																																																																																																																																																																																																																																																																																																																																							
K_ppm	North Upper	87,187	25,278	1.0	25,201	1.04																																																																																																																																																																																																																																																																																																																																																																																																																																							
Fe_ppm	North Upper	91,781	16,587	1.6	16,491	1.61																																																																																																																																																																																																																																																																																																																																																																																																																																							
Mg_ppm	North Upper	43,706	6,073	2.0	6,041	1.98																																																																																																																																																																																																																																																																																																																																																																																																																																							
Nb_ppm	North Upper	294	64	1.0	63	0.93																																																																																																																																																																																																																																																																																																																																																																																																																																							
Cs_ppm	North Upper	2,984	453	1.6	411	1.17																																																																																																																																																																																																																																																																																																																																																																																																																																							
Li2O_ppm	North Upper HG	44,654	13,208	0.9	13,102	0.84																																																																																																																																																																																																																																																																																																																																																																																																																																							
Ta2O5_ppm	North Upper HG	747	175	2.5	148	0.90																																																																																																																																																																																																																																																																																																																																																																																																																																							
Rb2O_ppm	North Upper HG	8,862	3,057	0.7	3,030	0.69																																																																																																																																																																																																																																																																																																																																																																																																																																							
K_ppm	North Upper HG	79,228	23,400	0.7	23,244	0.72																																																																																																																																																																																																																																																																																																																																																																																																																																							
Fe_ppm	North Upper HG	46,838	8,990	1.1	8,526	0.77																																																																																																																																																																																																																																																																																																																																																																																																																																							
Mg_ppm	North Upper HG	17,630	1,575	2.9	1,322	1.91																																																																																																																																																																																																																																																																																																																																																																																																																																							
