# Identifying and characterising potential pathfinder elements to IOCG style mineralisation on Central Yorke Peninsula

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## TITLE

Identifying and characterising potential pathfinder elements to IOCG style mineralisation on Central-Northern Yorke Peninsula

#### **RUNNING TITLE**

Identifying IOCG pathfinder elements: Yorke Peninsula

## ABSTRACT

The central Yorke Peninsula in the southern Gawler Craton, South Australia forms the Southern extent of the Olympic Copper-Gold Province. The area is known for its iron oxide-copper-gold (IOCG) style mineralisation, yet the pathfinder elements associated with mineralisation and the processes and/or lithological controls effecting the distribution of potential pathfinder elements within mineral phases are poorly understood. Analyses of whole rock geochemical data from basement lithologies identified Au, As, Ag, Bi, Ce, Cu, La, Mo, S and W as pathfinder elements towards potential (IOCG) mineralisation in the area. Mineral phases that preserved these trace element signatures were investigated with SEM back scatter analysis and Laser-Ablation Inductively-Coupled Mass Spectroscopy (LA-ICP-MS) analysis. SEM analyses determined LREE's are host within bastnäsite, which has a documented occurrence across Yorke Peninsula. Sulphide phases were identified to host elements Ag, As, Au, Bi, Cu, Mo and S. Within pyrite, pathfinder elements are preferentially hosted within inclusion rich areas. Chalcopyrite was determined to host concentrations of Au. A prospectivity index was created for central Yorke Peninsula by combining identified pathfinder elements to create an IOCG index for representative samples. The index highlights an area proximal to the township of Alford as a potential hotspot for mineralisation/exploration within central Yorke Peninsula.

## **KEYWORDS**

Iron oxide-copper-gold (IOCG), Whole rock geochemistry, Geochemical vectors, Mineral hosts, Pathfinder elements, Central Yorke Peninsula

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#### INTRODUCTION

Anomalous concentrations of elements that are linked to hydrothermal alteration can be recognised in whole rock geochemical data (eg. Mark et al. (2006); Belperio et al. (2007); Fabris et al. (2013a). Iron oxide-copper-gold (IOCG) mineralisation is demonstrably associated with specific pathfinder elements such as Ag, As, Au, Ba, Bi, Ca, Cd, Ce, Co, Cr, Cs, Cu, F, Fe, K, La, Mn, Mo, Ni, S, Sb, Se, Sn, Te, U and W (Mark et al. 2006; Fabris et al. 2013a; Wang et al. 2013; Hill et al. 2014; Fabris et al. 2013b) and are often preserved within hydrothermal alteration halos that can be linked to mineralisation, and can be used as geochemical vectors towards mineralisation (e.g. Mark et al. 2006; Large and McGoldrick 1998; Wang et al. 2013; Fabris et al. 2013b). Anomalous pathfinder element concentrations may be recognisable within one or more lithologies, and may be host within different mineral species that are linked to processes of hydrothermal alteration (Ismail et al. 2014; Belperio et al. 2007)

The Olympic Copper-Gold Province within the Gawler Craton is host to IOCG mineralisation including the Olympic Dam, Prominent Hill, Carapateena, and Hillside deposits and the historic Moonta/Wallaroo District (Fig. 1). IOCG mineralisation throughout the Gawler Craton has been temporally (Reid et al. 2011) and genetically Conor et al. (2010) linked to the shallow-intrusives of the ca. 1600-1580 Ma Hiltaba Suite Granites (e.g. Reid et al. 2011; Fanning et al. 2007; Morales-Ruano et al. 2002; Ferris et al. 2002; Skirrow et al. 2002). The geochemical signature of mineralisation associated with the IOCGs in the Gawler Craton Olympic Province and other known IOCGs from around the world have been described in detail, and are attributed to hydrothermal alteration during the mineralising event (Groves et al. 2010; Skirrow et al. 2002; Hitzman 2000; Belperio et al. 2007). The alteration halos are generally

characterised into a number of discrete alteration shells corresponding to elemental distributions. As an example, the alteration signature may progress from regional Naenriched, K-depleted alteration (Porter 2010) to camp-scale Mn, K, and Ba enriched and Na-depleted alteration to deposit scale alteration enriched in Ag, Au, Bi, Cu, Fe, K, Mo, Sb and U (Ernest Henry:(Mark et al. 2006). Mineralisation within the Moonta Wallaroo District on Yorke Peninsula in the southern Gawler Craton (Figs. 1 & 2) is host within the Palaeoproterozoic Wallaroo Group (e.g. Conor et al. 2010; Zang 2002; Conor 2002), and is considered part of the major IOCG mineralisation event throughout the Gawler Craton (Conor et al. 2010; Both et al. 1993; Reid et al. 2011). However, the pathfinder elements associated with mineralisation in this region and the processes and/or lithological controls affecting the distribution of potential pathfinder elements within mineral phases are poorly understood. The links between whole rock geochemical analysis, element deployment, hydrothermal alteration and mineralisation processes is then discussed and criteria for mineral exploration given.



Figure 1 - Geological map showing the Olympic Domain within the Galwer Craton and location of major IOCG mineralisation in the area. The location of the Yorke Peninsula area is also shown in boxed area. Modified after Conor et al. (2010)



Figure 2 - Interpreted solid geology map of Northern Yorke Peninsula highlighting the Moonta-Wallaroo and Hillside mineralised areas and associated alteration within and around these deposits. Locality of samples chosen for analysis highlighted by coloured circles. Location shown on figure 1. Modified after Conor at al. (2010)

#### **GEOLOGICAL SETTING/BACKGROUND**

The central Yorke Peninsula in the southern Gawler Craton, South Australia (Fig. 1) comprises the ~1.75 Ga Wallaroo Group. The Wallaroo Group is dominated by metasedimentary rocks comprising siltstones, calcsilicates, psammites, quartzites and iron-rich sediments of the Wandearah Formation (Conor 1995; Cowley et al. 2003; Zang 2002). Felsic volcanics of the Weetulta Formation and mafic volcanics and amphibolites of the Matta Formation were contemporaneously emplaced with deposition of the metasediments (Cowley et al. 2003).

The Wallaroo Group has undergone metamorphism and multiple stages of deformation. Metamorphic grades range from upper greenschist in the northern Yorke Peninsula to mid amphibolite facies in the central Yorke Peninsula (Conor et al. 2010; Conor 2002; Ferris et al. 2002). The timing of metamorphism of the Wallaroo Group is currently unconstrained. Deformation involved initial generation of isoclinal folds that are suggested to have developed either during the ca. 1730-1690 Ma Kimban Orogeny (Hoek and Schaefer 1988; Ferris et al. 2002; Hand et al. 2007) or during the ca. 1600-1580 Ma Olarian Orogeny (Conor et al. 2010; Conor 1995; Conor 2002). The isoclinal folds are overprinted by a second generation of upright folds that are suggested to have developed at ca. 1600-1580 Ma (Conor et al. 2010; Zang 2002; Carthew 1993) The Wallaroo Group has been intruded by the Tickera and Arthurton Granites (Fig. 2), which are equivalents to the ca. 1600-1580 Ma Hiltaba Suite Granites (e.g. Creaser and Cooper 1993; Conor 1995; Conor et al. 2010; Cowley et al. 2003). The granites vary across the A-, I- and S-type range and exhibit a considerably diverse grain size and texture (Conor et al. 2010). The granites also show evidence of variation in depth of emplacement from foliated and lineated granites emplaced at depth and non-foliated

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granites that contain miarolitic cavities and that are suggested to have been emplaced at shallow crustal levels (Conor et al. 2010). The granites are located proximal to known mineralisation and alteration in the central Yorke Peninsula (Fig. 2) (Reid et al. 2011; Morales-Ruano et al. 2002; Ferris et al. 2002). Magmatically-derived hydrothermal fluid has been suggested as a possible mechanism for intense alteration of rocks in the region, and is used as evidence for the intrusive Hiltaba Suite being linked with mineralisation and alteration of the Wallaroo Group (Conor 1995; Both et al. 1993; Jack 1917).

Mineralisation within the Wallaroo Group on Yorke Peninsula is generally copper dominated and can be separated into two main mineralisation styles. In the Moonta-Wallaroo district (Fig. 2), mineralisation is dominantly vein-style and associated with earlier developed ductile shear zones and intense regional magnetite-bearing alteration (Conor et al. 2010; Zang 2002). Mineralisation in the Moonta-Wallaroo region is hosted within the Moonta Porphyry (Weetulta Formation) and the Doora Schist (Wanderarah Formation). Mineralisation associated with the Hillside Cu Prospect in the eastern Yorke Peninsula (Fig. 2) is hosted within the Wallaroo Group metasediments (Wandearah Formation), and is controlled by north-trending steeply-dipping structures that are associated with the Pine Point Fault Zone (Conor et al. 2010). Mineralisation in the Hillside area appears be skarn-like and have developed higher in the crust as evidenced by hematite replacing magnetite (Conor et al. 2010; Ismail et al. 2014) Alteration in the Moonta-Wallaroo district has been constrained to ca. 1585 Ma from dating of hematite-chlorite-carbonate-sericite mineralisation associated with Moontastyle veins (Skirrow et al. 2007), which supports the interpreted association with emplacement of the 1600-1580 Ma Hiltaba Granite Suite.

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#### **METHODS**

## Logging and sampling

This project utilises the same sample set as used by Forbes et al. (2012), collected in April-July 2012. Samples were taken from open file drill holes across the central Yorke Peninsula, and stored at the Department of State Development Glenside and Moonta Core Library Facilities. Drill holes were selected based on them preserving the intersection of basement and cover, and to give maximum spread and representation of basement lithologies of the region.

Samples taken are composite 1m intervals from the top of basement and from 10-11m beneath the basement-cover interface where possible. In total, 96 samples were taken from the top of basement, and 68 samples were taken from 10-11m below the basement-cover interface.

## Geochemistry

All 164 samples were crushed, milled and analysed by Genalysis Laboratories Pty Ltd in Adelaide, South Australia. Samples were analysed using standard XRF ICP-MS, ICP-OES and fire assay techniques. Representative data is shown in (Table. 1,2). Raw data and detail of individual element analysis are given in Appendix A. Data was used to determine background chemistry of various lithologies, to identify potential pathfinder elements that may be used as geochemical vectors towards IOCG mineralisation, and to determine how pathfinder elements partition into particular lithologies and/or minerals.

#### Petrological analysis and elemental mapping

Petrological analysis and qualitative element maps were used to identify mineral phases that host potential pathfinder elements. Twelve drill core samples were chosen for thin section preparation. Samples were chosen to cover a range of lithologies, preserving the greatest amount of alteration, and being located proximal to known IOCG mineralisation in the area. Samples were prepared to a thickness of 30µm and polished at Adelaide Petrographic Labs, South Australia. Petrographic analysis was undertaken using standard transmitted and reflected light microscopy at the University of Adelaide. Selected areas were used for qualitative element mapping using a XL40 SEM with EDAX Genesis energy dispersive X-ray analyser at Adelaide Microscopy, University of Adelaide. Mapped areas were chosen based on preservation of possible IOCG alteration and mineral phase(s) that would best show deployment of potential pathfinder elements. Maps were produced under conditions of 20.0kV and 95µA using spot size of 6.000nm, dwell time of 250ns at a resolution of 512 x 400 pixels.

#### **Mineral chemistry**

Laser-Ablation Inductively-Coupled Mass Spectroscopy (LA-ICP-MS) analysis was undertaken to quantify the concentration of potential pathfinder elements within sulphide minerals. Analysis was undertaken using an Agilent 7500cx with attached New Wave UP-213 laser ablation (LA-ICPMS) system at Adelaide Microscopy. A fluence of 4.5J/cm<sup>2</sup>, laser energy of 45%, repetition rate of 5Hz and spot size of 30µm were used. Concentrations of Fe were set using known stoichiometric concentrations within the sulphides (59.8860 wt.% for pyrite).

Table 1 - F	Representative dat	ta of major eler	ments from W	allaroo basen	ent lithologies	8			
Samp #	Lith type	Samp type	AI2O3%	CaO %	Fe2O3%	MgO %	MnO%	SiO2%	TiO2%
1946904	Amphibolite	Core	14.97	3.65	16.24	5.81	0.17	47.79	1.58
1946905	Amphibolite	Core	15.18	2.8	12.1	5.48	0.1	51.45	1.33
1937386	Carbonate	Cuttings	4.63	20	4.96	14.55	0.31	19.59	0.24
1949043	Carbonate	Drill core	7.48	29.22	2.15	4.35	0.6	24.65	0.21
1949782	Carbonate	Drill core	4.87	18.18	5.01	10.79	1.08	29.12	0.25
1949038	Felsic volcanic	Drill core	11.09	0.13	10.03	5.75	0.01	62.65	0.51
1949054	Felsic volcanic	Drill core	15.65	0.65	12.78	3.85	0.06	56.17	0.63
1960847	Felsic volcanic	Drill core	11.06	0.45	12.67	1.02	0.03	65.57	0.78
1937190	Gneiss	Drill core	14.36	3.49	6.4	2.27	0.09	64.49	0.63
1937617	Gneiss	Cuttings	15.41	2.58	10.32	1.81	0.03	56.78	2.44
1946898	Gneiss	Cuttings	13.25	1.11	3.48	0.24	0.01	70.02	0.63
1937612	Granite	Drill core	12.54	0.92	7.94	1.08	0.03	67.47	0.8
1937615	Granite	Cuttings	12.69	0.88	2.4	0.47	0.01	73.27	0.47
1946900	Granite	Cuttings	14.16	0.12	1.13	0.23	0.01	72.73	0.21
1949011	Mafic extrusive	Drill core	1.02	26.43	4.12	17.73	0.98	5.26	0.08
1949020	Mafic extrusive	Drill core	15.3	1.53	10.58	4.97	0.18	53.59	1.58
1949802	Mafic intrusive	Drill core	12.72	7.98	9	6.83	0.41	44.98	1.14
1937185	Mafic intrusive	Drill core	14.74	8.36	11.2	5.63	0.19	44.05	1.28
1949819	Mafic intrusive	Drill core	13.34	6.89	15.04	5.92	0.21	47.83	1.29
1949812	Metasomatite	Drill core	16.95	0.58	6.84	5.06	0.01	56.3	0.69
1949780	Metasomatite	Drill core	11.43	4.54	12.59	3.2	0.15	58.88	0.51
1949813	Metasomatite	Drill core	9.08	13.35	3.53	3.7	0.17	53.45	0.39
1960886	Sediment	Drill core	16.11	0.28	5.74	3.23	0.03	63.85	0.65
1949023	Sediment	Drill core	14.26	2.87	5.43	2.71	0.15	60.16	0.59
1949789	Sediment	Drill core	15.74	0.29	5.37	3.25	0.01	63.69	0.6

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Table 2 – Representative trace element concentrations for lithological groups Се Cs Hf Мо Ni S Sb Та Th υ v w Zr Samp Au Ag As Ва Со Cr Cu La Те Lith type Samp # Bi ppm type ppb ppm 11.6 1946904 Amphibolite 183.8 0.04 25 79.8 69 74 0.5 0.05 2.9 381 0.05 97 Core 13 0.35 52.4 2.8 13.7 2.1 68 128 0.16 6.08 5 1946905 Amphibolite 2 0.24 9.7 245.7 0.08 23.9 44.6 154 4.95 171 0.17 0.4 0.05 2.83 349 2 110 Core 216 3.1 12.2 1.1 65 7.14 197.2 0.59 73 1937386 Carbonate Cuttings 0.5 0.51 153.6 33.9 23.9 21 1.64 481 1.5 21.8 5.6 12 99 1.45 0.4 0.2 6.6 2.36 54 0.05 1949043 0.025 3 465.6 0.005 3.33 2 20 0.7 0.05 66 Carbonate Drill core 1 36.8 3.7 38 180 18.7 0.4 25 0.15 10.4 1.61 18 1 Drill core 1949782 Carbonate 0.5 1.11 85.8 218.1 0.32 44.6 58.1 39 1.56 216 2.5 22.9 2.1 31 8616 1.1 0.5 0.07 7.83 5.38 44 1 94 Felsic 1949038 Drill core 0.5 19.9 97.3 29.2 0.92 20.3 0.16 8.96 519 0.26 0.13 41.1 10 173 14 3.8 26 25 3.6 0.18 36.12 14 1 volcanic Felsic 1949054 6 0.09 134 Drill core 0.28 35.9 442 0.76 59.5 24.3 83 6.97 201 3.7 28.3 0.7 26 894 0.29 1.3 14.05 2.88 116 4 volcanic Felsic 280 21.0 1960847 Drill core 2 0.025 9 767.1 0.005 540.3 47.9 10 0.82 117 17.3 6.1 100 57 0.52 3.8 0.2 27.12 52 633 1 7 9 volcanic 0.02 1937190 0.5 0.025 0.5 0.03 95.6 53 2.25 7.4 48.3 25 86 1.6 0.1 10.24 2.25 77 0.05 283 Gneiss Drill core 751.7 15.4 16 1.3 5 1937617 Gneiss 0.5 0.06 0.5 102.2 0.07 296.6 25 35 1.59 76 5.4 150 0.7 14 324 0.24 1.4 0.05 13.95 8.13 432 2 206 Cuttings 136 1946898 822.3 0.9 0.07 510 Gneiss Cuttings 0.5 0.18 1.9 0.14 231.3 4.3 172 3.01 17 13.4 5 168 2.8 0.07 22.5 7.29 32 3 5 0.02 1937612 0.5 509.9 0.005 25 0.05 0.05 489 Granite Drill core 0.5 0.11 69.8 16.6 10 1.24 25 13 34.4 0.6 9 2.1 14.44 2.4 45 5 1937615 Cuttings 449.9 0.08 19.2 1.08 200 0.49 13.27 322 Granite 0.5 0.11 4.1 2.3 88 7 8.8 10.1 1.9 3 1.8 0.06 2.07 56 2 742 25.0 1946900 Granite Cuttings 0.5 0.025 5.8 618.1 0.07 1345.8 1.6 97 2.78 15 6.2 0.5 3 257 0.27 3.5 0.05 196.29 32 3 200 4 2 Mafic 1949011 Drill core 0.5 0.69 87 0.26 602 0.94 0.1 0.05 2 0.05 19 33.7 0.17 15.1 19 38 0.5 7.8 2 7 6551 2.35 16 extrusive Mafic 1949020 15.9 151.3 0.81 39.1 101 33 0.75 0.7 0.05 4.05 361 13 99 Drill core 1 0.31 21.8 0.42 554 2.8 16.7 1.4 1285 4.7 extrusive Mafic 1949802 Drill core 12 0.025 63 140.1 0.39 73.9 47.6 219 3.83 331 3 40.6 0.4 108 2064 1.18 0.6 0.05 2.28 1.65 251 11 106 intrusive Mafic 1937185 0.025 0.07 1.02 565 0.22 0.5 1.85 0.83 318 91 Drill core 0.5 6 104 14.7 32.6 265 68 2.5 7.1 0.8 61 0.1 3 intrusive Mafic 1099 1174 Drill core 32 4.33 2.6 90 6.03 0.5 1949819 3.1 44 186.3 1.49 20.9 64.1 158 9 0.7 0.1 2.56 1.67 314 3 88 5 intrusive 1 Metasomatit 1949812 0.025 368 Drill core 0.5 8 555.1 0.13 121.3 19.1 71 3.2 25 10.4 88.3 0.05 39 176 0.23 1.7 0.1 26.06 6.7 84 2 e Metasomatit 1949780 0.5 212 Drill core 0.06 16.7 14.6 0.05 37.8 10.9 58 0.44 34 5.8 12.3 0.2 24 25 0.18 1.3 0.08 35.83 7.94 123 1 e Metasomatit 1949813 Drill core 2 0.025 8 0.005 112.6 1.88 0.05 233.1 35.5 54 33 6.6 60.9 0.8 10 954 0.34 1.2 16.06 5.1 40 0.05 245 e 166 1960886 Sediment Drill core 6 0.025 6 602 0.19 69.5 18.3 72 9.79 32 4.5 35.9 0.5 56 156 0.58 1.4 0.05 17.83 3.33 153 15 1044 1949023 30.7 15 16.59 155 170 Sediment Drill core 0.5 0.55 694.1 12.55 115.7 34.1 69 3.77 545 4.4 1.63 1.5 0.19 3.45 10 4.6 56.4 9 1949789 Sediment Drill core 0.06 17.8 613.6 0.25 88.5 17.5 71 7.28 10 4.1 46.8 0.2 32 124 0.34 1.5 0.15 18.17 2.69 124 5 143 1

#### **OBSERVATIONS AND RESULTS**

#### Logging and sampling

One hundred and sixty four drill holes were logged by (Forbes 2012b) to characterise the broad lithologies across Yorke Peninsula. Lithologies logged included granites, felsic volcanics, mafic volcainics, metasediments, gneiss and metasomatite (Table. 3). General lithological characteristics such as mineralogy, grain size, alteration and the degree of weathering were documented. Logging was undertaken to cover an area that incorporated the cover/basement relationship and to a depth of 10-12m into basement lithologies if core samples were available. Samples for geochemical analysis were taken from 1m of composite core taken from either 1m into basement or 10-12m into basement lithologies. 135 samples came from ¼ or ½ core depending on availability from the core library and 29 samples were from rock chips. Further detailed analysis of each identified lithology was undertaken in this study to characterise the separate rock types and geochemical relationships.

#### GRANITES

Granites generally consist of medium to course grained quartz, k-feldspar, and plagioclase with varying small amounts of (<15%) biotite, (<5%) amphibole and minor mafics (<10%). Samples come from the Tickera and Arthurton Granites (Table. 3). The Tickera Granite is coarser grained (~30% quartz up to 30mm; ~50% K-feldspar/Plagioclase up to 50mm) with an overall pinkish granitic appearance. The Tickera Granite displays small amounts weathering. The Arthurton Granite has a finer grained (~40% quartz 5-20mm, ~45% K-feldspar/Plagioclase 5-20mm) ademellite/monzonitic composition, more whiteish pink in colour and is highly

weathered. Both granites preserve a diverse range of alteration including hematite/magnetite and epidote-chlorite alteration. Minor sulphides (<2%) as pyrite and chalcopyrite were occasionally observed throughout the granites.