Nb_ppm	North Upper HG	243	63	0.9	61	0.74																																																																																																																																																																																																																																																																																																																																																																																																																																							
Cs_ppm	North Upper HG	1,535	432	0.8	420	0.62																																																																																																																																																																																																																																																																																																																																																																																																																																							
Li2O_ppm	North Upper Spur	15,609	3,374	1.2	3,272	1.13																																																																																																																																																																																																																																																																																																																																																																																																																																							
Ta2O5_ppm	North Upper Spur	256	112	0.7	111	0.67																																																																																																																																																																																																																																																																																																																																																																																																																																							
Rb2O_ppm	North Upper Spur	2,162	875	0.6	864	0.62																																																																																																																																																																																																																																																																																																																																																																																																																																							
K_ppm	North Upper Spur	24,100	7,390	0.7	7,270	0.68																																																																																																																																																																																																																																																																																																																																																																																																																																							
Fe_ppm	North Upper Spur	87,991	24,697	1.4	24,657	1.40																																																																																																																																																																																																																																																																																																																																																																																																																																							
Mg_ppm	North Upper Spur	40,577	10,319	1.5	10,292	1.54																																																																																																																																																																																																																																																																																																																																																																																																																																							
Nb_ppm	North Upper Spur	112	49	0.7	49	0.65																																																																																																																																																																																																																																																																																																																																																																																																																																							
Cs_ppm	North Upper Spur	422	163	0.7	163	0.64																																																																																																																																																																																																																																																																																																																																																																																																																																							
Li2O_ppm	South Lower	20,000	9,709	0.92	9,641	0.91																																																																																																																																																																																																																																																																																																																																																																																																																																							
Ta2O5_ppm	South Lower	863	123	1.41	100	0.67																																																																																																																																																																																																																																																																																																																																																																																																																																							
Li2O_ppm	South Upper	18,000	5,932	0.96	5,930	0.96																																																																																																																																																																																																																																																																																																																																																																																																																																							
Ta2O5_ppm	South Upper	331	118	0.68	118	0.68																																																																																																																																																																																																																																																																																																																																																																																																																																							

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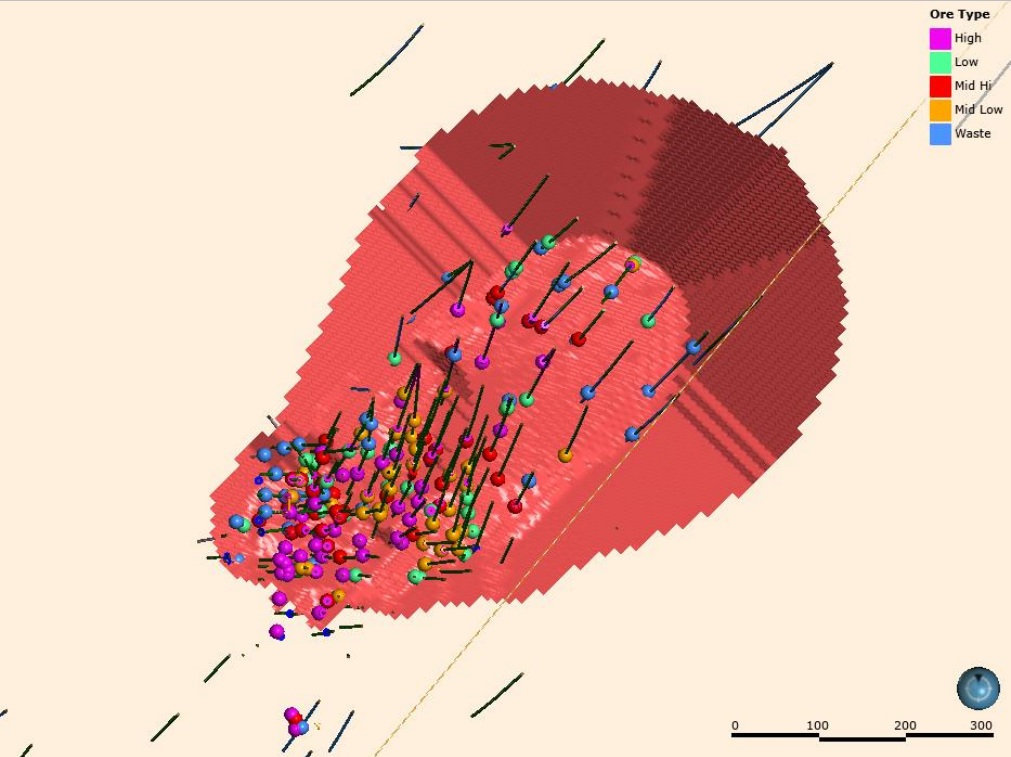
Criteria	JORC Code explanation	Commentary
		<p>Validation was carried out in several ways, including</p> <ul style="list-style-type: none"> • Visual inspection section, plan and 3D • Swath plot validation <div style="text-align: center;"> <p>North Upper HG Li2O_ppm</p> <p>Value</p> <p>Count</p> <p>Direction (45.00°, 0.00°, 0.00°)</p> </div> <ul style="list-style-type: none"> • Model vs composite statistics <div style="display: flex; justify-content: space-around;"> <div style="text-align: center;"> <p>North Upper HG - Block Model</p> </div> <div style="text-align: center;"> <p>North Upper HG - Composites</p> </div> </div>

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Criteria	JORC Code explanation	Commentary																																												
		<div style="text-align: center;"> <h3>North Upper HG</h3> <p>Ta2O5_ppm</p> </div> <ul style="list-style-type: none"> <div style="display: flex; justify-content: space-around;"> <div style="text-align: center;"> <h4>North Upper HG Model</h4> <p>Ta2O5_ppm</p> <table border="1"> <thead> <tr> <th>Stat</th> <th>Value</th> </tr> </thead> <tbody> <tr><td>Cut_Ta2O5_ppm</td><td>22.8</td></tr> <tr><td>Normal Mean</td><td>100.0</td></tr> <tr><td>Normal Stdev</td><td>50.0</td></tr> <tr><td>Std. Dev.</td><td>50.0</td></tr> <tr><td>Std. Error</td><td>15.8</td></tr> <tr><td>95% CI Lower</td><td>68.2</td></tr> <tr><td>95% CI Upper</td><td>131.8</td></tr> <tr><td>N</td><td>100</td></tr> <tr><td>Mean</td><td>100.0</td></tr> <tr><td>Max</td><td>200.0</td></tr> </tbody> </table> </div> <div style="text-align: center;"> <h4>North Upper HG Composites</h4> <p>Ta2O5_ppm</p> <table border="1"> <thead> <tr> <th>Stat</th> <th>Value</th> </tr> </thead> <tbody> <tr><td>Cut_Ta2O5_ppm</td><td>6.0</td></tr> <tr><td>Normal Mean</td><td>100.0</td></tr> <tr><td>Normal Stdev</td><td>50.0</td></tr> <tr><td>Std. Dev.</td><td>50.0</td></tr> <tr><td>Std. Error</td><td>15.8</td></tr> <tr><td>95% CI Lower</td><td>68.2</td></tr> <tr><td>95% CI Upper</td><td>131.8</td></tr> <tr><td>N</td><td>100</td></tr> <tr><td>Mean</td><td>100.0</td></tr> <tr><td>Max</td><td>200.0</td></tr> </tbody> </table> </div> </div> No reconciliation data is available. 	Stat	Value	Cut_Ta2O5_ppm	22.8	Normal Mean	100.0	Normal Stdev	50.0	Std. Dev.	50.0	Std. Error	15.8	95% CI Lower	68.2	95% CI Upper	131.8	N	100	Mean	100.0	Max	200.0	Stat	Value	Cut_Ta2O5_ppm	6.0	Normal Mean	100.0	Normal Stdev	50.0	Std. Dev.	50.0	Std. Error	15.8	95% CI Lower	68.2	95% CI Upper	131.8	N	100	Mean	100.0	Max	200.0
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Mean	100.0																																													
Max	200.0																																													
Moisture	<ul style="list-style-type: none"> Whether the tonnages are estimated on a dry basis or with natural moisture, and the method of determination of the moisture content. 	<ul style="list-style-type: none"> Tonnages are estimated on a dry basis 																																												

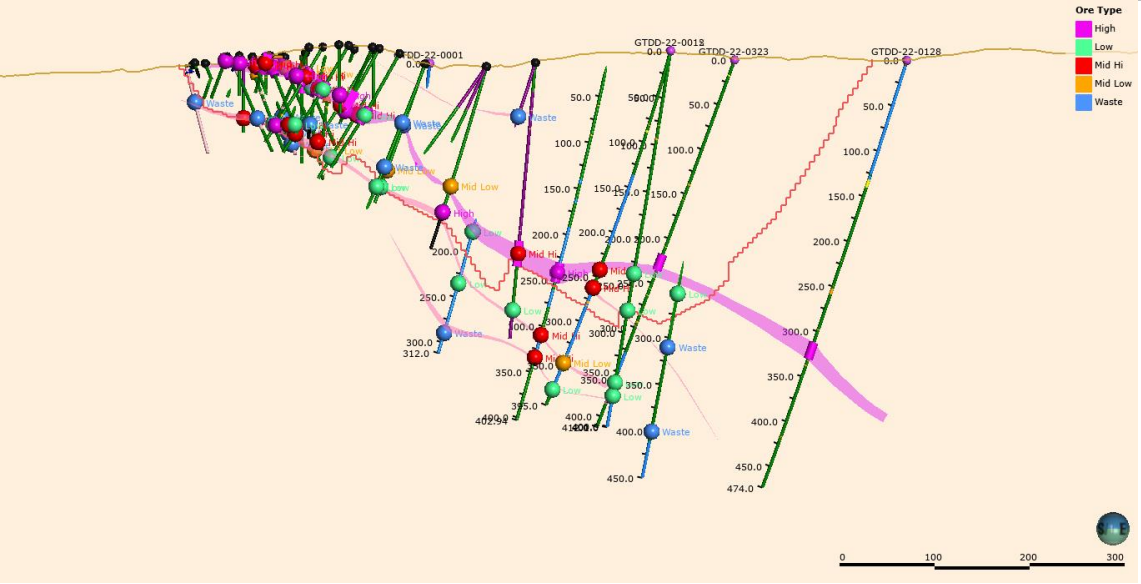
Criteria	JORC Code explanation	Commentary
Cut-off parameters	<ul style="list-style-type: none"> The basis of the adopted cut-off grade(s) or quality parameters applied. 	<p>The Seymour Mineral Resource is reported using open-pit mining constraints.</p> <p>The open-pit Mineral Resource is only the portion of the resource that is constrained within a US\$4,000 / t SC6 optimised shell and above a 0.2% Li₂O cut-off grade. The optimised open pit shell was generated using:</p> <ul style="list-style-type: none"> \$4/t mining cost \$15.19/t processing costs Mining loss of 5% with no mining dilution 55 degree pit slope angles 75% Product Recovery
Mining factors or assumptions	<ul style="list-style-type: none"> Assumptions made regarding possible mining methods, minimum mining dimensions and internal (or, if applicable, external) mining dilution. It is always necessary as part of the process of determining reasonable prospects for eventual economic extraction to consider potential mining methods, but the assumptions made regarding mining methods and parameters when estimating Mineral Resources may not always be rigorous. Where this is the case, this should be reported with an explanation of the basis of the mining assumptions made. 	<ul style="list-style-type: none"> The 2022 Mineral Resource Estimate is reported above 0.2% Li₂O cut-off. The cut-off is based on lowest potential grade at which a saleable product might be extracted using a conventional DMS and / or flotation plant and employing a TOMRA Xray sorter (or equivalent) on the plant feed. A number of pegmatites outcrop at surface thus the mineral resource is likely to be extracted using a conventional drill and blast, haul and dump mining fleet.