#### METASEDIMENTS

Metasediments cover a large array of rock types from the Wandearah Formation (Table. 3) including sandstones, siltstones, mudstones, clays, psammites, dolomites and limestones. The metasediments can broadly be separated into a sediment and carbonate component. The dominant sedimentary rock type is siltstone with many samples appearing brecciated. The siltstone is generally laminated, light or dark grey in colour and shows variation of pigment within the greys (red/blue/green). Grainsize is fine grained ranging from (0.25mm – 1mm). The dominant carbonate rock type is dolomite which can be siliceous or preserve fine laminations. The main alteration observed within the dominant sedimentary rock units is Fe and carbonate with lesser chlorite and epidote. Oxides are commonly hematite or magnetite and small amounts of malachite. Sulphides are very fine grained within carbonate veins therefore difficult to determine composition.

#### FELSIC VOLCANICS

Felsic volcanics form part of the Weetulta Formation and include the Moonta Porphyry (Table. 3). Samples are generally porphyritic or metaporphyritic, fine to medium grained (0.25-2 mm) and pink/grey in colour. Most felsic volcanics preserve strong hematite/magnetite alteration with lesser of chlorite/potassium and sericite alteration. Fe (hematite) alteration appears to be more prominent in samples proximal to the Moonta Porphyry member.

#### MAFIC VOLCANICS

Mafic volcanics from the Matta Formation (Table. 3) are comprised of amphibolites, basalts, dolerites or diorites with minor gabbro. Dolerite is the dominant rock type within the rock unit. Mineralogy is generally (~50% plagioclase, ~40% pyroxene/amphibole  $\pm$  biotite). Samples are very fine to medium grained (0.25-3mm), show euhedural to subhedural grain shape and dark green in colour. Carbonate veining is seen throughout samples with minor chlorite/epidote alteration. Fe staining and alteration is consistently seen throughout samples. Pyrite and chalcopyrite are present as disseminated sulphides and within carbonate and quartz veins.

#### GNIESS

Gneisses are dominantly comprised of (~15%) potassium-feldspar, (~20%) quartz and (~15%) plagioclase. Samples contain variable amounts of hornblende (20-70%) or biotite (20-50%) and depending on which mineral is most dominant the gneisses is categorised as either a hornblende or biotite bearing gneiss Samples are generally fine to coarse grained, pinkish coloured and with varying amounts of weathering seen across samples. Pegmatite veining up to ~10mm is common through samples. Alteration is restricted to Fe alteration. No sulphides were observed in the samples.

#### METASOMATITE

The metasomatite samples are from the Oorlano Metasomatite (Table. 3). Mineralogy is generally (~25%) quartz; ~1mm, (~50%) k-feldspar; 1-2mm and (~25%) mafics which are comprised of biotite-magnetite-amphibole. Most samples show a striking green/red

banded appearance. The red/pink altered layer shows feldspar/ potassic/albitic alteration and green layer is dominated by quartz and feldspar and green appearance possibly due to epidote/chorite/amphibolite alteration. Carbonate veining up to 10mm is common throughout samples. Fe alteration is seen as hematite and magnetite and minor sulphides (<2%) as pyrite and chalcopyrite. Logging data determined that samples generally altered siltstones or calcsilicates from the metasediments but determining the true protolith of the metasomatite is difficult to determine as the formation has been so heavily altered. Table 3 - Stratigraphy of the northern Yorke Peninsula region from (Forbes 2012a) modified after Cowley et al. (2003) with additional information from (Schwarz 2003), (Reid et al. 2008) and Conor et al. (2010).

Group/Suite	Formation	Member	Lithological description					
	Magnetite alteration?		Albite-magnetite ± calcsilicate ± biotite					
Alteration	Oorlano Metasomatite	1	Skarn-like metasomatic rock. Calcsilicate (feldspar, scapolite, actinolite, diopside, carbonate, phlogopite, epidote, magnetite, pyrite), feldspathic (albite>>microcline, lesser opaques and calcsilicate minerals) and iron-rich (magnetite>>hematite, feldspar, biotite, pyrite) metasomatite; precursor lithology usually obliterated or uncertain. Local kaolin alteration (± siderite ± alunite ± pyrite ± chalcocite) [late phase of metasomatism??]					
taba Suite 00-1580 Ma)	Tickera Granite		Variable I- and S-type granitoids; monzogranite, quartz monzonite, leucotonalite, commonly intensely deformed					
	Arthurton Granite		A-type granite, monzogranite, quartz monzonite, generally undeformed					
Hil (ca. 16	Curramulka Gabbronorite		Gabbronorite comprising plagioclase, clinopyroxene (augite), orthopyroxene, hornblende, biotite ± quartz or orthoclase.					
Hiltaba Suite?	Bute Metadolerite		Metadolerite comprising albitised and sericitised plagioclase laths, amphibole, biotite and/or chlorite replacing interstitial pyroxene, accessory epidote, opaques, carbonate, sericite and sphene, ophitic texture. Mostly massive, locally sheared. Locally metagabbro with feldspar glomerophenocrysts.					
ives	Wandearah Formation		Clastic and chemical metasediments.					
		Wokurna Member	Dominantly laminated muscovite-bearing argillite interlayered with quartz siltstone, occasionally graded, minor iron-oxide, carbonate, albite and calcsilicate, pale green and buff reduction spotting common. Locally becomes more siltstone- to sandstone-rich with disseminated magnetite and hematite as accessory phases; or chloritic metasediments, carbonaceous or calcareous. Includes thin calcsilicate, carbonate and albitic units.					
c intrus		Delken Member	Cherty', finely laminated, layer parallel albite-quartz metasediment, trace amphibole, magnetite and calcite.					
ibvolcani		New Cornwall Member	Metasiltstone and chemical sediments, often layered. Variable proportions of calcsilicate, minor limestone and dolomite, laminated albitite, layered albite-magnetite ironstone, graphit and calcareous metasiltstone, argillite and metasandstone.					
Ma) r associated su		Doora Member	Schistose, medium-grained, thinly planar layered to laminated metasediments of variable composition including quartz-plagioclase rich, iron-rich and calcsilicate metasediments (e.g. quartz-amphibole (actinolite and cummingtonite)-biotite ± magnetite calcsilicate; biotite- quartz-albite pelite; quartz-albite-magnetite-biotite iron formation). Minor marble and quartz- albitite.					
1750 1 10 minor		Aagot Member	Planar to bedded micaceous metasandstone or psammite, sandy or tuffaceous argillite with minor interlayered calcsilicate and albitic rocks. Locally includes volcanoclastic conglomerate.					
up (ci	Weetulta Formation		Felsic volcanics - porphyritic rhyodacite, dacite, latite with bedded tuff and tuffaceous siltston					
Wallaroo Gro afic metavolcar		Mona Volcanics Member	Acid porphyry (dacite to rhyodacite) with plagioclase phenocrysts (altering to K-feldspar or sericite) with lesser quartz and altered ferromagnesian minerals in quartz + plagioclase or f banded K-feldspar ± chlorite ± quartz ± Fe-oxides, locally amygdaloidal and spherulitic; and thinly bedding tuffaceous siltstone to tuff with local microphenocrysts, minor crystal tuff a lapilli tuff.					
and m		Moonta Porphyry Member	Plagioclase-phyric rhyodacite locally showing volcanic or ignimbritic characteristics. Plagioclas phenocrysts altered to K-feldspar.					
s, felsic		Wardang Volcanics Member	Plagioclase-phyric rhyodacite, dacite and latite, locally rhyolite; locally preserves flow banding flow folding, columnar cooling structures and hyaloclastics.					
nent	Matta Formation		Uniferration for the first later interview and an electronic					
Metasedir	matta romation	Willamulka Metabasalt Member	Fine- to medium-grained, massive to amygdaloidal basalt, replaced by albite, chlorite, biotite, amphibole, calcite, quartz and accessory minerals. Amygdales filled with quartz, chlorite and/or carbonate.					
		Renowden Metabasalt Member	Basalt with crystals (phenocrysts?) of hornblende and plagioclase. Locally contains quartz-rick vesicles that may be scoriaceous texture.					
		Wandilta Amphibolite Member	Amphibolite and layered plagioclase-homblende-biotite rock.					
Donington Suite (ca. 1850 Ma)			Intrusives of variable composition including gabbro, gabbronorite, charnockite, and granodiorite to alkali granite.					

#### Whole Rock Geochemistry

Major element geochemical analyses was undertaken to determine lithogeochemical relationships within the data. Relationships were assessed to determine geochemical signatures that can be used to distinguish the lithologies from one another, and to determine background geochemistry. Trace element analyses was used to determine which elements preferentially partition into certain lithological groups.

#### MAJOR ELEMENT CHEMISTRY

Granites characteristically show high SiO<sub>2</sub> (>65 wt.%) and low MgO (<2wt.%), Al<sub>2</sub>O<sub>3</sub> concentrations between (11-15 wt.%) and TiO<sub>2</sub> (<0.8 wt.%) (Fig. 3a,b,c). The geochemical data for granites is consistently clustered (Fig. 3a,b,c). Felsic volcanics show similar characteristics to the granites with respect to SiO<sub>2</sub> between (65-77 wt.%) and Mg (< 2.wt%) (Fig. 3a), but show broader Ti and Al composition (Fig. 3b,c). Mafics lithologies are distinguished by their typically high TIO<sub>2</sub> contents (>0.9 wt.%) and low SiO<sub>2</sub> concentrations between (55-45 wt.%)(Fig. 3a,c).

Metasediments show a very broad range of SiO<sub>2</sub> (20-70 wt.%), Al<sub>2</sub>O<sub>3</sub> (0.10-1.25 wt.%) and TiO<sub>2</sub> (2-22 wt.%), likely due to them being a mixture of sandstones, clays and carbonates. The formation was separated out into a sediment and carbonate component (Fig. 3a,b,c). The sediments are best distinguished by SiO<sub>2</sub> (50-67 wt.%), TiO<sub>2</sub> (0.4 – 0.7 wt.%) and MgO (2-5 wt.%) concentrations that plot between the felsic and mafic data (Fig. 3a,b,c). Carbonate component displays the least overlap between lithologies and was best discriminated by SiO<sub>2</sub> (<40 wt.%)(Fig. 3a) and Al<sub>2</sub>O<sub>3</sub> (<9 wt.%)(Fig. 3b).

As the protolith of the metasomatite is relatively unknown and could be derived from either the metasediments or possibly the Moonta porphyry member within the Yorke Peninsula basement rocks (Cowley et al. 2003; Conor et al. 2010), the data shows variable chemistry as it is highly altered. The alteration seen will depend on fluids involved within the alteration processes and the protolith chemistry, and will in turn dictate the chemistry of individual samples. The metasomatite data overlaps with data from most other basement lithologies apart from the carbonate component of the sedimentary unit (Fig. 3a,b,c). Broad element concentrations for the metasomatite as follows; SiO<sub>2</sub> (35-80 wt.%), MgO (1-12 wt.%) and Al<sub>2</sub>O<sub>3</sub> (2-21 wt.%) (Fig.4-a,b). The only preferential grouping of data is displayed in (Fig. 3c) with TiO<sub>2</sub> concentrations between (0.25-0.75 wt.%).

The gneissic samples show a broad range in data which overlaps with other lithologies but preferentially towards high SiO<sub>2</sub> (50-77 wt.%) and low MgO (1-5 wt.%) concentrations (Fig. 3a). Al<sub>2</sub>O<sub>3</sub> data displayed a range between (11-20 wt.%) which is similar for felsic, sediment and mafic derived lithologies but data strays away from the carbonate component of the metasediments (Fig. 3b). Gneissic TiO<sub>2</sub> content displays the largest variation over all lithologies. Data ranges from (0 – 2.5 wt. %) which incorporates all lithological groups within this range (Fig. 3c).



Figure 3 - Major element geochemical analysis of Wallaroo basement rocks within Central Northern Yorke Peninsula displaying elements that best distinguished lithogical groups from one another. a) Wt. % plot SiO<sub>2</sub> v's MgO. b) Wt. % plot SiO<sub>2</sub> v's Al<sub>2</sub>O<sub>3</sub>. c) Wt. plot SiO<sub>2</sub> v's TiO<sub>2</sub>

#### TRACE ELEMENT CHEMISTRY

Elements associated with IOCG mineralisation can preferentially be hosted within certain lithological units, therefor display high concentrations of that element when compared to other lithologies. Potential pathfinder elements Ba, Ce, La and U are preferentially hosted within felsic dominated lithologies (felsic volcanics and granites) (Fig. 4a,e,f,h). Mafic dominated lithologies (amphibolite, mafic intrusive, mafic extrusive) host high concentrations of Cr, Co and Ni compared to other lithological groups (Fig. 4i,j,k) and the metasedimentary unit preferentially host high concentrations of Sb, W, Te and Cs (Fig. 4m-p). Other elements not directly associated with IOCGs can also be preferentially hosted within certain lithological units. Hf, Nb, Th and Zr shows high concentrations of V (Fig. 4k).

The metasomatite and gneiss show minimal preferential partitioning of any one trace element. This is possibly attributed to the unknown protoliths, the varying levels of alteration seen throughout the lithologies and how the alteration will manifest in different lithological groups.

#### Felsic volcanics and Granites



Figure 4 - Probability plots for trace elements within Wallaroo basement rocks on the Central Northern Yorke Peninsula showing concentration of elements in individual lithologies. Black line shows element detection limit. . a) Ba ppm. b) Hf ppm. c) Zr ppm. d) Ta ppm. e) Ce ppm. f) U ppm. g) Th ppm. h) La ppm. i) Cr ppm. j) Co ppm. k) V ppm. l) Ni ppm. m) Sb ppm. n) W ppm. o) Te ppm. p) Cs ppm.

## Petrology

Twelve core samples were chosen for petrological analyses. Samples were chosen based on preservation of sulphide/oxide phase and if samples originated from close proximity to known mineralisation. Of these twelve samples, three samples (1949023, 1949789, 1960847) were chosen for further petrological analyses as they best preserved alteration phases and show elevated concentrations potential IOCG pathfinder elements. Sample localities given in (Fig. 2)

#### SAMPLE 1949789 -

Sample1949789 is a metasediment (psammite) from the Wandearah Formation. The sample is dominated by a quartz and k-feldspar rich matrix (up to 90% of the whole rock). Thin siderite veins (~0.5mm) are preserved throughout sample which appears to be late staged and cross cutting the laminations within the quartz/k-feldspar matrix. Angular inclusions of small (~50 $\mu$ m) pyrite grains are preserved within the thicker portions of the veining (Fig. 5b,c). Small (<5  $\mu$ m) galena and disseminated barite were also seen to be growing within the thicker zones of siderite veining (Fig. 5c). Minor chlorite and carbonate (dolomite or calcite) grains were observed proximal to siderite veining, hematite staining and an unidentified and highly weathered mineral (Fig. 5d)

## SAMPLE 1949023 -

Sample 1949023 is a metasediment (siltstone) from the Wandearah Formation. The sample contains large up to 3mm angular sulphides within a brecciated quartz rich matrix (Fig. 5e-h). Sulphides are dominated by chalcopyrite and pyrite. Chalcopyrite and pyrite are present as large (~3mm) grains (Fig. 5e-g), smaller growths along the

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edges of carbonates (Fig. 5e,f) and areas of pyrite growth within a larger chalcopyrite grain (Fig. 5h). Dolomite and chlorite are observed as infill within brecciated zones, and are commonly in contact with sulphides (Fig. 5e,g,h). Dolomite grains are subhedral to euhedral (~1mm). Chlorite grains vary from large (5mm x 2mm) clustered growths to small thin strips (<0.5mm) growing within the brecciated matrix (Fig.5e-h). Most larger sulphide grains are in contact with chlorite (Fig. 5e-h). Thin (<0.5mm) carbonate veins are also observed in areas to be occupying the brecciated cracks (Fig. 5g).

#### SAMPLE 1960847 -

Sample 1960847 is a felsic volcanic from the Weetulta Formation. The sample comprises (~70% quartz/k-feldspar matrix). Veins up to 5mm thickness comprising of quartz (~1mm) with Fe-oxides cross cut the sample (Fig. 5i,j). Oxides are predominately hematite with lesser magnetite. Oxides occur within the quartz veins as thin (~1mm) veins (Fig. 5i,j). Smaller disseminated oxide grains (~10µm) are seen throughout the quartz/k-feldspar matrix (Fig. 5i,j). Ilminite and apatite were observed as small (~ 200µm) grains growing within the quartz-oxide vein but not directly proximal to the oxide alteration (Fig. 5k,l,m). The LREEs are host within bastnäsite and was found to be occupying areas proximal to ilminite and apatite (Fig. 5k,l,m).



Figure 5 - Petrological images from selected samples; 1949789 (a,b,c,d), 1949023 (e,f,g,h), 1960847 (i,j,k,l,m). a) CPL image of siderite vein within a quartz/k-feldspar matrix. b) Reflectedlight image of same area in (a)) displaying small pyrite and galena growth within siderite veining. c) Micro probe image to highlight the subhedral to euhedral pyrite growth and small (5 $\mu$ m) galena within siderite veining. d) CPL image showing areas of chlorite growth proximal to few larger grains left within sample. e) CPL image displaying amalgamated dolomite/calcite and chlorite growth proximal to sulphides. f) Reflected light image of same area in (e)) which details different areas of sulphide growth . g) CPL – with reflected light image of larger pyrite and amalgamated dolomite/calcite and chorite growth within brecciated cracks within the matrix. h) CPL – with reflected light image displaying chalcopyrite and pyrite (py within cpy) within a single sulphide and sharing a border with chlorite growth. i) CPL image of oxide veining within larger quartz vein within k-feldspar-quartz matrix. j) Reflected light image of same area in (i)) detailing Hem/mag oxide vein and smaller disseminated oxides within k-feldspar-quartz matrix. k) Electron back scatter image highlighting the presence of LREE mineral bastnäsite and its spatial relationship to ilminite and apatite. l) Electron back scatter image of Ce to show evidence of high LREEs residing within bastnäsite. m) Electron back scatter image of La to show evidence of high LREEs residing within bastnäsite. Symbols – Apa – apatite, Bas – bastnäsite, Chl – chlorite, CPL – Cross polarised light, Cpy – chalcopyrite, Dol – dolomite, Gal – galena, Hem – hematite, Ilm – ilmenite, Mag – magnetite, Py – pyrite, Sid – siderite, Qtz – quartz. Mineralogical symbols will be used throughout paper

## **Element Mapping**

Qualitative elemental maps were collected from three samples

(1949023,1949789/1960847) that were identified to preserve elevated concentrations of trace elements that may be potential pathfinders towards IOCG mineralisation (Ag, Al, As, Au, Bi, Ca, Ce, La, Mg, Mo, Na, Si, S, W and Zn). Six areas within samples were mapped to target oxide veining and sulphides to characterise potential zoning and identify potential LREE hosts. Representative maps are given in (Fig. 6,7,8). All maps given in Appendix B.

#### SAMPLE 1949789

Sample was used to map an area preserving Fe alteration within a k-feldspar/quartz matrix. Siderite veining is highlighted by elevations of Fe (Fig. 6e). Fine grained sulphides within the oxide veining are highlighted on the S map (Fig 6h), and are coincident with elevated Fe (Fig. 6e) and no Cu (Fig. 6d), indicating they are pyrite. Elevated regions of La and Ce indicate fine-grained LREE bearing mineral(s) occur throughout the sample.

## SAMPLE 1949023

Sample preserves pyrite and chalcopyrite as both primary and secondary growth. Element maps show the pyrite is highlighted by elevated Fe and no Cu and the chalcopyrite by coincident lower Fe and higher Cu (Fig. 6m,q)(Fig. 7d,h,m,q). Ce and La maps show small LREE highs within the chlorite/dolomite/quartz matrix surrounding the sulphides (Fig. 6l,o)(Fig. 7c,f,l,o). Map (Fig. 7a) displays inclusion-rich areas within the pyrite structure. Inclusions appear to be zoned with areas of barren clean pyrite and areas of inclusion rich growth. Zoned areas are ( $\sim 100 \mu m$ ) and alternate across the pyrite. Inclusions within the inclusion rich areas range from very small ( $\sim 1 \mu m$ ) to larger ( $\sim 20 \mu m$ ) elongated cracks that have a brecciated appearance (Fig. 7a).

## SAMPLE 1960847

Sample preserves sporadic oxide alteration within a quartz rich alteration vein. High Fe content is hosted within an Fe-Ti oxide with variable Fe and Ti content (Fig. 8e,n). Mapping identified LREE-rich areas are host within the LREE mineral bastnäsite (Fig. 8a,c,f,j,l,o). Bastnäsite is hosted within the oxide/quartz veining and has a close spatial relationship to the Fe-Ti oxide and apatite growth. (Fig. 8a,b,c,e,f,j,k,l,n,o)

Kym Michael Custance Identifying IOCG pathfinder elements



Figure 6 - Selected qualitative element maps for sample 1949789 and 1949023 showing back scattered electron image (BSE) Ca, Ce, Cu, Fe, La, Mg, S and Si content. a) Annotated BSE image detailing minerals in sample. b) Qualitative element map for Ca. c) Ce qualitative element map showing small LREE mineral within quartz-k-feldspar matrix. d) Cu qualitative element map. e) Fe qualitative element map highlight Fe content of siderite vein and pyrites within vein. f) La qualitative element map highlighting small LREE mineral within quartz-k-feldspar matrix. g) Mg qualitative element map. h) S qualitative element map highlighting sulphides present in sample. i) Si qualitative element map showing Si content in background matrix. j) Annotated BSE image detailing minerals in sample. K) Qualitative element map for Ca showing Ca content in dolomite/calcite. l) Ce qualitative element map showing LREE mineral within quartz matrix. m) Cu qualitative element map highlighting small LREE mineral within quartz matrix. p) Mg qualitative element map showing Fe content in sulphides (light and dark blue). o) La qualitative element map highlighting small LREE mineral within quartz matrix. p) Mg qualitative element map shows Mg content in dolomite. q) S qualitative element map highlighting two different sulphides present (Pyrite lighter blue/chalcopyrite dull blue). r) Si qualitative element map showing Si content in background matrix.