Metallurgical factors or assumptions	<ul style="list-style-type: none"> The basis for assumptions or predictions regarding metallurgical amenability. It is always necessary as part of the process of determining reasonable prospects for eventual economic extraction to consider potential metallurgical methods, but the assumptions regarding metallurgical treatment processes and parameters made when reporting Mineral Resources may not always be rigorous. Where this is the case, this should be reported with an explanation of the basis of the metallurgical assumptions made. 	<p>Ardiden undertook preliminary metallurgical sampling in 2017 as follows:</p> <p>Metallurgical Dense Media Cyclone Separation (DMCS):</p> <ul style="list-style-type: none"> Ardiden's Chinese strategic partner, Yantai Jinyuan Mining Machinery Co. Ltd., carried out metallurgical test work on a 2,500 kg bulk sample collected using a large rock breaker and excavator from a trench at Aubry North. Using this equipment to collect the bulk sample avoided problems associated with natural particle size distribution (PSD) as a result of drilling and blasting. After mining, the large rocks were hand broken and homogenised, then using a number of highly controlled staged crushing and sample preparation procedures Yantai generated a 500 kg sample of <6 mm particles, which is a typical size range for lithium chemical plants. The crushed head sample size distribution of the particles achieved was 86.5% ranging from 0.5 mm to 6.0 mm at an average head grade of 1.37% Li₂O and 13.5% of the particles <0.5mm at an average head grade of 0.84% Li₂O. The overall average head grade was 1.29% Li₂O. The coarse particle size used for the testing showed that crushed ore, without using a roller crusher, reduces the crushing and processing times and costs while still producing a high quality marketable lithium concentrate. Heavy Liquid Separation (HLS) The HLS tests on the North Aubry pegmatite material showed that, with a heavy liquid density of 2.95g/ml, a very impressive spodumene concentrate of up to 7.04% Li₂O at a recovery of 91.6% is produced. Detailed metallurgical studies have only just begun but preliminary metallurgical test work undertaken by IMO in Perth Western Australia on behalf of Ardiden Ltd in April and again in December 2017 suggest a 2 phase Dense Media Separation processing plant may be sufficient to achieve acceptable spodumene recoveries.

Criteria	JORC Code explanation	Commentary
		<p>Dense Media Cyclone Separation (DMCS)</p> <ul style="list-style-type: none"> Dense Media Cyclone mineral separation tests were conducted under a number of different operating conditions on the 0.5mm to 6mm size fraction only. This testwork indicated that a lithium concentrate grade of 6.05% Li₂O can be achieved at a recovery rate of 85.6%. It was noted that should it be needed, the lithium concentrate grades can be improved with an increase in feed pressure but at the cost of recovery rate. Different medium densities were also tested with one lithium concentrate producing a grade of 6.92% Li₂O with a strong recovery rate of 81.7%. The most encouraging results occurred when using: <ul style="list-style-type: none"> Feeding density of 2400kg/m; Ore feeding pressure 0.045Mpa; Ratio of ore and medium at 1:6; and Feed size of 0.5mm to 6.0mm. <p>Metallurgical Testwork Conclusions</p> <p>Ardiden concluded the following from their testwork:</p> <ul style="list-style-type: none"> It was concluded from this initial metallurgical testwork that gravity separation is a viable method of producing a high grade commercial lithium concentrate from Seymour Lake pegmatites. The North Aubry spodumene concentrate quality appears to contain only traces amounts of deleterious minerals. The North Aubry spodumene appears to have a low iron content which will positively impact down-stream processing hence enhancing the commercial value of the lithium concentrate produced. HLS tests produced high-grade lithium concentrates up to 7.04% Li₂O at a recovery rate of 91.6%. <p>Green Technology Metals Ltd</p> <ul style="list-style-type: none"> GT1 has undertaken a comprehensive metallurgical sampling and testing program for North Aubry using historic and recent drill core from 60 holes distributed throughout the mineral resource and within the optimised pit boundaries, see figure below.

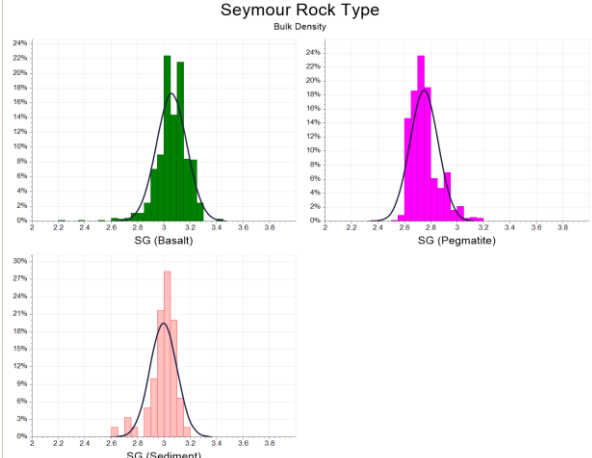
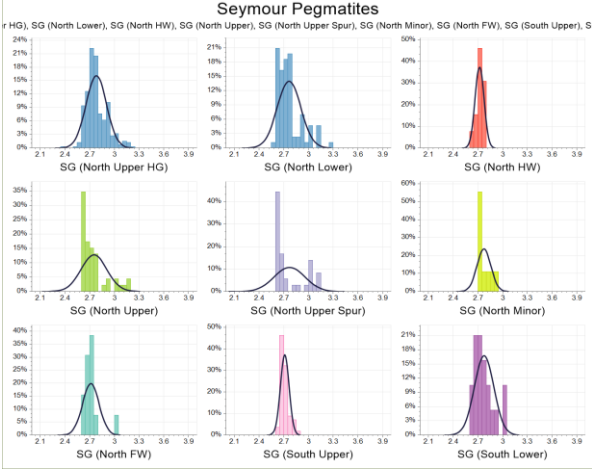
Criteria	JORC Code explanation	Commentary
		 <p data-bbox="762 946 1499 971">Plan View of Metallurgical sample locations within the optimised pit boundaries</p>

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Criteria	JORC Code explanation	Commentary																								
		 <p>Cross section through Deposit with Metallurgical Samples Selected</p> <p>Samples were selected from ¼ NQ₂ or BTW drill core and the entire pegmatite interval assigned to one of four composite groups. The groups represent the different mineralisation phases seen in the pegmatite within, what GT1 expect to be, the final open pit mine shell at Seymour.</p> <p>Results are tabulated below:</p> <table border="1" data-bbox="709 987 1207 1221"> <thead> <tr> <th>Group</th> <th>Kg's</th> <th>Proportion of Total</th> <th>Avg. Li₂O ppm</th> </tr> </thead> <tbody> <tr> <td>Mid Low</td> <td>255</td> <td>26%</td> <td>9,559</td> </tr> <tr> <td>Low</td> <td>149</td> <td>15%</td> <td>5,397</td> </tr> <tr> <td>Mid Hi</td> <td>324</td> <td>33%</td> <td>13,431</td> </tr> <tr> <td>High</td> <td>240</td> <td>25%</td> <td>19,889</td> </tr> <tr> <td>Total</td> <td>968</td> <td>100%</td> <td>12,779</td> </tr> </tbody> </table> <p>Heavy liquid separation (HLS) test work was undertaken at SGS Canada's facilities in Lakefield, Ontario. SGS compiled a master composite, based on the four samples selected, to provide a preliminary indication of the lithium beneficiation performance by dense media separation (DMS), DMS crush size, and requirement for a back-end flotation plant.</p> <p>Below summary table of the 8 mm HLS test results to achieve an HLS concentrate grade of 6.0% Li₂O and 1.2 % Fe₂O₃. The resulting global HLS recovery, post magnetic separation including losses to HLS fines bypass, is 73%.</p>	Group	Kg's	Proportion of Total	Avg. Li ₂ O ppm	Mid Low	255	26%	9,559	Low	149	15%	5,397	Mid Hi	324	33%	13,431	High	240	25%	19,889	Total	968	100%	12,779
Group	Kg's	Proportion of Total	Avg. Li ₂ O ppm																							