Kym Michael Custance Identifying IOCG pathfinder elements



Figure 7 - - Selected qualitative element maps for sample 1949023 showing back scattered electron image (BSE) Ca, Ce, Cu, Fe, La, Mg, S and Si content. a) Annotated BSE image detailing minerals in sample. b) Qualitative element map for Ca showing Ca content in dolomite/calcite. c) Ce qualitative element map showing small LREE within quartz matrix. d) Cu qualitative element map showing possible Cu rich area in pyrite inclusion zonation. e) Fe qualitative element map highlighting Fe content in dolomite. h) S qualitative element map highlighting sulphides present. i) Si qualitative element map showing Si content in background matrix. j) Annotated BSE image detailing minerals in sample. k) Qualitative element map for Ca showing Ca content in dolomite. l) Ce qualitative element map showing small LREE mineral within quartz matrix. m) Cu qualitative element map showing Cu rich area in chalcopyrite. e) Fe qualitative element map highlighting small LREE mineral within quartz matrix. g) Mg qualitative element sulphides and chlorite. f) La qualitative element map showing Si content in dolomite. l) Ce qualitative element map showing small LREE mineral within quartz matrix. m) Cu qualitative element map showing Cu rich area in chalcopyrite. e) Fe qualitative element map shows Mg content in dolomite. f) La qualitative element map highlighting small LREE mineral within quartz matrix. g) Mg qualitative element map shows Mg content in dolomite. h) S qualitative element map highlighting small LREE mineral within quartz matrix. g) Mg qualitative element map shows Mg content in dolomite/chlorite. h) S qualitative element map showing Si content in dolomite/chlorite. h) S qualitative element map highlighting small LREE mineral within quartz matrix. g) Mg qualitative element map shows Mg content in dolomite/chlorite. h) S qualitative element map highlighting two sulphides present. i) Si qualitative element map showing Si content in background matrix.

Kym Michael Custance Identifying IOCG pathfinder elements



Figure 8 - Selected qualitative element maps for sample 1960847 showing back scattered electron image (BSE) Ca, Ce, Cu, Fe, La, Mg, S and Si content. a) Annotated BSE image detailing minerals in sample. b) Qualitative element map showing Ca content in apatite. c) Ce qualitative element map showing Ce residing within bastnäsite . d) Cu qualitative element map. e) Fe qualitative element map highlighting varying Fe content in Ti/Fe oxide and ilmenite. f) La qualitative element map showing La residing within bastnäsite. g) Mg qualitative element map shows Mg content in variation of Ti/Fe oxide. h) S qualitative element map displaying no sulphides present. i) Si qualitative element map showing Ca content in apatite. l) Ce qualitative element map showing Ce residing within bastnäsite . m) Cu qualitative element map. n) Fe qualitative element map highlighting varying Fe content in Ti/Fe oxide and ilmenite. o) La qualitative element map showing La residing within bastnäsite . m) Cu qualitative element map. n) Fe qualitative element map highlighting varying Fe content in Ti/Fe oxide and ilmenite. o) La qualitative element map showing La residing within bastnäsite . m) Cu qualitative element map. n) Fe qualitative element map highlighting varying Fe content in Ti/Fe oxide and ilmenite. o) La qualitative element map showing La residing within bastnäsite. p) Mg qualitative element map shows Mg content in variation of Ti/Fe oxide. q) S qualitative element map displaying no sulphides present. r) Si qualitative element map showing Si content in background matrix. Symbols – Apa – Apatite, Bas – bastnäsite, ilm – ilmenite

## **Mineral chemistry**

Laser-Ablation Inductively-Coupled Mass Spectroscopy (LA-ICP-MS) analysis was undertaken on sample 1949023 as it preserved multiple stages of sulphide growth. Chalcopyrite and pyrite are preserved as large primary sulphides with inclusions, possible remobilized secondary phases and areas where chalcopyrite is replacing pyrite. Analysis was undertaken on three areas within sample to determine if potential pathfinder trace element signatures were apparent within these sulphide phases. Representative data in (Table. 4). Raw data in Appendix C.

#### PYRITE

Stoichiometric values for pyrite displayed a bimodal distribution of data (Fig. 9a). A general linear trend is apparent yet displays less clustering than chalcopyrite data (Fig. 9b). The negative linear trend begins at Fe rich/S poor values of (Fe: 0.90 - S:1.55 mol) and progress to Fe poor/S rich values of (Fe:0.88 - S:1.70 mol) (Fig. 9a). When compared to stoichiometric values for pyrite, data derived from Yorke Peninsula generally plots lower than stoichiometric values (Fig. 9a). Two high Fe outliers are from the inclusion-rich areas of a larger pyrite (Fig. 10a, spot 6,10).

#### **Trace Element Analysis**

Ag and Bi showed the majority of the pyrite data host some elevated concentrations of these elements. Ag data shows concentrations range between 5-25 ppm (Fig. 11a). Three high values correspond to zoned inclusion rich areas within a larger pyrite (Fig. 10a,spot 7,9,10). Bi data displays concentrations between 10-330 ppm (Fig. 11d). Three highest readings also correspond to the zoned inclusion-rich areas within a large pyrite

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(Fig. 10a,spots 8,9,10). As and Mo data display a bimodal distribution of data. As concentrations range from 50-4400 ppm (Fig. 11c) and Mo concentrations range from 10-615 ppm (Fig. 11d). High As readings correspond to the edge of a pyrite within a larger chalcopyrite (Fig. 10c, spot 1,2). High Mo readings correspond to inclusion poor areas within a larger pyrite (Fig. 10a,spot-5,7,8). Much of the data for Au is low except for four points. These four points data ranges from 37-74 ppb (Fig. 11e). The four points are attributed to a larger pyrite with zoned inclusions (Fig. 10a, spot-2,6,7,9). Cu content in pyrite is low when compared to chalcopyrite data with all readings falling <15000ppm (Fig. 11f). W data shows concentrations within pyrite are low except for one reading that corresponds to the edge of a pyrite growing within larger chalcopyrite (Fig. 10c, spot 11) ; high reading 1.32 ppm.

#### Chalcopyrite

Stoichiometric values for chalcopyrite displays a negative relationship between Cu and S content. Cu rich/S poor values begin at Cu: 0.52 - S:1.150 mol and progress to Cu poor/S rich values of Cu:0.47 - S:1.340 (Fig. 9b). Data shows a general grouping of concentrations between S: 1.2 - 1.35 mol and Cu: 0.47 - 0.51 mol, which when compared to true stoichiometric value for chalcopyrite shows that the overall S content for chalcopyrite high and Cu content is low (Fig. 9b). Data displays three clear outliers away from average concentrations. The S rich/Cu poor data was taken from a small pyrite inclusion within a larger chalcopyrite grain (Fig. 10c, spot 8). The high Cu/S poor data is from near the centre of a larger chalcopyrite (Fig. 10c, spots 4, 9).

### **Trace element analysis**

Ag displayed a clustering of data (~4ppm) but is low when compared to pyrite data (Fig. 11a). Bi, As and Mo displayed low concentrations <20ppm (Fig. 11 b,c,d). Au and W is predominantly at low concentrations within chalcopyrite except for one data point for each element; Au: 200ppb, W: 0.58ppm (Fig. 11e,h) and corresponds to an area of possible secondary immobilized chalcopyrite growth next to larger chalcopyrite (Fig. 10b,spot 4). Cu concentrations are host within chalcopyrite as expected (Fig. 11f).

Table 4 - Summary of LA-ICP-MS trace element data for sulphide phases (ppm)

Element	Sulphide	\$33	Fe57	Cu65	Zn66	As75	Mo95	Ag107	W184	Au197	Bi209
923-1-01	Pyrite	496005.09	465500.44	9703.67	13.36	126.49	11.74	15.48	0	0	149.79
923-1-02	Pyrite	508062.5	465500.44	3735.23	41.36	149.2	15.86	14.97	0	0.054	184.73
923-1-03	Pyrite	549157.63	465500.44	5742.8	13.89	83.53	48.7	10.76	0	0	170.28
923-1-04	Pyrite	532321.88	465500.47	8651.27	6.73	55.76	41.29	6.15	0	0	133.05
923-1-05	Pyrite	523433.97	465500.41	2146.25	29.83	122.84	391.48	7.01	0	0	136.1
923-1-06	Pyrite	466953.88	465500.44	530.53	12.72	144.2	59.58	17.94	0	0.074	130.89
923-1-07	Pyrite	551877.38	465500.47	5522.88	39641.85	125.46	613.54	20.96	0	0.058	153.49
923-1-08	Pyrite	538131.5	465500.47	15978.45	2125.47	140.11	423.97	16.26	0	0	313.01
923-1-09	Pyrite	526380.94	465500.47	1325.13	66.18	157.37	15.95	23.3	0	0	237.6
923-1-10	Pyrite	466541.44	465500.44	1767.13	89.97	715.04	34.73	20.8	0	0	247.7
923-2-01	Chalcopyrite	498030.81	304300.13	349077.47	16.17	0	0	4.71	0	0	0.176
923-2-02	Chalcopyrite	435278.31	304300.16	347644.94	14.56	0	0	5.35	0	0	0.209
923-2-03	Chalcopyrite	447556.25	304300.16	349083.69	20.83	0	0	4.43	0	0	0.59
923-2-04	Chalcopyrite	456490.94	304300.16	330785.66	39.55	5.08	0	6.23	0.58	0.2	16.88
923-2-05	Pyrite	496724.66	465500.47	153.42	76.57	181.18	0.19	0.79	0.163	0	23.92
923-2-06	Pyrite	567154.69	465500.44	932.67	10.75	360.32	0	3.89	0	0	29.99
923-2-07	Pyrite	490172.84	465500.44	260.98	2.95	50.22	0	0.69	0	0	14.97
923-2-08	Pyrite	551442.56	465500.47	281.01	7.72	671.21	1.67	0.97	0.095	0	35.39
923-2-09	Chalcopyrite	497333.53	304300.16	344485.16	9.31	3.78	0	3.8	0	0	0.275
923-2-10	Chalcopyrite	467420.94	304300.16	343024.19	35.2	0	0	4.46	0	0	0.261
923-2-11	Chalcopyrite	455710	304300.16	349644.81	12.19	0	0	4.41	0	0	0.301
923-3-01	Pyrite	532536.94	465500.47	4168.51	47.71	3103.26	0.193	3.49	0.098	0	35.77
923-3-02	Pyrite	538525	465500.44	24643.83	117.21	4490.07	0.61	11.97	1.32	0	111.67
923-3-03	Pyrite	527272.19	465500.44	1374.7	4.81	61.6	0	1.74	0	0	22.08
923-3-04	Chalcopyrite	406647.19	304300.16	347278.25	16.56	0	0	2.55	0	0	1.12
923-3-05	Chalcopyrite	433320.97	304300.16	340959.06	16.39	0	0	2.66	0	0	0.99
923-3-06	Chalcopyrite	459539.94	304300.19	341721.81	13.66	0	0	3.28	0	0	0.9
923-3-07	Chalcopyrite	452564.66	304300.16	342046.19	74.6	150.49	0	3.38	0	0	2.4
923-3-08	Chalcopyrite	479619.31	304300.13	261745.2	20.66	0	0	3.22	0	0	2.62
923-3-09	Chalcopyrite	384433.78	304300.16	344156.38	25.1	0	0	3.04	0	0	1.56
923-3-10	Chalcopyrite	445795.31	304300.19	349102.78	14.63	0	0	2.26	0	0	0.461
923-3-11	Pyrite	529581.88	465500.5	19019.35	7747.34	165.1	0.37	4.57	0	0.037	30.27
923-3-12	Chalcopyrite	450336.16	304300.19	351986.91	13.67	0	0	2.33	0	0	0.286
923-3-13	Chalcopyrite	435062.5	304300.19	349860.75	9.55	0	0	2.59	0	0	0.424
923-3-14	Chalcopyrite	438268.72	304300.19	347481.78	118.63	9.34	0.66	3.82	0	0	1.17
923-3-15	Chalcopyrite	493631.53	304300.19	341048.56	13.51	0	0	2.53	0	0	0.78



Figure 9 – BSE images highlighting where individual spots were analysed within sample 1949023. a) Analysis was undertaken on a larger pyrite with apparent inclusion. 10 spots were analysed; locations numbered recorded in image. b) Analysis undertaken on area that preserved large primary growth and smaller remobilized growth. 11 spots were analysed (7 chalcopyrite, 4 pyrite); locations numbered recorded in image. c) Analysis was undertaken on chalcopyrite as complete chalcopyrite and area of intergrown chalcopyrite growth. 15 spots were analysed (11 chalcopyrite, 4 pyrite); ); locations numbered recorded in image



Figure 10 - Details how the element values of sulphide grains measured against stoichiometric values for each sulphide. Stoichiometric values for chalcopyrite (S= 1.08 mol - Cu=0.54 mol) and pyrite (S= 1.66 mol - Fe= 0.83 mol). Purple dot in plots represents stoichiometric values for each sulphide



Figure 11 - Trace element v's Fe plots to determine trace element concentrations within selected sulphide phases. Pyrite data displayed in green; chalcopyrite data displayed in red. a) Ag showing slight elevations in chalcopyrite but mainly hosted in pyrite. b) As showing most samples low in Ag except for two high in pyrite data. c) Au showing the majority of sulphides sampled show no elevations except for one chalcopyrite and 4 pyrite. d) Bi showing majority of Bi readings are hosted in pyrite. e) Cu showing highest concentrations attributed to chalcopyrite and much smaller amounts in pyrite. f) Mo showing readings are hosted within pyrite. g) S showing content hosted within both sulphides. h) W showing concentrations are low from selected samples.

### DISSCUSSION

#### Whole rock geochemistry

Whole rock geochemical analysis was undertaken to determine background element concentrations and lithogeochemistry of each defined lithological group within the Wallaroo basement lithologies. Determining basic lithological characteristics of the lithological groups is crucial in determining a background chemistry of the formation units (e.g. Reimann et al. 2005; Kyser et al. 2015), determining an element standard for further normalising data (e.g. Warren et al. 2015; Sakan et al. 2015) and to be able to measure potentially anomalous data against (e.g. McQueen 2006; Kyser et al. 2015). Samples were separated out into six generic groups based on logging data; felsic volcanics, granites, metasediments, mafics, gneiss and metasomatite.

#### MAJOR ELEMENTS

The lithological groups were best discriminated using SiO<sub>2</sub> and MgO concentrations (Fig. 12a). For the purpose of this analysis the metasediments were separated into clastic and carbonaceous sediment (Fig.12a). The MgO vs SiO<sub>2</sub> diagram displays a negative linear trend that separates out lithologies. Felsic derived lithologies (felsic volcanics and granites) occupy the high SiO<sub>2</sub>/low MgO section of the diagram (65-80 wt.% SiO<sub>2</sub>;0-2 wt.% MgO). The carbonate component of the metasediments occupies the high MgO/ low SiO<sub>2</sub> area and plots between (9-14 wt. % MgO; 30-40 wt. % SiO<sub>2</sub>) (Fig. 12a). Mafic derived litholgies (mafic intrusive, mafic extrusive, amphibolite) and the purely sediment component of the metasediments plots between these two lithlogies (Sediment; (1-4 wt. % MgO/50-70 wt. SiO<sub>2</sub>), Mafic; (4.5-10 wt.% MgO/S45-55 wt.%

MgO) (Fig. 12a). This separation of lithological groups along this negative linear trend gives each lithology a clear distinctive area.  $SiO_2 vs TiO_2$  diagram (Fig.12b) also displays a separation of litholgies; felsic component (0.25-0.75 wt. %  $TiO_2$ )/(60-80 wt. %  $SiO_2$ ), sediment; (0.45-0.75 wt. %  $TiO_2$ )/(50-70 wt. %  $SiO_2$ ), mafic; (1-3 wt. %  $TiO_2/45-55$  wt. %  $SiO_2$ ) and carbonate; (0.10-0.35  $TiO_2$ )/(15-40 wt. %  $SiO_2$ ) (Fig. 12b). Yet displays much more overlap within element concentrations eg.  $TiO_2$ concentrations through felsic, sediment and carbonate lithologies. Distinguishing the metasomatite and the gneiss is difficult to determine as of an unknown photolith of the metasomatite and varying levels of metamorphic mineral growth in the gneiss, yet as the two lithological units still generally plot along this negative linear trend it suggests the lithologies are still controlled by mineral partitioning of  $SiO_2$  and MgO content (Fig. 12a).

#### TRACE ELEMENTS

Lithological groups discriminated nicely using Zr ppm and V ppm (Fig. 12c). Zr vs V shows that felsic derived data (felsic volcanics and granites) plots below a V:Zr ratio of 0.35-1, sediment and carbonate of the metasediments plots between ratios of V:Zr 0.35:1 and 3:2 and mafic data (mafic intrusive, mafic extrusive, amphibolite) plots above a V:Zr ratio of 3:2. Showing clearly that felsic lithologies have high Zr/low V and mafic lithologies have high V/low Zr. Sediment and carbonates plot clearly between the two (Fig. 12c)



Figure 12 - Geochemical plots of major and trace elements displaying separation of lithological groups. a) MgO v's SiO<sub>2</sub> displaying a negative linear trend showing clear separation of lithological groups. b) TiO<sub>2</sub> v's SiO<sub>2</sub> showing separation of lithological groups, yet displays overlap through TiO<sub>2</sub> concentrations. c) V v's Zr showing separation of lithological groups as ratios of element concentrations.

#### Pathfinder elements to potential IOCG mineralisation

The whole rock element suite (appendix D) was assessed for potential pathfinder elements, with emphasis placed on known elements associated with IOCG mineralisation Ag, As, Au, Ba, Bi, Ca, Cd, Ce, Co, Cr, Cs, Cu, F, Fe, K, La, Mn, Mo, Ni, S, Sb, Se, Sn, Te, U and W (Mark et al. 2006; Fabris et al. 2013a; Wang et al. 2013; Hill et al. 2014). Individual lithologies can show elevations of potential pathfinders (e.g. La and Ce in felsic lithologies, Co in mafics) (Fig. 4). However, the overall raw data of the broad lithological groups cannot be compared without normalisation (Warren et al. 2007; Sakan et al. 2015). Lithological groups across Yorke Peninsula can be discriminated using a number of elements – Mg, Si, Ti V, and Zr (Fig. 12). To compare data between the lithological groups so potential pathfinder elements can be recognised, data needs to normalise against an element that clearly displays unique element concentrations within the lithologies (Fig. 12a) and either element (SiO<sub>2</sub>/ MgO) could potentially be used for normalisation. Each element was assessed and tested for normalisation but ultimately MgO was selected as it best accommodated the data. A criteria was set for identifying and determining the validity of potential pathfinder elements.

- Elements histogram and probability plots must be relatively uniform, show three potential populations of data (background-interesting-anomalous) and display potential high anomalous element concentration outliers.
- High concentrations observed from histogram and probability plots must be anomalous (10x) average crustal abundance (e.g. Fabris et al. 2013). Average crustal abundance table in appendix xx.
- Elements with high concentrations must be above detections limits and preference given to elements associated with IOCGs.
- 4. Element concentrations must vector to known mineralisation or possible exploration prospects within the area.

From these analyses ten elements (Ag, As, Au, Bi, Ce, Cu, Mo, La, S, W) were selected and assigned as pathfinder elements for the central Yorke peninsula. These elements were assigned based on the geochemical pathfinder criteria and association with IOCGs (Fabris et al. 2013a; Fabris et al. 2013b; Corriveau 2006; Foster et al. 2007; Darwish and Poellmann 2010; Cook 2015; Marschik and Fontboté 2001; Haynes et al. 1995; Wood and Samson 2000; Heinrich 1990). All elements in (Fig. 13-15). Example of pathfinder criteria in Appendix:D.



Figure 13 – Assigned pathfinders elements within Yorke Peninsula basement rocks and their passed criteria a) Ag. b) As. c) Au. d) Bi



Figure 14 - Assigned pathfinders elements within Yorke Peninsula basement rocks and their passed criteria a) Ce. b) Cu. c) La. d) Mo



Figure 15 - Assigned pathfinders elements within Yorke Peninsula basement rocks and their passed criteria a) S. b) W

## **Element deployment**

The following elements have been determined as pathfinder elements towards potential IOCG mineralisation in the northern Yorke Peninsula: Ag, As, Au, Bi, Cu, Ce, La, Mo, S and W. SEM and (LA-ICP-MS) analysis was undertaken on sulphide and oxide phases to constrain which minerals these elements may reside within.

#### Ag

Silver is a common trace element within sulphides, especially pyrite (Large et al. 2009; Deditius et al. 2014; Abraitis et al. 2004; Foster et al. 2007). Silver concentrations within pyrite from sample 1949023 ranged from 5-25 ppm and showed elevations within majority of the pyrite data, and is particularly elevated within the inclusion-rich rim of the pyrite (Fig. 11a) (Fig. 9a, spot9,10). Silver is often coincident with elevated As and Bi within pyrite data (Fig. 11b,c), which is commonly observed in inclusion-rich pyrite growth (Foster et al. 2007; Huston et al. 1995; Reich et al. 2013). Silver is a trace element associated with mineralisation on Yorke Peninsula and has been documented as trace element concentrations within ore zones at the Moonta and Poonta deposits in the Moonta-Wallaroo mining district (Fig. 2) by Both et al. (1993); Keeling et al. (2003) and Drexel et al. (1993), yet a mineral host was not stated. However there are a number of possible mineral hosts on Yorke Peninsula including; electrum AuAg, hessite Ag<sub>2</sub>Te, petzie Ag<sub>3</sub>AuTe<sub>2</sub> and have been observed within ore zones at the Hillside deposit (Fig. 2) (Twining et al. 2013; Ismail et al. 2014) . SEM analyses determined small 5µm galena growing within siderite veining in sample 1989789. Galena PbS is a mineral that commonly holds trace element concentrations of silver (George et al. 2015; Simanenko 2007; Qian 1987) and has been reported across Yorke Peninsula in the Hillside and the

Moonta-Wallaroo area (Conor et al. 2010; Zang 2002; Ismail et al. 2014). The presence of galena may be attributing to the silver geochemical signature.

#### As

Arsenic is an element commonly associated with pyrite and is often with occurrences of other trace elements via coupled substitutions (Abraitis et al. 2004; Foster et al. 2007; Huston et al. 1995; Large et al. 2009). Element concentrations were observed to be residing within pyrite in sample 1949023. Concentrations proximal to the boundaries of the inclusion-rich pyrite growth were high ~3500ppm relative to other pyrite grains analysed <1000ppm (Fig. 11b). Other possible mineral hosts on Yorke Peninsula include; Arsenopyrite (FeAsS) which is the most common arsenic bearing mineral (King 2002), tennantite  $Cu_6[Cu_4(Fe,Zn)_2]As_4S_{13}$ , cobaltite CoAsS and smaltite (Co,Fe,Ni)As2 and have been documented as minor/trace minerals at Hillside and Moonta- Wallaroo (Fig. 2) (Ismail et al. 2014; Twining et al. 2013; Jones 1940). Hematite and magnetite also have an association with arsenic (Giménez et al. 2007) and as the two are the dominant oxide phases on Yorke Peninsula (Conor et al. 2010; Zang 2002; Kontonikas-Charos et al. 2014) the oxides may also be attributing to the arsenic anomaly.

#### Au

Gold is an element found within sulphide phases and is commonly present as inclusions within sulphide growth eg. (Foster et al. 2007; Large et al. 2009). Four data points within pyrite from sample 1949023 showed elevated Au concentrations within inclusion-rich growth and ranged between 37-74 ppb (Fig. 11c). One high reading of

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200ppb was attributed to a smaller secondary chalcopyrite (Fig. 11c). Trace element concentrations of gold within chalcopyrite has been documented by Both et al. (1993); Conor et al. (2010) and Skirrow et al. (2007) who observed Au mainly occurring as inclusions within chalcopyrite in the Moonta-Wallaroo area. As sulphide analysis determined Au concentrations to be hosted within chalcopyrite and previous studies documented a similar Au occurrence it is possible the sulphide may host a significant portion of the gold geochemical anomaly. Other possible mineral hosts on Yorke Peninsula include; electrum (AuAg) and petzie (Ag<sub>3</sub>AuTe<sub>2</sub>) as seen at nearby Hillside (Twining et al. 2013)

Bi

Bismuth is an element commonly accommodated within pyrite (Large et al. 2009; Deditius et al. 2014; Reich et al. 2013; Huston et al. 1995; Foster et al. 2007). High concentrations ranged from 300-330ppm and corresponded to the inclusion-rich edge of pyrite growth from sample 1949023 (Fig. 9a). High Bi concentrations were coupled with elevations of Ag and As (Fig. 9a,spot 9,10)(Fig. 11a,c). Trace element concentrations of Bi have been observed within ore rich areas in the Moonta-Wallaroo area by Both et al. (1993); Drexel et al. (1993) and Keeling et al. (2003) but a mineral host was not documented. However there are a number of possible mineral hosts on Yorke Peninsula including; bismuthinite  $Bi_2S_3$  which Conor et al. (2010) recorded seeing rare cases of across Yorke Peninsula and aikinite PbCuBiS<sub>3</sub> seen at nearby Hillside (Fig. 2) by (Twining et al. 2013). Galena PbS is a mineral that commonly holds trace element concentrations of bismuth (George et al. 2015; Simanenko 2007; Qian 1987) and has been reported across Yorke Peninsula (Conor 1995; Zang 2002; Twining et al. 2013). This occurrence of galena may also be attributing to the Bi anomaly.

## Ce/La –

Light rare earth trace element signature was observed within sample 1960847 which contained the LREE mineral bastnäsite (Ce,La,Y)CO<sub>3</sub>F (Fig. 8a,c,f,j,l,o). Bastnäsite was observed to have a spatial relationship with apatite and Ti-Fe oxides (Fig. 8a,j). Kontonikas-Charos et al. (2014) observed bastnäsite growing with a similar spatial relationship to Ca rich minerals and Fe-Ti oxides and documented widespread emplacement within igneous rich rocks throughout the Moonta-Wallaroo area. Bastnäsite is also documented at Hillside (Fig. 2) (Twining et al. 2013). As SEM analysis uncovered the LREE mineral host and Kontonikas-Charos et al. (2014) documented seeing the same bastnäsite relationship within Moonta-Wallaroo studies, it is likely that bastnasite is attributing to the Ce/La geochemical anomaly. Other rare earth minerals; synchysite Ca(Ce,La)(CO<sub>3</sub>)<sub>2</sub>F, allanite

 $(Ce,Ca,Y,La)_2(Al,Fe^{+3})_3(SiO_4)_3(OH)$  and monazite  $(Ce,La)PO_4$  have been documented across Yorke Peninsula (Kontonikas-Charos et al. 2014; Ismail et al. 2014; Conor et al. 2010; Twining et al. 2013) which may also be attributing to the geochemical anomaly.

#### Cu –

Copper trace element signature was observed within the chalcopyrite sampled as is expected (Fig. 11e). Analysis determined pyrite was also holding some trace element concentrations of Cu. High concentrations ranged from 8,000-15000ppm within a larger pyrite with inclusion-rich growth (Fig. 9a) and up to 25,000ppm from a pyrite within a larger chalcopyrite (Fig 9c) but may be picking up residual Cu from the surrounding chalcopyrite (Fig. 11e). Cu concentrations attained from pyrites analysed may attributing to the Cu geochemical signature, yet chalcopyrite Cu concentrations ~ 340,000ppm would be much more dominant. Common Cu minerals chalcocite Cu<sub>2</sub>S, bornite Cu<sub>5</sub>FeS<sub>4</sub> and covellite CuS and minor Cu minerals carrollite CuCo<sub>2</sub>S and digenite Cu<sub>9</sub>S<sub>5</sub> have been observed within ore zones in the Moonta-wallaroo area and Hillside and could be contributing to the Cu geochemical anomaly (Both et al. 1993; Zang 2002; Twining et al. 2013; Ismail et al. 2014; Hafer 1991; Morales Ruano et al. 2002).

#### Mo -

Molybdenum is a common trace element within pyrite (Abraitis et al. 2004; Large et al. 2009; Deditius et al. 2014; Huston et al. 1995). Molybdenum concentrations within pyrite from sample 1949023 displayed a bimodal distribution of data with concentrations within inclusion-poor areas displaying high values 450-610ppm compared to other pyrite areas which displayed values <100ppm (Fig. 8f). Studies of Moonta Wallaroo and Hillside have commonly recorded molybdemite MoS<sub>2</sub> to be associated within the ore rich areas which is a likely mineral host for the molybdenum geochemical signature. (Conor 1995; Both et al. 1993; Conor et al. 2010; Skirrow et al. 2007; Twining et al. 2013; Zang 2002)

**S** –

Sulphur trace element concentrations were hosted within sulphide phases as to be expected. SEM and (LA-ICP-MS) analysis determined high S concentrations coming from pyrite with lesser amounts attributed to chalcopyrite (Fig. 11g). Pyrite is a mineral commonly found in association with ore deposits across Yorke Peninsula (Conor et al. 2010; Twining et al. 2013; Zang 2002; Jones 1940) and is likely to be attributing to some of the S geochemical anomaly. High S stoichiometric values attained from chalcopyrite may indicate chalcopyrite is also possibly attributing to the high S anomaly. Other common S minerals seen across Yorke Peninsula; Pyrrhotite Fe<sub>7</sub>S<sub>8</sub>, galena PbS and chalcocite  $Cu_2S$  (Conor et al. 2010; Zang 2002; Twining et al. 2013) may also be attributing to the geochemical anomaly.

#### **W** –

Tungsten signature was difficult to determine where it may be residing. Sulphide analysis yielded no anomalous tungsten concentrations coming out of pyrite or chalcopyrite (Fig. 11h). Yet tungsten is an element with associations with IOCGs and has been reported as trace element concentrations within ore zones in other IOCG style deposits (Ootes et al. 2010; Montreuil et al. 2015). The most common tungsten bearing mineral is wolframite (Fe,Mn)WO<sub>4</sub> (Wood and Samson 2000), which has not yet been documented on Yorke Peninsula. Rare occurrences of scheelite (CaWO<sub>4</sub>) were documented by Jones (1940) petrological study of the Wallaroo and Moonta mining district could be accounting for the some of the tungsten geochemical anomaly. In an exploration context if elements (Ag, As, Au, Bi, Ce, Cu, La, Mo, S and W) were to be used as IOCG exploration vectors and sampling for mineral exploration was carried out in the field you would look for samples containing these elements. This study shows that interesting elements can be persevered in sulphide phases and LREE hosts within bastnasite, therefore analysis of samples containing these mineral phases will potentially increase the ability to recognise and map out these geochemical signatures. However, it is likely that not all interesting samples will contain pyrite/chalcopyrite or bastnasite, and other mineral phases may host significant concentrations of the pathfinder elements, therefore samples without these mineral phases should not be disregarded.

#### **IOCG prospectivity index**

A prospectivity index was created for central Yorke Peninsula in much the same way as Fabris et al. (2013a) constructed an prospectivity index for the eastern Gawler Craton . The index is determined by combining assigned IOCG elements (Ag, As, Au, Bi, Ce, Cu, La, Mo, S and W). Each element is assigned a value of one when above 10 times average crustal abundance and zero when below. The prospectivity index is the average of these assigned values, where a prospectivity index of 1 indicates a highly prospective area that contains high concentration of pathfinder elements, and a prospectivity index of 0 is a low prospectivity area. Each samples prospectivity index and locality are plotted in vectoring maps to highlight hot and not so hot areas for possible exploration/mineralisation across central Yorke Peninsula (Fig. 16).

The prospectivity index highlighted an area proximal to the township of Alford and Bute as a high prospectivity area, yet the index did not highlight areas of known mineralisation around the Moonto-Wallaroo district (Fig. 2). This could potentially be attributed to open file drill holes within the area not being representative of ore related zones. Also may be attributed to employing a 10x average crustal abundance as anomalous element indicator for samples. Potentially this could be lowered to highlight other known mineralised areas in the area.



# Central-Northern Yorke Peninsula Prospectivity Index Vecoring Map

Figure 16 - – Gridded maps displaying the prospectivity index concentrations across central northern Yorke Peninsula highlighting the Alford Ridge as a potential area for further exploration. Prospectivity index scores: >0.8 is highly significant, 0.6-0.8 is very significant, 0.4-0.6 is significant. Note representative highs along the Alford Ridge achieved a prospectivity score of 0.7.

#### Alford Ridge -

The Alford Ridge is a SE-NW trending structure that begins approximately 20kms above the township of Alford and extends down through Bute (Fig. 16). The area achieved 'very significant' prospectivity index numbers of 0.7 for respective highs along the Alford Ridge (Fig. 13). The Alford Ridge is considered a potential prospect for exploration and mineral exploration tenement is currently held by Sandfire Resources and Goyder Resources. Drill hole data from surrounding areas displays evidence for mineralisation in the area and was documented by Adelaide Resources Limited and AMALG Resources NL. Adelaide resources reported significant drill hole Cu intercepts of 10m @ 2.3% Cu from 59m (Washbourne 2014). AMALG returned significant Cu or Cu+Au values (e.g. Drillhole AL 9: 10 m @ 1.69% Cu, containing 5 m @ 2.78% Cu and 3.17 g/t Au) and documented a shallow, oxide copper resource comprising 60 000 t @ 2% Cu with potential for extraction by open cut. Data more proximal to the Alford Ridge was documented by Argonaut Resources whom have a joint venture with Sandfire Resources. Argonaut Resources recorded multiple significant drill hole intercepts of Cu and Ag (eg. Drillhole ALDDH01: 76m @ 0.95% Cu from 138m including 12 @ 2.18% Cu from 146m and ALDDH027: 10.04m @ 0.32% Cu and 6.69 g/t Ag from 180m including 5m @ 0.5% and 11.76 g/t from 180m) (Argonaut 2015). The presence of known mineralisation in the area highlights the prospectivity index being a useful exploration tool and could potentially be used in less explored areas to highlight areas of mineralisation/exploration prospects.

#### **Implications for exploration**

If exploration was carried out on central Yorke Peninsula 1m composite core samples would be adequate to generate geochemical data to determine geochemical anomalies. An understanding of background geochemical characteristics of lithological units is crucial for assigning pathfinder elements. Interesting pathfinder elements Ag, As, Au, Bi, Ce, Cu, La, Mo, S and W can be present within pyrite, chalcopyrite and bastnasite and should be targeted when sampling. However, samples that do not contain these minerals should not be disregarded as interesting element concentrations may be present in other mineral phases. A prospectivity index can be created unique to central Yorke Peninsula to highlight prospective areas for mineralisation by assigning a threshold of 10 times average crustal abundance as an anomalous indicator.

#### **Future work**

Sulphide analysis from samples derived from areas directly proximal to ore zones to camp scale area to categorise trace element concentrations within sulphides to determine if concentrations increase toward ore zones.

Further analysis on different sulphide/oxide phases and mineral species to further constrain where interesting pathfinder elements may be residing. Create an alteration index for central Yorke Peninsula independent of Au and Cu concentrations and compare to prospectivity index created in this study.

#### CONCLUSIONS

Whole rock geochemical analysis can distinguish the six defined lithological groups from one another by major and trace element concentrations that are unique to each lithology.

Elements Ag, As, Au, Bi, Ce, La, Mo, S, and W can be recognised from whole rock geochemical analysis and assigned as pathfinder elements toward potential IOCG mineralisation on central Yorke Peninsula

Element deployment uncovered inclusion-rich zonation growth seen from pyrite holds concentrations of Ag, As, Bi and Mo. Chalcopyrite holds concentrations of Au. LREE trace element concentrations are host within bastnäsite.

A prospectivity index unique to central Yorke Peninsula can be created by combing assigned pathfinder elements to highlight prospective areas. An area proximal to the township of Alford and Bute is highlighted as a prospective area. As area is already known for significant mineralisation it strengthens the usefulness of the prospectivity index as an exploration tool.

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## APPENDIX A: RAW GEOCHEMICAL DATA AND ELEMENT SUITE ANALYSED AND ANALYTCICAL TECHNIQUES

Element(s)	Analytical technique
Au, Pt, Pd	Lead collection fire assay
Cu, Li, Ni, Pb, S, Zn	4 acid digest (ICP-OES)
Ag, As, Bi, Cd, Co, Cs, Ge, In, Mo, Nb, Re, Sb, Se, Te, Ti	4 acid digest (ICP-MS)
F	Carbonate fusion/SIE
Al, Ca, Cr, Fe, K, Mg, Mn, Na, P, Si, Ti, V	Lithium borate fusion (ICP-OES)
Ba, Be, Ce, Dy, Er, Eu, Ga, Gd, Hf, Ho, La, Lu, Nd, Pr, Rb, Sc, Sm, Sn, Sr, Ta, Tb, Th, Tm, U, W, Y, Yb, Zr	Lithium borate fusion (ICP-MS)

Samp_R#	DH_No	East	North	Depth from	Depth to	Strat_Name	Lith_type	Samp_type	Samp_from	Au ppb	Ag ppm	Al2O3%	As ppm	Ba ppm	Be ppm	Bi ppm	CaO%
1946904	139458	763604	6257871	36.5	37.5	Amphibolite and Wandearah Formation	Amphibolite	Core	B1	13	0.35	14.97	52.4	183.8	3.9	0.04	3.65
1937386	178246	776838	6250511	24	26	Wandearah Formation	Carbonate	Cuttings	B1	0.5	0.51	4.63	197.2	153.6	2.2	0.59	20
1949042	23033	762660	6232342	30.9	31.9	Unnamed GIS unit	Carbonate	Drill core	B1	0.5	0.025	3.43	2.3	166.6	0.5	0.005	31.34
1949782	23729	776876	6249028	88.2	89.2	New Cornwall Member	Carbonate	Drill core	B1	0.5	1.11	4.87	85.8	218.1	1.5	0.32	18.18
1960880	30065	778298	6242989	158.35	159.4	Wandearah Formation	Carbonate	Drill core	B1	4	0.2	5.65	33	715.3	1.5	1.51	17.51
1960883	30051	778309	6244444	169.8	170.8	Wandearah Formation	Carbonate	Drill core	B1	0.5	0.3	8.19	49	2924.3	2.8	0.02	13.78
000005		736061	6169308	143.7	144.7		Felsic volcanic	Drill core	B1	0.5	0.025	12.66	3	702.9	3.1	0.04	5.73
1933082	23015	742809	6208841	14.3	15.3	Weetulta Formation	Felsic volcanic	Drill core	B1	0.5	0.025	9.14	2.6	136.1	5.3	0.03	0.32
1937389	23748	775431	6244530	16	18	Weetulta Formation	Felsic volcanic	Cuttings	B1	0.5	0.31	11.45	23.8	515.3	1.4	0.08	0.05
1937392	174055	777406	6244863	68	74	Weetulta Formation	Felsic volcanic	Cuttings	B1	0.5	0.7	3.68	61.2	137.1	1.7	0.1	22.73
1937395	23413	746513	6236028	7.19	8	Moonta Porphyry Member	Felsic volcanic	Drill core	B1	0.5	0.025	9.75	12	570.5	0.7	0.32	0.11
1937398	143253	751915	6208687	4	6	Wallaroo Group	Felsic volcanic	Cuttings	B1	0.5	0.2	13.85	2.6	325.6	3.2	0.07	0.9
1949037	23039	740506	6225440	25.6	26.6	Moonta Porphyry Member	Felsic volcanic	Drill core	B1	0.5	0.18	11.11	17.7	197	5.4	0.13	0.1
1949053	23730	775944	6248599	40.25	41.25	Mona Volcanics Member	Felsic volcanic	Drill core	B1	0.5	0.21	12.66	16.9	954.4	2.7	0.1	0.34
1949786	23723	777295	6249394	70	71	Mona Volcanics Member	Felsic volcanic	Drill core	B1	0.5	0.38	10.23	13.5	1185.5	1.8	0.58	0.14
1949794	23701	741389	6240671	42.8	43.8	Weetulta Formation	Felsic volcanic	Drill core	B1	0.5	0.25	11.75	25.6	1275	2.5	0.01	0.94
1960846	23028	733429	6208620	36.27	37.3	Wandearah Formation	Felsic volcanic	Drill core	B1	2	0.025	6.87	59	15	13.6	0.21	9.88
1937187	143973	753159	6166371	21	22	Unnamed	Gneiss	Drill core	B1	0.5	0.025	11.48	0.5	805	1.8	0.02	0.33
1937189	143974	745809	6165690	21	22	Unnamed	Gneiss	Drill core	B1	0.5	0.025	12.34	1.7	567.7	2.6	0.02	2.57
1937617	142693	757481	6214765	12	14	Wallaroo Group	Gneiss	Cuttings	B1	0.5	0.06	15.41	0.5	102.2	2.9	0.07	2.58
1946898	143186	746712	6202276	18	20	Wallaroo Group	Gneiss	Cuttings	B1	0.5	0.18	13.25	1.9	822.3	2.9	0.14	1.11
1946909	142632	739599	6204215	26	28	Wallaroo Group	Gneiss	Cuttings	B1	0.5	0.025	18.86	4.5	1237.2	4.6	0.05	5.31
1947179	143281	745443	6213456	18	20	Wallaroo Group	Gneiss	Cuttings	B1	0.5	0.07	20.33	11	572.6	3.3	0.05	0.02
1947184	143215	738441	6207270	16	18	Wallaroo Group	Gneiss	Cuttings	B1	0.5	0.025	18.91	0.5	196.9	3.8	0.02	4.03
1947188	142667	747311	6209650	10	12	Wallaroo Group	Gneiss	Cuttings	B1	0.5	0.13	11.01	5.8	108	1.9	0.04	0.64
1933070	139476	763729	6257846	71.5	72.5	Wandearah Formation	Granite	Drill core	B1	93	0.8	13.94	12	46.8	1.6	0.11	1.25
1933088	23197	764842	6266413	121	122	Tickera Granite	Granite	Drill core	B1	0.5	0.025	12.18	0.5	256.8	5.4	0.005	0.1
1933091	22717	750382	6219225	24	25	Arthurton Granite	Granite	Drill core	B1	0.5	0.025	14.86	1.7	1328.1	3	0.06	1.09
1937401	22676	750690	6160130	100	104	Arthurton Granite	Granite	Cuttings	B1	0.5	0.025	11.4	5.9	831.5	1.3	0.005	0.29
1937404	142663	751829	6208254	12	14	Arthurton Granite	Granite	Cuttings	B1	0.5	0.025	10.17	3.4	432.6	1.5	0.04	0.38
1937611	143971	751329	6162690	21	22	Unnamed	Granite	Drill core	B1	0.5	0.15	12.69	0.5	816.5	2.8	0.04	1.81
1937615	142692	757073	6214771	4	6	Arthurton Granite	Granite	Cuttings	B1	0.5	0.11	12.69	4.1	449.9	1.3	0.08	0.88
1946900	143188	746720	6204029	20	22	Arthurton Granite?	Granite	Cuttings	B1	0.5	0.025	14.16	5.8	618.1	2.9	0.07	0.12
1946911	143205	743608	6203678	12	14	Arthurton Granite	Granite	Cuttings	B1	0.5	0.025	15.97	1.5	2307.6	2.1	0.005	0.67