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		<table border="1"> <thead> <tr> <th colspan="3">HLS on master composite - crush size 8 mm</th> <th colspan="2">Grade</th> <th colspan="2">Distribution</th> </tr> <tr> <th>Description</th> <th>HLS SG</th> <th>Mass Distribution</th> <th>% Li₂O</th> <th>% Fe₂O₃</th> <th>Li₂O</th> <th>Fe₂O₃</th> </tr> </thead> <tbody> <tr> <td>HLS non-mag¹ product</td> <td>2.82</td> <td>12.7</td> <td>6.0</td> <td>1.2</td> <td>72.6</td> <td>19.2</td> </tr> <tr> <td>HLS mag¹ product</td> <td>2.82</td> <td>3.6</td> <td>1.3</td> <td>9.1</td> <td>4.5</td> <td>34.7</td> </tr> <tr> <td>HLS tailings</td> <td>-2.65</td> <td>46.7</td> <td>0.0</td> <td>0.2</td> <td>2.0</td> <td>9.3</td> </tr> <tr> <td>Flotation feed (fines and middlings)</td> <td>-</td> <td>37.0</td> <td>0.6</td> <td>0.9</td> <td>20.9</td> <td>36.9</td> </tr> <tr> <td>Total feed</td> <td>-</td> <td>100</td> <td>1.1</td> <td>0.9</td> <td>100</td> <td>100</td> </tr> </tbody> </table> <p>¹ Dry magnetic separator used for HLS testwork</p> <p>HLS and assay test work on 8mm feed (12mm & 10mm samples being QAQC and written up)</p> <p>Bulk sample pilot test work has been approved by MNDM and sampling is planned to begin in January 2023 where we will be extracting approximately ninety-nine tonnes from the North Aubry pegmatite outcrop within the Seymour project. The spodumene concentrate produced through the pilot work will be used as feed stock for the Lithium Hydroxide Conversion Program (LHCP), all forming part of the ongoing feasibility study.</p>	HLS on master composite - crush size 8 mm			Grade		Distribution		Description	HLS SG	Mass Distribution	% Li ₂ O	% Fe ₂ O ₃	Li ₂ O	Fe ₂ O ₃	HLS non-mag ¹ product	2.82	12.7	6.0	1.2	72.6	19.2	HLS mag ¹ product	2.82	3.6	1.3	9.1	4.5	34.7	HLS tailings	-2.65	46.7	0.0	0.2	2.0	9.3	Flotation feed (fines and middlings)	-	37.0	0.6	0.9	20.9	36.9	Total feed	-	100	1.1	0.9	100	100
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Environmental factors or assumptions	<ul style="list-style-type: none"> Assumptions made regarding possible waste and process residue disposal options. It is always necessary as part of the process of determining reasonable prospects for eventual economic extraction to consider the potential environmental impacts of the mining and processing operation. While at this stage the determination of potential environmental impacts, particularly for a greenfields project, may not always be well advanced, the status of early consideration of these potential environmental impacts should be reported. Where these aspects have not been considered this should be reported with an explanation of the environmental assumptions made. 	<ul style="list-style-type: none"> Waste rock characterization work has begun but was not far enough advanced for inclusion in this report at the time of writing. Some sulphur results are available from assaying of diamond core with low level Sulphur haloing observed several metres adjacent to the pegmatite contacts. More exhaustive waste rock testing is currently underway in order to be able to characterise all the waste rock types and their likely environmental impacts. Diamond core samples over the entire North Aubry deposit on a semi regular grid have been selected and submitted for multi-elemental, including Nickel and Sulphur, testwork to Actlabs in Thunder Bay Ontario. 																																																	
Bulk density	<ul style="list-style-type: none"> Whether assumed or determined. If assumed, the basis for the assumptions. If determined, the method used, 	<ul style="list-style-type: none"> 1, 518 density measurements exist in the database of which 226 are from recent water immersion testwork undertaken by Actlabs Thunder Bay Ontario on ½ NQ core samples with intervals consistent with the assay intervals submitted to the laboratory (nominally 1m). 1181 results are from laboratory pycnometer tests and the remainder are unrecorded. No obvious bias was noted between the measurements based on method, however samples whose test method was not recorded 																																																	