Samp R#	Cd ppm	Ce ppm	Co ppm	Cr ppm	Cs ppm	Cu ppm	Dy ppm	Er ppm	Eu ppm	F ppm	Fe2O3%	Ga ppm	Gd ppm	Ge ppm	Hf ppm	Ho ppm	In ppm	K2O%	La ppm	Li ppm	Lu ppm
1946904	0.24	25	79.8	69	11.65	74	5.13	2.85	1.73	1177	16.24	22.5	4.94	1.45	2.8	1.09	0.126	2.74	13.7	29.8	0.36
1937386	0.13	33.9	23.9	21	1.64	481	3.09	1.62	0.6	624	4.96	6.2	3.16	1.01	1.5	0.6	0.04	2.74	21.8	7.1	0.26
1949042	0.05	41.5	2.9	35	2.16	143	2.52	1.29	0.48	463	1.85	4.6	2.9	0.16	0.9	0.49	0.032	0.36	20.6	17	0.19
1949782	0.06	44.6	58.1	39	1.56	216	6.66	3.63	1.26	346	5.01	7.5	6.05	0.48	2.5	1.35	0.104	1.61	22.9	9.8	0.44
1960880	0.01	59.9	20.3	31	2.76	181	3.49	1.74	1.1	535	3.11	9.8	4.01	0.6	2.4	0.66	0.06	2.59	25.8	21.2	0.25
1960883	0.01	39.5	27.7	36	3.29	158	3.46	2	0.61	767	3.84	13	3.51	0.7	3.2	0.71	0.05	3.57	18.9	31	0.28
0000005	0.3	120.9	22.3	28	9.27	37	16.36	9	4.15	1160	14.38	24.1	16.9	0.9	11	3.32	0.15	2.5	57.7	20.4	1.3
1933082	0.01	6.5	3.7	10	2.46	32	3.04	2.84	0.17	474	1.83	22.1	1.08	0.68	4.4	0.79	0.0025	2.78	2.9	0.7	0.77
1937389	0.11	147.7	1.4	21	1.1	35	11.63	7.56	1.9	358	4.9	21.3	10.29	1.38	15.6	2.54	0.027	1.44	66.5	9.9	1.44
1937392	0.01	26.5	13	28	2.09	94	1.95	1.09	0.47	648	3.39	5.9	2.15	0.76	1.3	0.4	0.05	2.3	15.1	12.5	0.17
1937395	0.03	30.7	4.1	71	4.22	35	2.65	1.74	0.42	695	6.96	20.4	1.97	1.31	8	0.6	0.043	1.55	18.5	8.9	0.3
1937398	0.09	127.4	4.1	29	5.28	36	7.84	4.54	1.83	643	8.76	22.8	7.43	1.24	11.5	1.53	0.063	1.88	75.3	5.9	0.8
1949037	0.01	171.2	18.6	10	1.45	1180	19.4	14.25	2.3	2532	8.12	24.9	12.63	0.67	13.8	4.74	0.07	3.79	89.8	55.2	2.26
1949053	0.01	206.4	5.7	10	1.99	134	10.75	6.08	2.97	787	7.03	19.9	14.3	0.87	9.5	2.15	0.023	3.48	108.9	27.4	0.96
1949786	0.06	146.7	7	10	1.24	53	9.98	6.69	2.14	403	6.57	21.9	10.32	1.11	11.7	2.23	0.019	6.17	70.2	30.3	1.16
1949794	0.01	150.4	13.1	10	1.73	55	18.82	12.44	2.47	1548	12.85	28.4	16	0.82	15.8	4.2	0.028	6.8	80.5	48.8	1.98
1960846	1.4	1979.8	179.8	83	0.39	1085	176.08	86.56	22.71	1809	38.69	15.7	228.5	4.4	7.3	33.91	0.2	0.18	868.3	67.3	8.15
1937187	0.05	97.5	9.1	25	2.06	35	9.4	5.57	2.02	180	7.32	17.6	8.72	0.49	11.3	1.91	0.063	4.77	53.7	7.4	0.91
1937189	0.08	115	14.5	29	1.07	26	16.12	9.45	2.72	814	8.3	19.2	14.62	0.66	12.4	3.3	0.118	2.76	53.1	10.4	1.41
1937617	0.01	296.6	25	35	1.59	76	28.92	15.37	6.95	686	10.32	23.6	30.42	1.63	5.4	6.01	0.307	0.63	150	6.1	2.13
1946898	0.01	231.3	4.3	172	3.01	17	16.29	8.92	3.45	345	3.48	18.7	16.86	0.81	13.4	3.36	0.101	4.78	136.5	19.6	1.27
1946909	0.01	130	29.2	132	1.41	93	17.78	12.2	3.49	1201	12.72	24.4	16.62	1.02	2.9	4.29	0.107	1.28	62.3	21.1	1.7
1947179	0.01	35.1	6.5	214	12.97	59	2.36	1.48	0.45	2405	3.42	31.5	1.91	2.06	4.1	0.52	0.105	7.24	15	33	0.28
1947184	0.01	58.7	31.6	263	3.64	32	3.06	1.73	0.85	2508	8.96	20.1	3.12	1.16	0.9	0.61	0.212	2.17	30.1	18.9	0.3
1947188	0.01	50.7	1.2	177	0.46	5	7.28	4.54	0.95	116	1.93	14.5	5.48	0.87	10	1.59	0.008	0.43	25.4	11.5	0.69
1933070	0.01	14.1	36.4	53	1.48	2254	1.83	1.31	0.23	750	3.47	18.3	1.53	0.9	5.8	0.43	0.05	1.01	6.6	16.1	0.31
1933088	0.01	191.9	3.3	10	3.44	16	14.64	9.98	1.19	762	3.23	18.1	10.47	0.42	7	3.21	0.0025	3.72	46.4	13.5	2.02
1933091	0.12	218.2	11.3	10	3.53	22	23.37	11.64	4.09	1428	6.24	19.8	24.71	0.68	9.6	4.53	0.045	6.08	95.8	23.4	1.51
1937401	0.05	120.3	6.6	10	2.22	18	10.67	5.96	2.58	491	5.61	16.8	10.93	0.98	11.7	2.17	0.062	6.27	57.9	13.8	0.93
1937404	0.01	50.1	5.7	64	2.13	9	1.56	1.16	0.55	962	6.92	18.1	1.77	0.89	4.4	0.36	0.036	2.32	25.5	4.7	0.27
1937611	0.01	276.1	9.6	10	2.6	25	34.46	18.23	6.73	358	5.53	17.5	33.66	1.4	13.6	6.89	0.049	3.88	115.8	6.6	2.47
1937615	0.01	19.2	2.3	88	1.08	7	4.45	3.08	0.47	256	2.4	18.4	2.39	0.61	8.8	0.99	0.027	0.64	10.1	4.1	0.48
1946900	0.01	1345.8	1.6	97	2.78	15	21.57	8.97	5.77	261	1.13	13	36.71	3.06	6.2	3.81	0.011	6.77	742.4	4.3	1.13
1946911	0.01	58.8	1.8	95	3.71	7	2.49	1.41	1.01	436	3.43	21.1	2.9	0.74	8.5	0.5	0.035	4.95	39.8	1	0.29

Samp R#	MgO%	MnO%	Mo ppm	Na2O%	Nb ppm	Nd ppm	Ni ppm	P2O5%	Pb ppm	Pd ppb	Pr ppm	Pt ppb	Rb ppm	Re ppm	S ppm	Sb ppm	Sc ppm	Se ppm	SiO2%	Sm ppm	Sn ppm	Sr ppm
1946904	5.81	0.17	2.1	3.82	7.8	16.6	68	0.15	33	2	3.71	2	161.3	0.002	128	0.16	39	2.4	47.79	4.15	5	43.2
1937386	14.55	0.31	5.6	0.13	2.4	13.3	12	0.04	23	0.05	3.56	0.05	87.8	0.002	99	1.45	10	2	19.59	3.21	2	142.3
1949042	3.21	0.85	0.9	1.41	0.95	17.4	2	0.19	2.5	0.05	4.65	0.05	26.5	0.002	25	0.025	10	1.7	28.42	3.41	0.05	61.3
1949782	10.79	1.08	2.1	0.85	5.22	18.9	31	0.1	24	0.05	4.8	0.05	60.8	0.007	8616	1.1	14	3.9	29.12	5.03	1	37.2
1960880	12.21	0.8	1.1	0.03	3.82	22.8	49	0.19	7	0.05	6.11	2	87.2	0.002	4171	1.3	1	0.025	30.26	4.66	2	30.6
1960883	9.63	0.6	1	0.1	7.86	16.6	46	0.1	18	0.05	4.48	1	157.9	0.002	2337	0.88	1	0.025	37.93	3.53	2	31.7
0000005	2.07	0.2	2.2	2.68	31.36	67	29	0.83	18	0.05	15.73	0.05	134.9	0.002	1033	0.3	27	0.025	56.44	15.85	4	149.2
1933082	0.13	0.01	0.5	3.04	91.9	2.3	0.05	0.06	2.5	0.05	0.67	0.05	246.2	0.002	137	0.39	10	0.025	80.42	0.69	15	16.9
1937389	0.19	0.01	3.1	0.11	36.3	56.7	2	0.12	8	0.05	14.96	0.05	34.8	0.002	303	0.43	15	8	75.71	11.63	3	14.2
1937392	15.2	0.43	2.4	0.1	1.9	10.9	4	0.09	17	0.05	2.96	0.05	73.5	0.002	2848	1.08	10	2	16.66	2.26	1	83.9
1937395	1.05	0.01	5.3	0.79	19.5	12.3	7	0.05	10	0.05	3.42	0.05	81.5	0.007	377	0.49	12	8.6	69.65	2.35	5	105.9
1937398	0.41	0.01	1.2	4.54	43.9	48.2	19	0.08	27	0.05	13.76	0.05	133.2	0.002	272	0.32	15	3.6	65.58	9.31	7	230.2
1949037	6.38	0.01	0.2	0.2	35.96	66.5	23	0.11	2.5	0.05	18.93	0.05	126.6	0.002	25	0.12	17	2.2	64.14	12.19	4	8
1949053	1.87	0.03	0.9	2.83	29.53	89.9	16	0.09	2.5	0.05	23.65	0.05	89.1	0.002	25	0.24	14	1.4	68.65	17.51	1	35.3
1949786	1.2	0.04	0.3	0.06	31	61.1	5	0.08	8	0.05	16.57	0.05	122.2	0.002	1075	0.57	13	1.7	73.25	12.15	2	9.6
1949794	2.91	0.01	0.8	0.2	27.9	67.1	22	0.74	9	0.05	17.91	0.05	204.9	0.003	25	0.3	13	2.7	60.41	14.51	4	33.3
1960846	3.45	0.07	20.9	0.53	124.68	1012.2	1016	7.64	88	23	240.39	14	7.5	0.002	834	2.65	1	26	25.76	226.83	2	146.8
1937187	0.37	0.01	0.5	2.31	29.5	46.7	9	0.15	2.5	0.05	12.07	0.05	242.3	0.002	113	0.025	10	3.6	69.43	9.53	11	120.8
1937189	1.35	0.06	0.8	3.1	36	59.9	12	0.2	15	0.05	14.35	0.05	111.6	0.002	69	0.025	18	0.9	66.47	13.82	8	174.1
1937617	1.81	0.03	0.7	4.83	18.7	133.4	14	0.25	6	0.05	34.1	0.05	28.1	0.006	324	0.24	60	0.025	56.78	27.67	3	804.5
1946898	0.24	0.01	0.9	3.05	38.8	93.3	5	0.14	27	0.05	24.88	0.05	250.9	0.002	168	0.07	13	4.7	70.02	18.68	6	273.7
1946909	1.57	0.1	0.9	3.64	7.5	64	63	0.7	54	0.05	15.6	0.05	52.5	0.003	279	0.07	23	3.9	48.85	12.65	2	1109.3
1947179	2.13	0.01	0.8	0.34	26.2	10.3	12	0.04	9	1	3.06	0.05	479.8	0.003	243	0.36	31	3.5	59.48	2.16	10	12.5
1947184	4.91	0.05	0.3	4.68	4.3	17.6	90	0.05	8	6	5.18	9	270	0.002	143	0.025	45	2.4	50.49	3.37	3	107.2
1947188	0.11	0.01	0.5	5.66	25.7	23.2	2	0.04	2.5	0.05	6.03	0.05	23.7	0.006	58	0.22	10	4.7	77.55	5.08	5	49.3
1933070	1.69	0.02	1.5	7.05	13.1	7	54	0.28	49	0.05	1.76	2	58	0.002	640	0.22	10	0.025	69.16	1.46	5	32.6
1933088	0.68	0.01	0.2	2.45	52.5	36.3	3	0.08	2.5	0.05	9.7	0.05	354.4	0.002	54	0.33	10	4.4	74.86	9.47	9	19.5
1933091	0.98	0.03	0.8	2.77	26.1	101.5	4	0.23	2.5	0.05	24.77	0.05	279.3	0.002	205	0.27	12	0.025	65.43	22.39	5	121.9
1937401	1.1	0.01	1.8	1.69	30.1	54.8	6	0.29	12	0.05	13.9	0.05	225.4	0.002	25	0.12	10	0.025	71.84	11.67	5	70.4
1937404	0.6	0.01	0.9	2.95	7.1	19.1	7	0.07	2.5	0.05	5.54	0.05	103.4	0.002	228	0.22	10	4.4	70.65	2.86	6	68.6
1937611	0.78	0.02	0.7	2.77	36.3	150	3	0.12	6	0.05	36.4	0.05	177.3	0.008	102	0.025	19	2.2	69.04	36.08	8	256.3
1937615	0.47	0.01	1.9	5.38	21.6	9.6	3	0.015	12	0.05	2.52	0.05	23.4	0.005	200	0.49	11	5.2	73.27	2.24	4	54.7
1946900	0.23	0.01	0.5	1.5	36.7	460.2	3	0.22	42	0.05	146.04	0.05	423	0.005	257	0.27	11	6.7	72.73	65.04	3	138.8
1946911	0.21	0.01	0.2	4.44	16.1	22.5	0.05	0.06	16	0.05	6.22	0.05	148.3	0.003	225	0.09	10	3.9	67.62	4.09	3	212.5

Samp_R#	Ta ppm	Tb ppm	Te ppm	Th ppm	TiO2%	Tl ppm	Tm ppm	U ppm	V ppm	W ppm	Y ppm	Yb ppm	Zn ppm	Z
1946904	0.5	0.8	0.05	2.9	1.58	1.28	0.43	6.08	381	0.05	29.6	2.38	307	
1937386	0.4	0.47	0.2	6.6	0.24	0.8	0.33	2.36	54	0.05	18.7	1.36	84	
1949042	0.2	0.39	0.05	4.78	0.15	0.07	0.19	1.03	13	1	15.3	1.07	62	
1949782	0.5	1.04	0.07	7.83	0.25	2.33	0.52	5.38	44	1	36.1	3.01	51	
1960880	0.8	0.57	0.05	8.04	0.26	1.31	0.2	2.2	30	2	19.8	1.54	28	
1960883	0.8	0.55	0.05	11.85	0.36	0.58	0.26	2.23	46	4	20.7	1.92	27	
0000005	2.5	2.5	0.05	11.5	2.12	0.76	1.33	3.13	31	1	90.9	8.42	163	
1933082	15.1	0.31	0.05	17.52	0.23	0.36	0.57	2.26	17	0.05	28.3	4.27	169	
1937389	4.8	1.59	0.11	39.74	0.63	0.09	1.27	6.86	37	7	68.9	8.43	14	
1937392	0.4	0.31	0.05	4.64	0.24	0.9	0.2	2.2	59	1	11.9	1	27	
1937395	1.6	0.36	0.05	12.51	0.99	0.36	0.3	4.97	171	3	16.6	2.08	47	
1937398	3.6	1.12	0.22	19.83	0.76	0.36	0.74	11.54	92	2	41.4	4.91	32	
1949037	3.7	2.46	0.16	36.17	0.48	0.28	2.16	11.06	15	2	148.6	13.23	113	
1949053	3	1.84	0.05	41.95	0.43	0.29	0.97	5.31	20	4	53.5	6.08	37	
1949786	3.1	1.54	0.06	27.06	0.4	0.8	1.07	6.64	13	4	62.5	7.02	43	
1949794	3.4	2.68	0.14	27.54	0.71	0.34	1.9	9.54	67	1	141	11.82	24	
1960846	4.6	30.34	16.1	23.71	0.49	0.04	11.28	148.78	441	5	1003	61.05	249	
1937187	2.4	1.36	0.05	16.67	0.69	1.08	0.85	3.11	66	0.05	54.5	5.62	30	
1937189	2.8	2.34	0.05	22.69	1.03	0.36	1.36	4.22	117	1	97.8	8.56	79	
1937617	1.4	4.41	0.05	13.95	2.44	0.06	2.28	8.13	432	2	155.6	13.73	35	
1946898	2.8	2.55	0.07	22.5	0.63	1.22	1.38	7.29	32	3	91.8	8.52	60	
1946909	0.6	2.52	0.05	8.59	1.39	0.12	1.85	1.94	312	0.05	156.4	11.09	97	
1947179	2.8	0.35	0.05	27.45	0.8	1.03	0.26	5.43	216	1	14.7	1.88	27	
1947184	0.2	0.47	0.05	1.43	0.48	0.29	0.28	0.72	211	0.05	18.7	1.64	38	
1947188	2.1	0.98	0.05	16.83	0.31	0.06	0.75	2.37	15	0.05	45.5	5.02	13	
1933070	1.6	0.26	0.2	16.44	0.56	0.1	0.26	3.55	70	3	13	1.58	197	
1933088	6.7	1.97	0.05	77.07	0.28	0.72	1.64	8.9	21	0.05	103.7	11.08	23	
1933091	2.7	3.61	0.05	27.91	0.59	0.72	1.59	9.83	37	2	148.6	9.82	37	
1937401	2.4	1.64	0.05	12.36	0.54	0.98	0.89	1.62	21	3	59.4	5.41	45	
1937404	0.8	0.23	0.11	7.84	0.44	0.36	0.23	2.67	77	2	9.5	1.41	13	
1937611	2.6	5.33	0.05	21.98	0.76	0.31	2.84	3.07	31	0.05	172.4	17.12	32	
1937615	1.8	0.56	0.06	13.27	0.47	0.14	0.52	2.07	56	2	26.6	3.47	21	
1946900	3.5	4.1	0.05	196.29	0.21	0.89	1.24	25.02	32	3	94.9	7.25	26	
1946911	0.9	0.41	0.05	12.41	0.52	0.37	0.22	2.52	28	0.05	10.9	1.46	23	

Kym Michael Custance Identifying IOCG pathfinder elements

Samp_R#	DH_No	East	North	Depth from	Depth to	Strat_Name	Lith_type	Samp_type	Samp_from	Au ppb	Ag ppm	Al2O3%	As ppm	Ba ppm	Be ppm	Bi ppm	CaO%
1946913	141135	759129	6258811	39	40	Tickera Granite	Granite	Cuttings	B1	0.5	0.1	13.6	2.7	930.6	3.9	0.06	0.47
1947182	142651	738430	6206920	16	18	Arthurton Granite	Granite	Cuttings	B1	0.5	0.09	14.96	1	2243.4	2	0.05	3.8
1947186	143230	746607	6206986	4	6	Arthurton Granite	Granite	Cuttings	B1	0.5	0.19	12.04	2.9	964.5	1.8	0.14	0.6
1949014	139483	758529	6267471	179.6	180.6	Tickera Granite	Granite	Drill core	B1	0.5	0.11	12.96	16.7	748.5	3	0.02	0.64
1949028	22991	737229	6222021	27.2	28.2	Hiltaba Suite	Granite	Drill core	B1	0.5	0.1	12.54	18.6	292.5	4.5	0.05	0.52
1949032	23021	735724	6234746	22.86	23.85	Tickera Granite	Granite	Drill core	B1	0.5	0.23	16.93	18.7	1324.8	3.2	0.4	0.16
1949035	23023	734929	6212970	58.5	59.5	Weetulta Formation	Granite	Drill core	B1	0.5	0.025	16.15	21.5	479.5	10	0.11	0.24
1949047	23036	758334	6224301	25.35	26.35	Arthurton Granite	Granite	Drill core	B1	0.5	0.025	14.18	17.2	1369.5	3.1	0.03	0.33
1949048	23052	752291	6225060	113.7	114.7	Arthurton Granite	Granite	Drill core	B1	0.5	0.025	13.33	15.4	1508.1	3.1	0.13	0.27
1949808	23031	755663	6223745	82	83	Arthurton Granite	Granite	Drill core	B1	0.5	0.025	13.49	8	2255.5	3	0.01	0.53
1960843	143972	744329	6163730	39.5	40.5	Unnamed GIS unit	Granite	Drill core	B1	0.5	0.3	11.73	16	985.5	1.6	0.005	4.38
1960889	23051	754929	6224206	83.45	84.5	Tickera Granite	Granite	Drill core	B1	0.5	0.025	13.6	7	1609.1	3.1	0.04	1.27
1949010	30059	778561	6248868	108.95	109.95	Willamulka Basalt Member	Mafic extrusive	Drill core	B1	0.5	0.69	1.02	33.7	87	0.5	0.17	26.43
1949020	30055	778282	6248945	101.85	102.85	Willamulka Basalt Member	Mafic extrusive	Drill core	B1	1	0.31	15.3	15.9	151.3	0.8	0.81	1.53
0000002		734268	6169901	144.1	145.1		Mafic intrusive	Drill core	B1	0.5	0.025	15.12	0.5	534.7	2.2	0.13	3.33
1937182	29692	776523	6266268	24.6	25.6	Bute Metadolerite	Mafic intrusive	Drill core	B1	3	0.025	15.97	4.3	240	0.5	0.07	14.52
1937185	29752	783721	6248572	440	441	Bute Metadolerite	Mafic intrusive	Drill core	B1	0.5	0.025	14.74	6	104	0.25	0.07	8.36
1937609	143970	747769	6164961	30	31	Unnamed	Mafic intrusive	Drill core	B1	0.5	0.06	16.45	2.1	628.4	1.5	0.04	5.19
1949013	23699	770228	6235144	179.9	180.9	Unnamed GIS unit	Mafic intrusive	Drill core	B1	2	0.17	11.39	9.3	145.9	4.8	0.08	2.12
1949040	23054	774179	6231246	406.1	407.1	Wandilta Amphibolite Member	Mafic intrusive	Drill core	B1	0.5	0.025	12.98	17.9	464.7	0.7	0.005	5.96
1949044	23663	763406	6250765	74.06	75	Unnamed GIS unit	Mafic intrusive	Drill core	B1	7	1.5	14.47	30	260.3	2.2	0.08	0.53
1949797	30047	784258	6246312	338.9	339.9	Bute Metadolerite	Mafic intrusive	Drill core	B1	0.5	0.1	14.62	75.5	121.3	0.7	2.12	4.02
1949802	30064	782811	6239489	505.05	506	Willamulka Basalt Member	Mafic intrusive	Drill core	B1	12	0.025	12.72	63	140.1	1.6	0.39	7.98
1949819	30048	784207	6244524	426.34	427.3	Bute Metadolerite	Mafic intrusive	Drill core	B1	32	3.1	13.34	44	186.3	1	1.49	6.89
1960891	23697	770244	6235564	225.7	226.7	Unnamed GIS unit	Mafic intrusive	Drill core	B1	2	0.025	13.2	9	121.5	5.2	0.09	2.15
1933075	23265	747787	6247265	3.05	4.26	Oorlano Metasomatite	Metasomatite	Drill core	B1	0.5	0.025	7.73	4.3	455.2	1.8	0.05	0.86
1933085	143968	761879	6160620	209.3	210.3	Oorlano Metasomatite	Metasomatite	Drill core	B1	0.5	0.31	10.63	9.4	225	3.1	0.04	8.18
1946895	139474	771179	6264821	54.6	55.6	Oorlano Metasomatite	Metasomatite	Drill core	B1	0.5	0.025	15.8	13.4	414.5	3.1	0.13	0.93
1949008	30060	777888	6243276	137.3	138.3	Oorlano Metasomatite	Metasomatite	Drill core	B1	1	0.21	2.26	31.3	331.3	0.5	0.15	18.38
1949050	23674	759856	6246503	116.48	117.5	Oorlano Metasomatite	Metasomatite	Drill core	B1	0.5	0.025	9.43	17.9	293.9	0.6	0.15	0.28
1949055	139488	759504	6267846	157.3	158.3	Oorlano Metasomatite	Metasomatite	Drill core	B1	8	0.09	10.43	20.7	301.1	11.1	1.3	6.94
1949812	23669	751171	6243568	116.74	117.8	Oorlano Metasomatite	Metasomatite	Drill core	B1	0.5	0.025	16.95	8	555.1	1.2	0.13	0.58
1960849	22962	767704	6214605	134	136	Oorlano Metasomatite	Metasomatite	Cuttings	B1	0.5	0.025	15.31	9	657.8	5.7	0.03	1.39
1933073	23389	772223	6241274	478.8	479.7	Wandearah Formation	Sediment	Drill core	B1	0.5	0.025	14.77	6.8	845.5	2	0.08	3.69
1933078	139489	734539	6211211	33.1	34.1	Wandearah Formation	Sediment	Drill core	B1	0.5	0.025	18.25	5.1	833.5	3.8	0.06	0.51

Samp R#	Cd ppm	Ce ppm	Co ppm	Cr ppm	Cs ppm	Cu ppm	Dy ppm	Er ppm	Eu ppm	F ppm	Fe2O3%	Ga ppm	Gd ppm	Ge ppm	Hf ppm	Ho ppm	In ppm	K2O%	La ppm	Li ppm	Lu ppm	
1946913	0.01	208	5.8	131	1.69	41	14.61	8.46	2.58	924	4.26	17.8	15.52	1.14	10.6	2.95	0.037	5.49	100.2	6.5	1.3	
1947182	0.01	43.2	2	110	2.23	8	2.31	1.34	1.26	835	3.61	24.4	2.65	0.35	14.2	0.51	0.06	4.31	27	13.9	0.26	
1947186	0.01	117.7	1.3	157	1.4	13	8.61	4.55	1.87	286	1.47	17.6	9.12	0.9	8.1	1.74	0.057	4.72	64	0.8	0.69	
1949014	0.02	195.7	7.6	10	1.24	21	10.36	5.95	2.13	1027	5.18	18.2	11.68	0.29	11.2	2.08	0.063	5.24	99.6	17.5	0.87	
1949028	0.01	415.6	5.6	10	0.49	9	10.37	4.81	3.36	580	4.18	18.1	16.58	0.55	12.5	1.9	0.01	2.14	225	21.8	0.75	
1949032	0.01	108.5	1.9	10	1.2	26	13.87	9.04	1.54	284	1.94	20.9	9.99	0.43	23.2	3.02	0.008	8.37	53.2	6	1.39	
1949035	0.01	97.1	19.3	91	11.55	13	4.78	2.95	1.3	2216	11	24	5.37	0.91	3.7	1.03	0.115	6.8	51.3	41.7	0.42	
1949047	0.06	97.3	4.6	10	4.54	44	15.93	10.48	1.98	964	3.93	19.2	11.27	0.77	10.9	3.55	0.046	6.86	43.6	17.4	1.44	
1949048	0.02	137.1	2.1	10	5.49	84	10.35	6.51	1.69	1178	5.06	18.1	8.19	0.65	9.9	2.25	0.047	9.65	73.8	9.2	0.93	
1949808	0.01	183.9	7.5	26	1.65	25	7.4	3.9	2.32	696	4.95	17.1	9.3	0.6	10.4	1.42	0.02	6.7	93.7	15.8	0.62	
1960843	0.1	73.8	12.1	32	4.53	72	6.5	3.8	1.38	631	3.04	15.9	6.3	0.7	9	1.36	0.05	6.86	34.9	7.2	0.62	
1960889	0.01	207	8.1	10	2.68	34	10.8	6.25	2.63	1362	5.12	17.9	11.68	1	11.7	2.15	0.03	6.32	91.5	34	0.81	
1949010	0.06	15.1	19	38	0.26	602	2.24	1.1	0.52	222	4.12	2.3	2.35	0.025	0.5	0.43	0.082	0.66	7.8	1.4	0.12	
1949020	0.04	39.1	21.8	101	0.42	554	2.3	1.42	1.07	403	10.58	19.5	3.2	0.56	2.8	0.46	0.035	1.69	16.7	34.7	0.29	
0000002	0.1	93.1	28	76	6.71	72	5.67	3.18	1.45	785	9.34	22.6	6.44	0.7	4.8	1.16	0.07	2.8	48.5	32.9	0.46	
1937182	0.01	10.5	45.3	581	1.8	94	1.96	1.21	0.53	157	6.77	12.8	1.68	1.75	0.8	0.41	0.027	1.19	7.2	32	0.2	
1937185	0.01	14.7	32.6	265	1.02	68	3.75	2.25	0.82	549	11.2	18.4	3.16	0.87	2.5	0.78	0.04	0.7	7.1	57.4	0.39	
1937609	0.07	93.6	36.2	33	3.1	66	8.85	4.74	2	1119	10.09	19.7	9.13	1.39	6	1.87	0.072	2.05	44.8	27.4	0.71	
1949013	0.03	148.6	12.9	23	42.17	19	14.02	8.22	3.99	2311	12.01	23.7	15.01	1.05	6.9	2.97	0.139	3.42	75.2	64.5	1.21	
1949040	0.03	46.1	32.4	256	10.36	/2	3.76	2.04	1	1706	11.94	23.8	4.21	0.49	2.9	0.74	0.044	3.52	25.3	47.7	0.29	
1949044	0.05	42.8	66.3	230	11.//	84	3.54	2.17	0.84	15/3	9.97	19.1	3.47	0.98	2.1	0.77	0.049	3.54	22.8	62.7	0.32	
1949797	0.02	18.2	47.9	179	1.99	209	3.75	2.33	0.77	487	13.37	17.5	3.43	0.42	1.8	0.8	0.087	1.53	8.2	68	0.33	
1949802	0.01	73.9	47.6	219	3.83	331	4.57	2.56	1 22	200	15.04	19.7	5.43	0.9	3	0.91	0.07	3.27	40.6	46.3	0.34	
1949819	0.1	20.9	04.1 29.2	10	4.33	10995	4.8	2.83	2.12	00C	10.04	19.4	4.4Z	0.8	2.0	0.90	0.33	2.05	9	55	0.45	
1900091	0.1	145.0 25.2	20.5	10	0.99	26	2 26	1 59	0.55	410	2 65	20.2	2 12	0.11	15 5	2.21	0.04	2.95	12.2	0.7	0.70	
1933075	0.15	11.6	23.9	20	2.63	20	2.30	1.58	0.55	410	13 16	9.4 21 /	1 65	0.11	5.0	0.5	0.027	2.90	5.0	21.6	0.30	
19/6895	0.01	53	13.6	303	4.54	55	2.42	1.55	1.05	2652	11.10	10.3	3.54	0.71	1.8	0.52	0.050	2.50	28.9	121.0	0.55	
19/9009	0.01	1/1.8	43.0	37	0.58	35	2.57	1.00	0.78	356	3 61	5.2	2 55	0.71	1.0	0.01	0.031	0.71	63	121.0	0.20	
19/9050	0.01	38.0	5.8	22	0.30	96	1.9	1.15	0.78	360	2 32	9.2	2.55	0.13	5.4	0.45	0.031	2 1	19.6	15 /	0.14	
1949055	0.01	632.6	34	66	0.91	135	8 41	1.2 4 21	2 21	539	2.52	12.9	12.05	0.12	3.9	1 63	0.011	2.1	356.4	11 5	0.15	
1949812	0.05	121.3	19.1	71	3.2	25	3.03	2.09	1 36	1536	6.84	22.5	4 63	0.51	10.4	0.67	0.475	10 56	88.3	41	0.55	
1960849	0.1	103	74	71	15 46	23	6 92	<u> </u>	1.50	2306	9 49	22.2	05 6 89	1 1	3 5	1 45	0.01	8 08	54.8	71 27 4	0.55	
1933073	0.01	55.8	9,3	50	3,39	18	4.89	2.63	1.23	1255	7.09	21	4,99	0.86	4.4	0.96	0.063	4.21	34	47.8	0.41	
1933078	0.01	168.3	18.8	82	2.65	10	5.15	2.28	1.49	2504	9.84	31.1	8.39	1.53	4.2	0.92	0.018	8.64	83.4	23.6	0.36	
Samp R#	MgO%	MnO%	Mo ppm	Na2O%	Nb ppm	Nd ppm	Ni ppm	P2O5%	Pb ppm	Pd ppb	Pr ppm	Pt ppb	Rb ppm	Re ppm	S ppm	Sb ppm	Sc ppm	Se ppm	SiO2%	Sm ppm	Sn ppm	Sr ppm
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1946913	0.5	0.02	0.9	2.49	35.9	89.1	3	0.13	107	0.05	23.93	0.05	318.9	0.005	269	0.1	10	4	69.6	17.89	4	180.8
1947182	0.55	0.01	0.7	4.05	19	17.8	2	0.015	9	0.05	4.7	0.05	172	0.003	246	0.17	12	3.7	61.95	3.32	4	217.3
1947186	0.23	0.01	0.5	2.84	25.3	47.2	2	0.04	2.5	0.05	12.67	0.05	204.9	0.002	97	0.3	12	3.5	75.13	9.21	5	69.4
1949014	1.36	0.05	1.1	2.85	30.78	76.8	4	0.14	10	0.05	21.6	0.05	259.8	0.002	541	0.12	13	3	67.46	14.08	5	56.9
1949028	1.08	0.01	0.9	4.86	30	149.3	1	0.08	8	0.05	44.09	0.05	102.6	0.002	83	0.08	11	1.1	71.45	23.79	5	48.3
1949032	0.75	0.01	0.2	3.86	73.47	45.7	3	0.005	11	0.05	12.6	0.05	274.6	0.003	61	0.22	10	1.7	64.89	9.49	7	98.4
1949035	3.65	0.07	0.6	0.63	11.77	40.6	37	0.16	2.5	0.05	11.03	0.05	244.9	0.002	65	0.22	19	3.2	55.11	7.14	4	45.6
1949047	0.85	0.03	0.7	2	25.22	45	6	0.19	7	0.05	11.17	0.05	253.9	0.003	159	0.25	12	3	69.27	9.63	5	267.9
1949048	0.52	0.01	0.7	0.23	25.46	46.7	0.05	0.23	2.5	0.05	14.11	0.05	350.7	0.002	107	0.23	11	3.3	68.2	8.56	9	57.4
1949808	1.72	0.08	0.5	2.07	21.57	70.9	4	0.18	2.5	0.05	20.07	0.05	263.7	0.002	135	0.13	11	0.025	69.01	11.91	3	70.8
1960843	1.39	0.07	2.4	1.59	18.57	32.6	29	0.16	44	0.05	8.56	1	226.7	0.002	310	0.44	1	0.025	66.29	6.67	4	105.4
1960889	1.67	0.03	0.5	2.2	24.95	85.9	25	0.2	10	0.05	23.73	0.05	209.3	0.002	374	0.18	10	0.025	67.09	15.09	4	108.8
1949010	17.73	0.98	2	0.08	1.54	6.7	7	0.005	10	0.05	1.63	0.05	20.9	0.014	6551	0.94	10	1.5	5.26	1.85	0.05	67.6
1949020	4.97	0.18	1.4	4.15	9.97	19.4	33	0.16	20	1	4.59	1	42.2	0.003	1285	0.75	32	2.3	53.59	4.09	2	24.2
0000002	3.25	0.14	0.2	4.25	14.16	38.6	30	0.16	6	0.05	10.24	0.05	142.8	0.002	2652	0.38	22	0.025	60.24	7.32	3	105.2
1937182	9.06	0.17	0.4	1.38	1.9	6.5	118	0.06	2.5	9	1.53	3	54.8	0.002	96	0.23	52	0.025	48.8	1.59	1	178.1
1937185	5.63	0.19	0.8	3.89	2.3	8.7	61	0.18	2.5	11	1.93	11	25.3	0.002	565	0.22	41	0.025	44.05	2.51	3	79.6
1937609	5.17	0.1	0.8	3.49	15.9	45.5	80	0.3	15	0.05	11.3	0.05	91.9	0.007	94	0.07	21	1.5	50.98	9.65	7	363.3
1949013	4.5	0.08	0.2	1.31	15.36	73.3	2	0.91	8	0.05	18.38	0.05	250.8	0.006	155	0.46	39	0.8	55.48	15.4	3	33.4
1949040	6.6	0.13	0.5	2.06	5.84	19.3	60	0.07	2.5	1	5.11	3	330.9	0.002	372	0.08	45	0.025	50.17	4.45	2	97.4
1949044	6.79	0.11	0.3	1.41	5.34	17.5	74	0.17	6	8	4.79	5	332.8	0.002	25	0.25	41	2.2	53.56	3.65	2	27.2
1949797	6.98	0.1	0.3	1.47	1.96	10.9	86	0.09	2.5	0.05	2.44	2	66.8	0.004	1461	0.42	39	2.7	49.59	3.03	0.05	28.6
1949802	6.83	0.41	0.4	0.4	2.76	30.2	108	0.09	9	6	8.11	6	130	0.002	2064	1.18	35	0.025	44.98	6.22	3	16.4
1949819	5.92	0.21	0.7	2.11	6.09	12.6	90	0.11	2.5	1	2.74	3	69.4	0.002	11/41	6.03	43	0.025	47.83	3.61	2	94.8
1900891	4.40	0.04	0.2	0.92	10.33	10.9	30 2	0.14	19	0.05	15.92	0.05	162.2	0.002	200	0.00	40	0.025	20.23	2 20	10	10.5
1022095	2.05	0.01	2.4	2.27	9.5	10.0 E 2	12	0.14	25	0.05	1.27	0.05	211.2	0.002	140	0.21	10	4.4	60.05 E2.22	1.39	19	67.5 40.2
1935085	2.05	0.00	0.7	1 56	24.0	5.5 72.2	145	0.10	2.5	0.05	6.08	0.05	211.2	0.002	1709	2.25	10	2.7	JZ.JZ	1.52	2	40.5
1940895	12 31	0.5	2.5	0.02	0.72	23.3 8 1	145	0.1	2.5	0.05	1 9	, 0.05	303.4	0.007	6075	0.39	49	1 8	32.00	4.4 2.25	3	40.5
1949050	1 82	0.00	0.5	2 9	4.77	16.6	7	0.00	2.5	0.05	1.5	0.05	77 8	0.002	25	0.35	10	0.025	78 55	2.25	2	18.7
1949055	4 56	0.04	0.05	2.5	18.07	178.6	, 133	0.10	7	3	59.69	0.05	197.2	0.002	25	0.625	10	2.4	46 77	2.05	11	88.5
1949812	5.06	0.42	0.05	0.15	16.07	63.8	39	0.23	, 9	0.05	18 46	2	473	0.002	176	0.05	11	0.025	56.3	9 12	5	23.6
1960849	1 94	0.04	0.05	0.23	13.05	39.5	53	0 19	10	0.05	11 03	- २	290 5	0.002	218	0.23	13	0.025	59 37	7 58	6	20.0
1933073	5.12	0.27	0.05	0.1	12.2	27.1	16	0.16	2.5	0.05	7.49	0.05	198.9	0.002	362	0.73	13	0.025	54.54	5.36	Q.	10.6
1933078	2.52	0.01	1.4	1.46	15	70.9	54	0.31	2.5	0.05	18.83	0.05	225	0.002	25	0.25	18	2.7	53.61	11.56	9	45.2

Raw whole	rock geoch	iemical data	a: Central Y	orke Penins	sula									
Samp_R#	Ta ppm	Tb ppm	Te ppm	Th ppm	TiO2%	Tl ppm	Tm ppm	U ppm	V ppm	W ppm	Y ppm	Yb ppm	Zn ppm	Zr ppm
1946913	2.9	2.27	0.05	39.63	0.56	0.86	1.29	14.09	52	0.05	86	8.33	31	399
1947182	1.1	0.35	0.05	12.22	0.55	0.42	0.22	2.77	26	0.05	11.1	1.49	31	589
1947186	2	1.33	0.05	16.57	0.4	0.49	0.73	3.32	30	2	45.3	4.63	17	303
1949014	2.9	1.68	0.07	50.2	0.75	0.59	0.91	8.97	43	2	59.3	5.6	21	449
1949028	3.3	1.99	0.06	50.71	0.65	0.2	0.74	7.49	31	1	54.8	4.62	61	453
1949032	8.8	1.88	0.12	71.85	0.6	0.63	1.43	6.07	16	0.05	99	8.63	52	814
1949035	1.2	0.75	0.09	16.99	0.63	0.93	0.45	2.48	91	0.05	30.8	2.71	169	145
1949047	2.2	2.1	0.13	24.03	0.7	0.71	1.57	10.64	37	4	111.6	9.33	197	433
1949048	3.8	1.43	0.12	32.07	0.65	0.8	0.96	4.71	41	4	73.8	5.87	28	393
1949808	1.5	1.29	0.05	26.09	0.72	0.71	0.62	4.92	44	2	40	3.94	127	423
1960843	1.6	0.98	0.05	11.46	0.52	2.26	0.59	4.72	33	0.05	38.3	3.91	14	330
1960889	1.9	1.71	0.05	24.49	0.73	0.44	0.86	4.94	52	0.05	63.8	5.58	48	46
1949010	0.1	0.37	0.05	2	0.08	1.83	0.14	2.35	16	0.05	13	0.84	33	1
1949020	0.7	0.4	0.05	4.05	1.58	0.39	0.24	4.7	361	13	12.4	1.58	143	9
0000002	1.5	0.88	0.05	15.35	0.98	0.82	0.48	4.16	195	1	31.4	3.09	215	16
1937182	0.4	0.29	0.05	1.06	0.38	0.36	0.21	0.47	186	0.05	11.3	1.17	51	2
1937185	0.5	0.51	0.1	1.85	1.28	0.01	0.42	0.83	318	3	21.8	2.28	44	9
1937609	1	1.37	0.05	6.05	1.04	0.3	0.7	0.99	153	0.05	48.3	4.27	81	24
1949013	1.9	2.24	0.07	15.47	2.36	0.67	1.24	4.07	29	22	85.3	7.62	85	25
1949040	0.5	0.63	0.1	3.19	1.18	0.8	0.29	0.91	348	4	20.6	1.86	75	10
1949044	0.4	0.54	0.07	4.3	0.88	1.87	0.33	2.29	202	4	21.4	2.08	1906	7
1949797	0.3	0.57	1.12	1.61	1	0.3	0.33	1.1	291	3	23.2	2.03	45	67
1949802	0.6	0.76	0.05	2.28	1.14	0.34	0.37	1.65	251	11	26.3	2.32	74	10
1949819	0.5	0.71	0.1	2.56	1.29	0.29	0.44	1.67	314	3	27.1	2.78	73	88
1960891	1.8	1.7	0.05	15.02	3.58	0.59	0.78	5.16	130	70	64.2	5.01	83	21
1933075	1.2	0.35	0.05	18.11	0.39	0.27	0.31	7.09	40	3	14.7	1.92	11	62
1933085	22.1	0.31	0.05	14.03	0.37	0.45	0.34	6.39	74	0.05	16.5	1.73	30	11
1946895	0.5	0.49	0.05	2.39	0.71	0.92	0.26	3.12	253	28	16.2	1.69	89	6
1949008	0.2	0.4	0.05	1.71	0.08	0.28	0.17	1.67	18	2	14	0.96	9	1
1949050	0.7	0.28	0.07	9.52	0.23	0.36	0.2	2.18	21	2	11.5	1.3	18	20
1949055	1.2	1.46	0.64	19.4	0.48	0.36	0.62	2.96	197	0.05	58.6	4.09	29	13
1949812	1.7	0.53	0.1	26.06	0.69	0.92	0.4	6.7	84	2	18.8	2.89	156	36
1960849	1.4	1.01	0.05	16.93	0.56	0.8	0.63	3.11	95	4	44.6	3.97	15	12
1933073	1.5	0.78	0.1	17.83	0.56	0.5	0.42	2.58	92	6	28.7	2.35	39	16
1933078	1.2	0.98	0.11	9.88	0.68	0.27	0.35	4.1	80	1	28.8	2.13	25	14