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	<p>whether wet or dry, the frequency of the measurements, the nature, size and representativeness of the samples.</p> <ul style="list-style-type: none"> The bulk density for bulk material must have been measured by methods that adequately account for void spaces (vugs, porosity, etc), moisture and differences between rock and alteration zones within the deposit. Discuss assumptions for bulk density estimates used in the evaluation process of the different materials. 	<p>were excluded from the data analysis process. These were typically older samples with unknown test conditions applied.</p> <ul style="list-style-type: none"> Previous mineral resource estimates have determined pegmatite bulk densities of 2.78 and country rock, mainly meta-basalts, to be approximately 3.0. 698 density measurement are within the interpreted pegmatite boundaries the bulk within the North Upper HG domain. This domain confirmed previous bulk density values of 2.78. Fresh waste rocks averaged 3.0 consistent with basalt and sediment averages. No bulk density data is available for the largely glacial cover over the deposit due to the difficulty in recovering this material in the drilling process. This material is volumetrically negligible ranging in depths from 0 to 14m and averaging around 3m. An assumed bulk density of 2.2 was used for overburden.  <ul style="list-style-type: none"> 
<p>Classification</p>	<ul style="list-style-type: none"> The basis for the classification of the Mineral Resources into varying confidence categories. Whether appropriate account has been taken of all relevant 	<ul style="list-style-type: none"> The Mineral Resources have been classified as Indicated and Inferred based on drill spacing and geological continuity and modifying factor confidence levels The Resource model uses a classification scheme based upon drill hole spacing plus block estimation parameters, including kriging variance, number of composites in search ellipsoid informing the block cell and average distance of data to block centroid.

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	<p>factors (ie relative confidence in tonnage/grade estimations, reliability of input data, confidence in continuity of geology and metal values, quality, quantity and distribution of the data).</p> <ul style="list-style-type: none"> Whether the result appropriately reflects the Competent Person's view of the deposit. 	<ul style="list-style-type: none"> The results of the Mineral Resource Estimation reflect the views of the Competent Person.
Audits or reviews	<ul style="list-style-type: none"> The results of any audits or reviews of Mineral Resource estimates. 	<ul style="list-style-type: none"> No audits or reviews have been undertaken by GT1
Discussion of relative accuracy/confidence	<ul style="list-style-type: none"> Where appropriate a statement of the relative accuracy and confidence level in the Mineral Resource estimate using an approach or procedure deemed appropriate by the Competent Person. For example, the application of statistical or geostatistical procedures to quantify the relative accuracy of the resource within stated confidence limits, or, if such an approach is not deemed appropriate, a qualitative discussion of the factors that could affect the relative accuracy and confidence of the estimate. The statement should specify whether it relates to global or local estimates, and, if local, state the relevant tonnages, which should be relevant to technical and economic evaluation. Documentation should include assumptions made and the procedures used. These statements of relative accuracy and confidence of the estimate should be compared with production data, where available. 	<ul style="list-style-type: none"> The relative accuracy of the Mineral Resource is reflected in the reporting of the Mineral Resource as being in line with the guidelines of the 2012 JORC Code. The statement relates to local estimates of tonnes and grade, with reference made to resources above a certain cut-off that are intended to assist mining studies.