Kym Michael Custance Identifying IOCG pathfinder elements

Samp R#	DH_No	East	North	Depth from	Depth to	Strat_Name	Lith_type	Samp_type	Samp_from	Au ppb	Ag ppm	Al2O3%	As ppm	Ba ppm	Be ppm	Bi ppm	CaO%
1933080	23045	751229	6215071	61.4	62.4	Wandearah Formation	Sediment	Drill core	B1	0.5	0.025	15.96	1.7	599.3	4.5	0.13	0.36
1937180	22963	774067	6219496	291.4	292.4	Wandearah Formation	Sediment	Drill core	B1	1	0.025	14.21	15.3	89	0.6	2.25	5.15
1937383	30263	778393	6272671	313.1	314.1	Wandearah Formation	Sediment	Drill core	B1	0.5	0.025	10.41	7	494	2.4	0.03	5.62
1937613	23394	775422	6249469	7.89	8.9	New Cornwall Member	Sediment	Drill core	B1	0.5	0.025	10.82	9.7	113.3	1.8	0.13	5.47
1946901	139467	770829	6262371	71.2	72.2	Wandearah Formation	Sediment	Drill core	B1	0.5	0.025	13.79	5	25.6	2.1	0.1	0.7
1946902	139472	775504	6265446	38.5	39.5	Wandearah Formation	Sediment	Drill core	B1	0.5	0.025	11.03	15.2	741.1	2.3	0.005	8.78
1946906	139459	764604	6257546	93	94	Unnamed GIS unit	Sediment	Core	B1	2	4.26	13.74	19.7	641.3	6.7	0.32	0.36
1947176	23738	771077	6254479	44	46	Wandearah Formation	Sediment	Cuttings	B1	6	2.64	21.88	48.7	185.1	8.3	0.45	0.06
1949006	30062	777132	6256109	49.4	50.4	Wandearah Formation	Sediment	Drill core	B1	2	0.025	18.28	11	1067	3.2	0.11	0.02
1949017	30054	777508	6247943	108	108.2	Wandearah Formation	Sediment	Drill core	B1	6	1.28	2.73	341.6	162.9	0.25	10.03	1.73
1949023	30056	777364	6250259	108.8	109.8	Wandearah Formation	Sediment	Drill core	B1	0.5	0.55	14.26	30.7	694.1	2.7	12.55	2.87
1949026	30058	779440	6245971	202.2	203.2	New Cornwall Member	Sediment	Drill core	B1	1	0.72	11.63	51.2	653.2	2.1	6.45	7.26
1949034	23638	750354	6244057	142.04	143	Unnamed GIS unit	Sediment	Drill core	B1	2	0.025	11.61	25.1	705	4.4	0.11	0.18
1949789	23049	769300	6233887	82.3	82.9	Unnamed GIS unit	Sediment	Drill core	B1	1	0.06	15.74	17.8	613.6	2.7	0.25	0.29
1949791	23048	770327	6233976	41.3	42.3	Unnamed GIS unit	Sediment	Drill core	B1	2	0.11	15.94	24.6	613.6	4	0.1	0.3
1949799	23027	745198	6225492	17.37	18.4	Doora Member	Sediment	Drill core	B1	0.5	0.06	17.74	28.3	951.6	3.2	0.01	0.53
1949804	23672	752805	6242398	139.6	140.6	Wandearah Formation	Sediment	Drill core	B1	0.5	0.025	12.07	5	389.6	0.9	0.02	0.36
1949806	23030	745177	6225864	22.86	23.8	Doora Member	Sediment	Drill core	B1	0.5	0.025	16.62	5	961.3	4	0.005	0.27
1949809	23046	768768	6233533	48.77	49.8	Wandearah Formation	Sediment	Drill core	B1	30	0.025	14.19	6	220.3	4.9	0.37	0.4
1949815	23696	768986	6235239	182.12	183.1	Unnamed GIS unit	Sediment	Drill core	B1	120	0.025	16.09	8	333.8	5.1	0.04	0.35
1949817	23726	777294	6247803	53.65	54.7	Wandearah Formation	Sediment	Drill core	B1	3	0.3	14.13	10	1106.9	2.5	1.19	1.11
1949822	30050	781823	6257617	292	293	Wandearah Formation	Sediment	Drill core	B1	1	0.025	15.64	27	1362.7	3.7	0.33	2.97
1949825	23700	741294	6240971	50.6	51.6	Doora Member	Sediment	Drill core	B1	0.5	0.025	17.49	9	847.6	3.8	0.08	0.33
1949827	23691	760919	6243044	47.7	48.7	Unnamed GIS unit	Sediment	Drill core	B1	3	0.2	9.13	8	1790.1	4.7	0.11	12.65
1960877	23047	769461	6233877	101.47	102.5	Unnamed GIS unit	Sediment	Drill core	B1	35	0.6	16.26	10	176.8	9.1	0.61	0.62
1960886	23037	769374	6233729	53.03	54	Unnamed GIS unit	Sediment	Drill core	B1	6	0.025	16.11	6	602	3.8	0.19	0.28
1960894	23698	769811	6234551	271.1	272.1	Unnamed GIS unit	Sediment	Drill core	B1	4	0.025	14.72	4	452.7	6.3	0.08	0.69
000003		734268	6169901	154.1	155.1		Mafic intrusive	Drill core	B2	0.5	1.1	6.46	7	656.4	1.4	0.13	1.43
000006		736061	6169308	153.7	154.7		Mafic intrusive	Drill core	B2	0.5	0.025	8.26	0.5	345.5	2.2	0.67	11.89
1933071	139476	763729	6257846	81.4	81.95	Wandearah Formation	Granite	Drill core	B2	5	0.025	10.83	1.7	40	0.7	0.07	5.92
1933076	23265	747787	6247265	12.72	13.71	Oorlano Metasomatite	Metasomatite	Drill core	B2	0.5	0.025	20.85	0.5	593.3	3.8	0.05	6.11
1933079	139489	734539	6211211	43.1	44.1	Wandearah Formation	Sediment	Drill core	B2	0.5	0.025	18.04	0.5	242.2	4.9	0.04	0.96
1933081	23045	751229	6215071	71.4	72.4	Wandearah Formation	Sediment	Drill core	B2	0.5	0.025	17.81	4.3	325.2	2.9	0.04	0.59
1933083	23015	742809	6208841	24.3	25.3	Weetulta Formation	Felsic volcanic	Drill core	B2	0.5	0.2	10.31	3	60.6	1.7	0.005	0.28

Samp R#	Cd ppm	Ce ppm	Co ppm	Cr ppm	Cs ppm	Cu ppm	Dy ppm	Er ppm	Eu ppm	F ppm	Fe2O3%	Ga ppm	Gd ppm	Ge ppm	Hf ppm	Ho ppm	In ppm	K2O%	La ppm	Li ppm	Lu ppm
1933080	0.01	79.9	28.3	74	4.29	11	4.29	2.24	1.19	1618	9.04	22.7	5.3	0.56	3.6	0.81	0.09	6.61	49.6	79.1	0.33
1937180	0.01	64.2	27.6	36	1.4	12	3.03	1.76	1.13	257	3.24	17.9	3.7	0.7	2.8	0.61	0.03	4.14	36.6	7.8	0.29
1937383	0.01	65.7	3.5	35	4.19	7	5.71	3.14	1.18	1424	8.82	15.9	5.42	1.43	3	1.2	0.103	4.06	33.1	15.6	0.41
1937613	0.03	42.8	14.6	36	0.99	8	4.84	2.71	1.67	498	10.02	15.3	4.95	1.31	2.6	1.03	0.035	1.24	19.1	8	0.44
1946901	0.01	285.7	5.5	55	0.45	95	7.13	3	2.98	812	3.67	14.7	12.02	1.4	4.1	1.23	0.007	0.22	147.6	11.6	0.43
1946902	0.01	27.1	3.3	34	1.62	11	2.96	1.77	0.62	1394	3.13	16.2	2.95	1.26	2.6	0.62	0.028	2.56	14.8	24.7	0.29
1946906	0.09	90	41.3	121	8.99	328	4.85	2.57	1.01	1700	6.6	20.9	5.49	1.74	5.7	0.95	0.051	4.83	47.2	33.6	0.41
1947176	0.28	350.4	51.3	76	11.44	163	22.22	11.76	5.25	1648	6.96	33.4	26.76	2.85	9.6	4.49	0.081	2.73	209.4	54.6	1.74
1949006	0.01	41	6	72	7.57	25	4.71	3.09	0.73	1707	5.62	27.6	3.65	1.31	3.6	1.04	0.114	6.41	18.7	30.3	0.51
1949017	0.24	6.4	62.3	10	0.79	980	1.31	0.72	0.23	138	4.66	2.5	1.21	0.42	0.8	0.26	0.071	1.73	3	6.7	0.15
1949023	0.04	115.7	34.1	69	3.77	545	9.38	5.51	1.62	1817	5.43	22.6	9.17	0.36	4.6	1.99	0.159	5.45	56.4	31.4	0.7
1949026	0.03	75.2	19.8	60	4.19	225	3.35	1.99	0.74	1193	3.91	16.9	3.74	0.6	3.2	0.69	0.092	3.52	33.3	22.5	0.31
1949034	0.01	153.9	8.6	60	4.58	10	3.58	1.92	1.48	1199	6.37	18.4	5.61	0.6	2.9	0.7	0.061	5.67	82.5	44.2	0.3
1949789	0.01	88.5	17.5	71	7.28	10	3.6	2.31	0.99	1701	5.37	21.5	4.52	0.75	4.1	0.78	0.023	6.73	46.8	34.7	0.39
1949791	0.01	29.2	11.4	131	12.57	55	4.49	2.76	0.82	2106	5.94	24.2	3.96	0.97	4.6	0.96	0.099	6.39	12.8	54.7	0.42
1949799	0.01	82.4	11.3	91	4.3	122	6.73	3.83	1.77	2574	7.13	23.7	7.49	0.42	3.8	1.36	0.047	4.89	44.2	38	0.63
1949804	0.01	80.1	4.1	51	1.92	74	2.85	1.78	1.42	686	3.35	11.6	3.7	0.6	7.5	0.59	0.01	9.47	41.2	11.9	0.32
1949806	0.01	132.6	7.7	100	4.5	33	8.57	4.94	2.18	2370	7.45	23.3	9.05	0.6	4.5	1.79	0.09	5.94	66.8	27.1	0.84
1949809	0.01	83.7	42.4	69	8.52	61	4.77	3.08	1.05	2210	4.9	19.5	5.02	0.8	5.6	1.05	0.03	4.23	44.2	39.6	0.55
1949815	0.1	39.1	23.4	86	10.73	73	3.35	2.11	0.61	2172	5.1	24.6	2.96	0.6	3.7	0.74	0.07	5.33	21.6	44.2	0.36
1949817	0.01	64.4	12.3	69	2.85	49	4.46	2.55	1.05	944	6.17	20.9	4.96	0.7	4.4	0.9	0.08	5.43	27	38.8	0.4
1949822	0.01	86.2	20.7	72	5.83	28	4.93	2.85	1.24	1784	5.49	23	5.66	0.9	4.8	1.02	0.09	4.96	43.6	47.5	0.41
1949825	0.01	286.8	27	110	4.55	116	6.58	2.77	2.56	1760	13.95	17.3	13.54	1.2	3.9	1.13	0.02	10.03	140.1	50.2	0.47
1949827	0.1	51.8	19.8	47	5.43	42	3.89	2.13	0.98	2354	3.63	16.6	4.32	0.7	3	0.77	0.04	4.68	26	33.4	0.28
1960877	0.1	105.9	50.1	44	15.75	164	48.87	33.9	4.48	3052	5.92	16	30.15	1.1	11.8	11.2	0.03	5.15	59.4	94	5.66
1960886	0.1	69.5	18.3	72	9.79	32	5.6	3.07	0.94	1907	5.74	26.4	5.05	1.3	4.5	1.13	0.07	6.3	35.9	72.1	0.42
1960894	0.01	47.4	16.4	64	13.61	50	5.52	3.2	0.86	2995	6.74	22.2	4.37	1.1	4.2	1.14	0.04	5.76	24.4	69.4	0.42
000003	0.01	44.2	11	23	2.56	31	2.64	1.45	0.74	278	2.66	8.3	2.83	0.2	4.2	0.51	0.02	2.04	22.5	12.5	0.23
0000006	0.01	78.9	8.8	64	2.16	30	7.18	4.08	1.44	751	5.82	12.9	7.3	0.7	8	1.46	0.07	3.53	38.5	20.3	0.58
1933071	0.01	8.6	5	27	0.45	147	1.86	1.35	0.26	300	1.62	11	1.37	0.1	8	0.47	0.009	0.5	4.6	9.4	0.37
1933076	0.46	97.2	13.6	96	6.1	51	6.51	3.71	1.76	2070	6.37	32.6	6.51	0.7	5.4	1.33	0.06	3.13	49.6	45.7	0.56
1933079	0.01	538.3	30	72	1.47	13	15.44	8.47	2.92	1513	7.31	27.8	22.34	1.85	6.7	2.94	0.02	5.07	280.9	24	1.34
1933081	0.01	106.7	15.5	78	2.69	8	5.94	3.72	1.44	883	7.71	23	5.91	0.21	3.1	1.25	0.036	3.93	64.4	29.5	0.56
1933083	0.06	6.3	36.4	10	7.26	32	2.67	2.01	0.22	6541	29.56	33.5	1.5	2.2	9.6	0.62	0.018	2.14	2.4	90.8	0.43

Samp R#	MgO%	MnO%	Mo ppm	Na2O%	Nb ppm	Nd ppm	Ni ppm	P2O5%	Pb ppm	Pd ppb	Pr ppm	Pt ppb	Rb ppm	Re ppm	S ppm	Sb ppm	Sc ppm	Se ppm	SiO2%	Sm ppm	Sn ppm	Sr ppm
1933080	4.91	0.03	0.3	1.26	10.9	36	45	0.25	2.5	0.05	9.64	0.05	337.1	0.002	25	0.64	16	0.025	56.68	6.47	8	23.8
1937180	2.65	0.26	0.9	5.22	11.5	28.8	6	0.24	2.5	0.05	7.92	1	112.9	0.002	1119	1.03	13	0.025	57.08	4.96	10	17.9
1937383	3.4	0.73	0.6	0.08	11.1	29.7	9	0.2	2.5	0.05	7.8	0.05	216.2	0.011	99	0.93	12	4	56.11	5.77	4	19.8
1937613	3.61	0.6	1.2	4.75	7.2	19.8	6	0.1	2.5	0.05	4.77	0.05	47.6	0.003	64	1.18	10	2.8	52.74	4.52	9	47
1946901	1.55	0.15	6.7	6.42	13.6	111.9	15	0.22	2.5	0.05	31.68	0.05	14.4	0.028	989	0.85	10	5.6	70.7	18.26	5	38.1
1946902	6.1	0.82	0.8	2.4	10	14.4	10	0.09	7	0.05	3.61	0.05	130.3	0.005	25	0.67	10	3.6	49.23	3.42	4	32.7
1946906	1.74	0.04	6	0.29	14.9	37.7	82	0.28	183	3	10.08	2	408.9	0.003	139	0.78	12	4.1	65.11	6.96	6	40.9
1947176	0.93	0.15	2.7	0.31	62.8	183	46	0.16	297	0.05	49.1	0.05	228.8	0.02	692	3.17	28	3	54.75	35.69	12	33.6
1949006	1.39	0.01	0.5	0.09	10.61	17	18	0.005	6	0.05	4.42	0.05	344.2	0.002	56	1.15	21	3.7	61.87	3.64	6	11.4
1949017	1.02	0.12	24.6	0.07	2.26	3.3	46	0.005	58	7	0.8	3	57.8	0.025	12929	2.48	10	2	82.28	0.95	1	11.4
1949023	2.71	0.15	4.4	0.12	15.16	50.1	15	0.16	22	0.05	13.17	1	200.1	0.005	10449	1.63	19	3.6	60.16	9.97	6	11.3
1949026	5.97	0.32	5	0.99	6.81	31.2	12	0.17	17	2	8.38	1	152.4	0.002	4756	0.83	12	3.4	51.34	5.44	3	21.8
1949034	2.33	0.11	1.3	0.07	11.5	57.4	20	0.14	7	0.05	16.32	0.05	353.4	0.002	25	1.29	13	2.9	69.49	9.23	3	29.6
1949789	3.25	0.01	0.2	0.14	9.32	38.6	32	0.25	10	0.05	10.83	1	295.4	0.002	124	0.34	16	2	63.69	6.77	4	45.3
1949791	2.4	0.07	0.7	0.24	14.9	15.7	23	0.23	2.5	0.05	3.92	0.05	288.8	0.003	128	0.41	16	4.3	62.83	3.69	21	45.7
1949799	2.26	0.01	2.5	5.16	10.86	34.8	16	0.12	2.5	0.05	9.35	0.05	235.1	0.006	93	0.05	19	1.5	59.93	7.43	13	93.6
1949804	1.07	0.01	0.2	0.14	9.42	30.8	24	0.23	2.5	0.05	8.4	0.05	266.8	0.002	160	0.18	1	0.025	72.52	5.25	4	24
1949806	2.33	0.01	1.3	1.76	13.6	55.9	49	0.08	12	0.05	15.09	0.05	327.5	0.002	207	0.025	19	0.025	63.38	10	9	232.7
1949809	3.43	0.01	1.1	0.62	7.63	32.5	45	0.15	7	0.05	9	0.05	219	0.002	154	0.28	14	0.025	68.89	5.82	3	30.2
1949815	3.99	0.03	0.2	1.08	10.05	16	35	0.17	8	2	4.45	1	287.7	0.002	199	0.025	18	0.025	64.76	3.09	4	17.4
1949817	2.85	0.08	2.1	0.07	13.72	28.2	35	0.16	9	1	7.46	1	165.4	0.002	762	0.44	14	0.025	65.91	5.58	4	22.8
1949822	3.63	0.26	0.8	0.85	9.63	36.4	55	0.19	7	0.05	9.73	3	272.1	0.002	8286	1.32	16	0.025	59.63	6.89	4	15.2
1949825	3.64	0.03	0.4	0.81	18.84	121	49	0.22	10	0.05	32.81	4	260.3	0.002	114	1.02	16	0.025	50.76	19.61	2	40.5
1949827	9.71	1.33	3.1	0.22	3.4	21.2	53	0.15	6	0.05	5.77	2	214.4	0.002	1257	1.45	1	0.025	37.36	4.41	4	27.8
1960877	4.66	0.01	0.3	0.07	17.02	52.9	56	0.37	13	7	14.11	4	293.9	0.002	248	0.5	22	0.025	62.06	14.25	2	12.2
1960886	3.23	0.03	0.5	0.12	13.3	27.9	56	0.17	18	1	7.75	1	296.3	0.002	156	0.58	17	0.025	63.85	5.44	4	24.9
1960894	4.39	0.03	0.4	0.1	13.47	19.3	39	0.16	20	5	5.2	2	314.3	0.002	230	0.38	14	0.025	62.52	3.95	4	28.5
000003	0.59	0.05	0.4	1.09	4.51	18	18	0.06	22	0.05	4.9	2	78.6	0.002	335	0.52	1	0.025	84.15	3.51	4	73.7
000006	8.84	0.17	0.6	1.14	11.33	32.6	44	0.27	5	0.05	8.75	0.05	100.5	0.002	306	0.33	1	0.025	42.34	7.28	3	50.2
1933071	0.91	0.06	0.7	5.79	11.4	4	6	0.33	26	0.05	1.02	1	20.6	0.002	218	0.13	10	1.8	68.85	1.08	4	30.6
1933076	2.13	0.04	2	6.01	19.3	42.8	28	0.35	2.5	0.05	11.17	1	239.5	0.002	703	0.32	16	3	51.22	7.83	15	184
1933079	2.65	0.01	1	4.23	74.7	222.7	67	0.21	7	0.05	63.19	0.05	167.2	0.002	323	0.31	15	2	57.28	32.72	8	47.4
1933081	2.04	0.01	0.5	6.08	11.8	44.2	34	0.21	2.5	0.05	12.21	0.05	189.3	0.002	25	0.4	16	3.6	59.77	7.23	7	42.7
1933083	4.06	0.01	2.8	3.89	31.7	3.8	96	0.05	2.5	0.05	0.86	0.05	345.3	0.002	334	0.13	10	2	46.19	1.31	10	25.5

amp R#	Ta ppm	Tb ppm	Te ppm	Th ppm	TiO2%	Tl ppm	Tm ppm	U ppm	V ppm	W ppm	Y ppm	Yb ppm	Zn ppm	Zr ppm
933080	1.4	0.71	0.05	16.98	0.69	0.8	0.37	6.06	103	1	24.1	2.13	164	122
937180	1.5	0.51	0.3	11.57	0.54	0.7	0.33	4.32	53	6	18.3	1.71	26	97
937383	0.8	0.85	0.05	12.78	0.41	0.41	0.46	2.92	61	4	32.6	2.98	11	104
937613	0.8	0.74	0.05	10.48	0.36	0.26	0.39	3.05	111	9	31.4	2.51	20	98
946901	1.3	1.38	0.05	22.94	0.54	0.06	0.4	6.51	58	16	37.7	2.68	21	154
946902	0.9	0.46	0.05	12.8	0.39	0.17	0.29	0.96	65	8	18.8	1.82	16	96
946906	1.4	0.76	0.51	19.51	0.65	2.23	0.38	22.48	147	3	26.7	2.61	553	203
947176	5.5	3.66	0.05	81.43	1.63	1.93	1.84	11.31	188	5	118.2	11.86	246	326
949006	1.5	0.67	0.13	15.84	0.65	1	0.49	1.77	100	5	31.8	3.15	87	125
949017	0.1	0.21	0.15	4.92	0.11	3.16	0.1	0.46	39	0.05	7.4	0.74	117	33
949023	1.5	1.45	0.19	16.59	0.59	2.02	0.82	3.45	155	10	60.6	5.05	45	170
949026	1.3	0.52	0.07	13.63	0.5	0.92	0.31	2.85	65	6	19.3	1.96	21	110
949034	0.9	0.66	0.16	13.38	0.45	1.36	0.29	9.65	62	6	21.2	1.78	107	111
949789	1.5	0.59	0.15	18.17	0.6	0.58	0.37	2.69	124	5	23.2	2.25	48	143
949791	1.5	0.66	0.22	18.66	0.62	0.82	0.42	4.25	131	8	27.5	2.66	349	173
949799	1.2	1.08	0.06	18.7	0.65	0.59	0.57	3.46	80	4	45.2	3.69	26	134
949804	1.5	0.46	0.05	16.38	0.5	0.68	0.31	3.42	31	7	17.4	2.07	13	289
949806	1.5	1.34	0.05	20.88	0.68	0.75	0.76	6.62	107	11	53.5	4.95	212	151
949809	1.4	0.72	0.2	17.42	0.55	0.56	0.46	2.21	79	5	31.4	3.26	72	188
949815	1.5	0.49	0.3	13.77	0.58	0.58	0.32	25.74	149	7	22.1	2.14	143	128
949817	1.6	0.71	0.05	18.04	0.6	2.27	0.41	3.69	147	6	25.1	2.47	36	153
949822	1.5	0.81	0.05	20.71	0.59	2.57	0.41	1.91	87	4	28.8	2.77	22	141
949825	1.1	1.35	0.05	24.41	0.61	0.54	0.36	2.74	103	2	34	2.72	25	137
949827	0.8	0.63	0.6	10.25	0.36	0.69	0.25	11.59	64	4	23.5	1.82	178	99
960877	2.8	6.4	2.4	26.92	0.74	0.39	5.11	9.26	121	13	373.2	33.36	92	416
960886	1.4	0.84	0.05	17.83	0.65	0.61	0.42	3.33	153	15	34.5	2.77	149	166
960894	1.3	0.79	0.05	15.34	0.58	0.74	0.42	3.78	132	6	34.9	2.74	35	154
000003	0.8	0.43	0.05	7.18	0.25	0.45	0.18	1.33	24	5	14.9	1.41	19	154
000006	1.7	1.12	0.05	17.22	0.61	0.37	0.6	3.72	58	3	46.3	3.69	33	290
933071	1.6	0.32	0.11	13.66	0.4	0.01	0.32	6.71	33	4	11.9	1.75	42	296
933076	2	0.96	0.05	24.22	0.8	0.4	0.59	8.9	103	2	38.1	3.52	30	193
933079	3.4	2.65	0.05	46.8	0.56	0.2	1.25	32.17	39	1	82.6	8.13	31	205
933081	1.4	0.87	0.05	15.95	0.64	0.36	0.55	4.12	96	2	43.8	3.29	41	112
933083	2.8	0.33	0.1	18.05	0.46	0.3	0.41	7.73	249	0.05	16.7	2.4	78	353

Kym Michael Custance Identifying IOCG pathfinder elements

Samp R#	DH_No	East	North	Depth from	Depth to	Strat_Name	Lith_type	Samp_type	Samp_from	Au ppb	Ag ppm	Al2O3%	As ppm	Ba ppm	Be ppm	Bi ppm	CaO%
1933086	143968	761879	6160620	219.3	220.3	Oorlano Metasomatite	Metasomatite	Drill core	B2	0.5	0.025	9.8	4	183.9	4.1	0.09	0.55
1933089	23197	764842	6266413	131	132	Tickera Granite	Granite	Drill core	B2	0.5	0.025	12.58	0.5	344.1	4.1	0.05	0.22
1933092	22717	750382	6219225	34	35	Arthurton Granite	Granite	Drill core	B2	0.5	0.025	15.11	0.5	1155.4	2.8	0.04	1.26
1937181	22963	774067	6219496	301.4	302.4	Wandearah Formation	Sediment	Drill core	B2	3	0.025	11.94	7.7	50.6	1	0.47	9.41
1937183	29692	776523	6266268	34.6	35.6	Bute Metadolerite	Mafic intrusive	Drill core	B2	2	0.025	18.64	3	267.1	0.9	0.17	6.87
1937186	29752	783721	6248572	450	451	Bute Metadolerite	Mafic intrusive	Drill core	B2	0.5	0.025	13.3	3	109.3	0.25	0.03	6.96
1937188	143973	753159	6166371	31	32	Unnamed	Gneiss	Drill core	B2	0.5	0.025	11.26	4	449.4	2.2	0.06	0.33
1937190	143974	745809	6165690	31	32	Unnamed	Gneiss	Drill core	B2	0.5	0.025	14.36	0.5	751.7	3	0.03	3.49
1937384	30263	778393	6272671	323	324	Wandearah Formation	Sediment	Drill core	B2	0.5	0.025	13.64	4.4	823.1	2.7	0.09	0.25
1937387	178246	776838	6250511	32	34	Wandearah Formation	Carbonate	Cuttings	B2	6	0.8	17.15	97	143.9	2	0.49	2.21
1937390	23748	775431	6244530	26	28	Weetulta Formation	Felsic volcanic	Cuttings	B2	0.5	0.3	15.35	6.8	1014.1	1.9	0.07	0.03
1937393	174055	777406	6244863	75	80	Bute Metadolerite	Mafic intrusive	Cuttings	B2	2	0.025	16.71	15	368.9	1.1	0.34	1.04
1937396	23413	746513	6236028	17.2	18.2	Moonta Porphyry Member	Felsic volcanic	Drill core	B2	0.5	0.025	11.16	5.6	128.5	6	0.03	0.53
1937399	143253	751915	6208687	14	18	Wallaroo Group	Felsic volcanic	Cuttings	B2	0.5	0.025	10.43	0.5	1388.1	1.7	0.07	0.63
1937402	22676	750690	6160130	110	114	Arthurton Granite	Granite	Cuttings	B2	0.5	0.025	12.43	0.5	845.8	2.3	0.02	1.2
1937405	142663	751829	6208254	22	24	Arthurton Granite	Granite	Cuttings	B2	0.5	0.025	11.68	0.5	293.5	2.9	0.03	0.24
1937610	143970	747769	6164961	40	41	Unnamed	Mafic intrusive	Drill core	B2	0.5	0.025	16.15	2.5	553.5	2.2	0.04	6.56
1937612	143971	751329	6162690	31	32	Unnamed	Granite	Drill core	B2	0.5	0.11	12.54	0.5	509.9	3	0.005	0.92
1946896	139474	771179	6264821	64.6	65.6	Oorlano Metasomatite	Metasomatite	Drill core	B2	0.5	0.025	14.2	4.7	242.4	1.5	0.06	9.54
1946903	139472	775504	6265446	48.5	49.5	Wandearah Formation	Sediment	Drill core	B2	0.5	0.025	8.4	7.1	743	2	0.03	11.69
1946905	139458	763604	6257871	46.5	47.5	Amphibolite and Wandearah Formation	Amphibolite	Core	B2	2	0.24	15.18	9.7	245.7	3.2	0.08	2.8
1946907	139459	764604	6257546	103	104	Unnamed GIS unit	Sediment	Core	B2	2	0.58	16.6	16.3	586.6	4.6	0.39	0.42
1946912	143205	743608	6203678	22	24	Arthurton Granite	Granite	Cuttings	B2	0.5	0.025	16.75	3.1	2216.4	2.1	0.02	0.56
1947177	23738	771077	6254479	52	53.5	Wandearah Formation	Sediment	Cuttings	B2	1	1.11	15.88	41.4	418.2	4.8	0.39	0.16
1947180	143281	745443	6213456	28	30	Wallaroo Group	Sediment	Cuttings	B2	0.5	0.07	21.63	12.8	561.4	2.9	0.02	0.44
1949009	30060	777888	6243276	147.3	148.3	Oorlano Metasomatite	Metasomatite	Drill core	B2	1	0.07	12.84	9.2	549.1	2.2	0.09	3.57
1949015	139483	758529	6267471	189.6	190.6	Tickera Granite	Granite	Drill core	B2	0.5	0.07	13.44	19.6	826.9	3.8	0.005	0.63
1949018	30054	777508	6247943	117.3	118.3	Wandearah Formation	Sediment	Drill core	B2	0.5	0.16	11.59	25.4	324.7	3.1	0.38	1.36
1949021	30055	778282	6248945	111.85	112.85	Willamulka Basalt Member	Mafic extrusive	Drill core	B2	0.5	0.38	13.64	27.4	162.5	1.2	2.47	8.91
1949024	30056	777364	6250259	118.8	119.8	Wandearah Formation	Sediment	Drill core	B2	4	0.21	12.53	9.4	717.1	3	0.77	0.84
1949027	30058	779440	6245971	212.2	213.2	New Cornwall Member	Sediment	Drill core	B2	1	0.14	12.96	24.3	800.9	2.3	1.5	5.83
1949029	22991	737229	6222021	37.2	38.2	Hiltaba Suite	Granite	Drill core	B2	0.5	0.09	13.2	22	871.4	3	0.02	0.49
1949038	23039	740506	6225440	37	37.5	Moonta Porphyry Member	Felsic volcanic	Drill core	B2	0.5	0.26	11.09	19.9	97.3	6.4	0.13	0.13
1949043	23033	762660	6232342	40.9	41.9	Unnamed GIS unit	Carbonate	Drill core	B2	1	0.025	7.48	3	465.6	1.2	0.005	29.22
1949045	23663	763406	6250765	84	85	Unnamed GIS unit	Mafic intrusive	Drill core	B2	0.5	0.14	15.86	24.2	381.3	3	0.13	0.34

Samp R#	Cd ppm	Ce ppm	Co ppm	Cr ppm	Cs ppm	Cu ppm	Dy ppm	Er ppm	Eu ppm	F ppm	Fe2O3%	Ga ppm	Gd ppm	Ge ppm	Hf ppm	Ho ppm	In ppm	K2O%	La ppm	Li ppm	Lu ppm
1933086	0.01	13.6	102.9	36	1.84	42	1.84	1.38	0.22	687	29.78	19.7	1.23	1.7	3.5	0.43	0.05	1.81	6.8	57.6	0.32
1933089	0.01	205.8	4	10	1.92	9	12.85	8.47	1.32	637	4.45	19.2	11.07	0.78	9.1	2.7	0.009	4.73	89.1	12	1.65
1933092	0.07	196	7.1	10	2.58	14	13.84	7.27	3	1562	4.96	21	14.79	0.87	11.8	2.62	0.045	5.51	95	36.9	1.01
1937181	0.01	54.7	6	37	0.5	57	2.92	1.58	0.87	290	3.79	18	3.49	1.45	2.4	0.54	0.036	0.8	28.2	22.4	0.27
1937183	0.01	9.7	41	713	13.15	222	1.29	0.7	0.37	413	6.86	14.9	1.13	1.13	0.8	0.26	0.05	2.66	4.8	70.4	0.13
1937186	0.01	16.9	33.9	248	0.82	11	4.1	2.33	1.41	612	14.13	18.3	3.54	1.27	2.1	0.82	0.06	0.74	8.3	75.1	0.37
1937188	0.07	67.9	26.5	24	2.25	28	8.33	5.03	1.48	532	9.82	24.9	7.06	0.61	9.4	1.73	0.101	3.86	35.2	67.7	0.88
1937190	0.08	95.6	15.4	53	2.25	16	8.62	4.59	2.01	973	6.4	20.7	8.59	0.64	7.4	1.72	0.063	2.72	48.3	31.9	0.67
1937384	0.01	92.5	3.9	54	5.95	6	4.48	2.47	1.09	1678	9.57	19.8	4.83	1.61	3.5	0.96	0.058	4.83	42.5	17.8	0.39
1937387	0.1	82.2	39.3	51	4.77	484	5.64	3.24	1.68	1124	5.16	20.8	5.46	1.54	4.3	1.13	0.04	3.15	43.3	30.9	0.48
1937390	0.19	260.9	0.6	10	1.1	11	18.39	11.09	3.93	544	1.29	29.6	20.71	1.25	18.1	3.68	0.03	3.47	131.8	9.9	1.93
1937393	0.01	20.4	28.4	271	3.48	28	1.81	1.27	0.53	1100	8.38	22	1.67	0.98	3.1	0.38	0.04	5.08	9.1	47.3	0.31
1937396	0.15	77.4	6	10	3.15	57	5.3	3.98	0.84	708	6.58	25.5	4.87	1.24	14	1.28	0.006	1.46	44.2	0.3	0.93
1937399	0.09	71.9	1.1	10	1.9	10	4.93	3.16	0.94	206	4.33	18.3	4.02	0.6	11.9	1.02	0.036	3.28	39.5	0.7	0.64
1937402	0.09	127.2	6.2	10	1.29	32	11.57	6.82	2.64	527	5.71	18.1	11	0.75	14	2.36	0.099	4.72	64	13.4	1.04
1937405	0.01	14.2	1.7	10	2.19	5	1.33	0.89	0.38	158	2.84	13.8	1.19	0.85	1.8	0.29	0.018	3.45	9	1.5	0.19
1937610	0.06	101	35	41	4.75	70	8.23	4.73	1.82	1606	9.54	20.5	8.49	1.58	4.4	1.74	0.079	1.95	49.1	16.8	0.69
1937612	0.01	69.8	16.6	10	1.24	25	8.28	5.01	1.96	529	7.94	20.2	8.16	0.99	13	1.8	0.022	3.22	34.4	22.5	0.87
1946896	0.01	58.8	28.4	309	4.09	29	2.78	1.64	0.76	1937	8.38	14.8	2.7	1.56	1.1	0.6	0.065	2.4	36.9	37.9	0.28
1946903	0.01	70.9	3.9	10	1.86	10	4.12	2.3	1.13	940	6.61	12.5	5.03	1.29	1.9	0.83	0.06	2.15	41.1	19	0.31
1946905	0.14	23.9	44.6	154	4.95	216	4.12	2.5	0.99	687	12.1	22.1	3.88	1.47	3.1	0.93	0.089	2.48	12.2	34	0.47
1946907	0.86	85.5	147	71	8.75	290	4.41	2.36	0.92	2170	4.89	21.5	5.23	1.4	4.9	0.89	0.029	5.36	43.3	49.5	0.31
1946912	0.01	107.1	0.9	108	1.32	7	6.97	3.03	2.4	341	1.87	19.7	8.85	0.95	5	1.3	0.019	5.47	59.1	3.3	0.35
1947177	0.27	255.8	21.4	78	9.18	75	12.93	7.34	2.56	1614	5.74	26.1	13.71	2.22	7.4	2.72	0.051	6.59	133	53.4	1.12
1947180	0.01	373.3	4.5	173	8.2	148	13.18	4.98	5.19	1454	3	32.6	19.14	2.45	5.1	2.19	0.104	5.95	179.2	16.3	0.55
1949009	0.01	144	14.8	71	7.58	14	3.02	1.64	1.95	2057	11.32	22.7	4.61	0.55	2.9	0.58	0.043	3.79	72.9	44.3	0.3
1949015	0.01	205.9	7.8	10	1.31	17	9.69	5.41	2.2	1166	4.85	18.4	10.99	0.15	10.8	1.94	0.04	5.65	106.4	16.1	0.77
1949018	0.02	38.6	11.6	59	3.14	35	2.73	1.51	0.68	685	4.95	13	2.73	0.43	3.7	0.55	0.031	2.74	19.2	59.9	0.22
1949021	0.17	16.6	28.7	101	1.61	109	4.32	2.61	0.99	549	13.12	21.1	3.81	0.43	2.5	0.92	0.094	0.57	7.8	41.1	0.41
1949024	0.01	110.8	15.3	39	5.07	48	10.65	7.25	1.29	1432	6.39	18.3	10.28	0.33	8.4	2.37	0.069	5.13	53.3	53.9	1.2
1949027	0.01	103.1	7.4	61	3.83	21	4.04	2.13	1.03	1413	4.22	19.4	5.13	0.66	3.1	0.8	0.077	3.88	52.2	27.2	0.27
1949029	0.01	202.8	9.7	10	0.68	15	11.93	7.04	2.37	1226	5.52	20.3	12.59	0.2	10.3	2.44	0.0025	5.22	92.4	12.4	1.1
1949038	0.01	41.1	29.2	10	0.92	173	12.45	10.73	0.96	2285	10.03	24.9	6.36	0.81	14	3.21	0.073	3.57	20.3	52.9	2.13
1949043	0.03	36.8	3.7	38	3.33	180	2.31	1.46	0.37	1248	2.15	11.6	2.34	0.44	2	0.49	0.032	1.17	18.7	21.4	0.21
1949045	0.01	21.5	85	202	6.44	264	3.38	2.07	0.58	1505	7.86	19.9	2.82	1.08	2.2	0.71	0.042	5.43	11.4	93.7	0.31

Samp R#	MgO%	MnO%	Mo ppm	Na2O%	Nb ppm	Nd ppm	Ni ppm	P2O5%	Pb ppm	Pd ppb	Pr ppm	Pt ppb	Rb ppm	Re ppm	S ppm	Sb ppm	Sc ppm	Se ppm	SiO2%	Sm ppm	Sn ppm	Sr ppm
1933086	6.12	0.06	0.5	2.21	12.3	5.6	20	0.21	2.5	3	1.6	0.05	105	0.002	25	0.14	10	4	45.93	1.12	20	14.7
1933089	0.77	0.03	0.6	2.96	57.4	62.6	0.05	0.18	2.5	0.05	18.1	0.05	404.3	0.002	25	0.41	10	2.7	72.2	12.8	10	27.9
1933092	1.45	0.04	0.5	3.22	32.5	89.7	2	0.29	2.5	0.05	23.23	0.05	235	0.002	66	0.15	17	3.6	66.33	17.51	4	119.3
1937181	2.58	0.31	2.8	5.6	8.2	22.8	10	0.26	2.5	2	6.01	1	18	0.002	233	0.47	11	3.6	55.92	4.24	8	38.5
1937183	6.39	0.15	0.3	2.26	2.2	4.1	123	0.14	2.5	10	1	2	140.2	0.002	25	0.46	48	4	48.99	0.98	1	194.2
1937186	6.87	0.18	0.7	2.25	1.5	9.9	70	0.17	2.5	11	2.17	11	21	0.002	259	0.29	39	3	43.52	2.92	3	49.5
1937188	3.4	0.06	0.8	2.27	25.8	32.5	40	0.21	8	0.05	8.16	0.05	168.6	0.002	25	0.08	18	0.025	65.32	6.92	8	51.9
1937190	2.27	0.09	1.3	3.44	16.6	46.3	25	0.26	9	0.05	11.45	0.05	125.1	0.002	86	0.025	15	3	64.49	9.59	14	261.4
1937384	1.11	0.01	1.1	0.63	13.2	38.2	15	0.17	2.5	0.05	10.71	0.05	261.6	0.008	81	1.26	14	5.6	67.33	6.73	4	14.7
1937387	3.15	0.26	3	5.15	15	34.9	21	0.23	27	8	9.19	0.05	146.9	0.002	55	2.04	14	0.025	56.52	6.22	8	16.5
1937390	0.2	0.01	0.8	0.12	50.7	114.6	0.05	0.1	8	0.05	28.8	0.05	68.3	0.002	596	0.45	21	3	72.93	23.83	2	28.8
1937393	4.78	0.11	1	3.15	4.6	10.5	77	0.12	6	2	2.52	3	180.6	0.002	1849	0.54	33	3	54.61	2.2	5	26.1
1937396	0.72	0.01	2.3	4.32	46.1	28.2	0.05	0.05	2.5	0.05	8.04	0.05	183.9	0.002	356	0.1	19	6.3	71.61	5.3	9	51.8
1937399	0.15	0.01	1.1	2.75	31.9	26.7	3	0.1	16	0.05	7.67	0.05	151.1	0.002	556	0.29	10	2.7	74.38	4.91	5	99.4
1937402	1.15	0.05	2.6	2.95	34.4	56	3	0.27	18	0.05	14.54	0.05	204.9	0.002	101	0.09	10	0.025	69.06	11.63	6	97.1
1937405	0.11	0.01	0.3	4.07	0.05	7	2	0.015	2.5	0.05	1.88	0.05	165.5	0.002	99	0.17	10	0.9	76.47	1.26	0.05	45
1937610	5.16	0.1	0.8	3.49	20.2	44.6	62	0.28	11	0.05	11.81	0.05	108	0.005	113	0.06	24	2.1	51.8	8.6	4	315.3
1937612	1.08	0.03	0.6	3.52	31.6	33.4	9	0.19	2.5	0.05	8.42	0.05	108.3	0.005	25	0.025	11	2.9	67.47	7.77	4	90
1946896	8.87	0.18	8	2.91	7.5	17.8	94	0.09	2.5	8	5.52	6	180.5	0.004	167	0.9	42	4.8	48.72	3.16	2	143.6
1946903	7.73	1.3	0.9	1.69	6.3	34.1	6	0.09	2.5	0.05	9.19	0.05	104.2	0.005	25	0.51	10	5.8	42.63	6.49	5	41.5
1946905	5.48	0.1	1.1	4.17	7.7	13	65	0.15	9	2	3.07	2	210.6	0.014	171	0.17	41	1.4	51.45	3.24	3	89.4
1946907	3.28	0.03	19	0.72	16	35.5	101	0.23	269	3	9.6	3	301.5	0.038	12863	0.7	16	1.9	61.24	6.9	8	25.1
1946912	0.16	0.01	0.2	4.85	7.8	48.9	0.05	0.09	2.5	0.05	12.42	0.05	158.1	0.003	171	0.13	10	4.6	68.67	9.43	1	265.6
1947177	2.33	0.05	1.4	1.17	50.8	98.4	41	0.15	113	0.05	27.64	0.05	327.3	0.01	14832	1.69	20	2	58.43	17.7	9	37.5
1947180	1.19	0.01	0.8	1.49	17.6	149.2	14	0.07	9	0.05	41.91	0.05	357.9	0.003	227	0.18	29	3.7	58.29	28.04	8	25.1
1949009	6.65	0.15	0.05	0.09	4.26	52.8	37	0.21	2.5	1	15.36	0.05	217.7	0.002	796	0.87	14	0.7	50.83	8.05	31	20.7
1949015	1.58	0.05	0.8	2.71	26.02	76.5	4	0.15	2.5	0.05	21.99	0.05	280.9	0.002	171	0.13	12	1.3	67.26	13.75	4	65.2
1949018	2.72	0.05	0.3	2.45	9.05	16.4	24	0.18	5	0.05	4.42	0.05	107.9	0.003	705	0.25	15	3.7	68.12	3.17	0.05	56.4
1949021	5.77	0.26	6.2	3.11	7.95	10.4	47	0.18	24	0.05	2.21	0.05	27.5	0.009	2357	0.76	46	3.8	41.04	3.15	3	27.6
1949024	2.29	0.04	0.3	0.07	30.18	51.6	39	0.06	2.5	0.05	12.88	0.05	162.9	0.002	323	0.62	10	3.2	67.08	11.05	3	10.6
1949027	5.18	0.26	0.5	0.11	10.7	41.2	9	0.15	2.5	0.05	11.34	0.05	186.9	0.002	1564	0.52	13	2.6	54.85	7.35	4	14.5
1949029	1.18	0.02	1.1	3.12	32.41	83.1	2	0.16	9	1	23.41	0.05	281	0.002	834	0.11	12	4.4	67.33	15.45	6	46.5
1949038	5.75	0.01	3.8	0.19	40.21	17.8	26	0.07	10	0.05	4.78	0.05	82.7	0.002	25	0.16	15	2.6	62.65	4.29	4	9.7
1949043	4.35	0.6	0.4	2.29	4.61	15.1	20	0.22	2.5	0.05	4.06	0.05	65.8	0.002	25	0.15	10	3.8	24.65	2.81	1	60.7
1949045	6.8	0.05	0.5	1.94	5.47	9.4	56	0.14	8	1	2.54	4	221.9	0.002	63	0.28	41	1.5	53.67	2.43	3	32.6

Samn B#	Tannm	Thoom	Tonnm	Thorm	TiO2%	Tinnm	Tmnnm	ll nnm	Vnnm	W/ nnm	Vnnm	Vh nnm	70 000	7r nnm
1933086	2.8	0.23	0.05	12.37	0.5	0.27	0.27	4.63	112	<b>w pp</b> in 2	12.3	1.72	<b>211 pp111</b> 69	21 ppin 118
1933089		1.78	0.05	104.02	0.38	0.9	1.44	15.32	27	0.05	80.8	9.82	19	310
1933092	3.3	2.21	0.11	31.9	0.7	0.63	1.11	6.45	41	0.05	73.1	6.65	31	440
1937181	1.2	0.49	0.22	9.73	0.43	0.01	0.3	3.98	160	4	16.6	1.5	34	94
1937183	0.4	0.18	0.05	1.06	0.38	1	0.16	0.57	131	0.05	7.4	0.66	91	34
1937186	0.4	0.57	0.05	1.53	1.15	0.01	0.38	1.12	298	10	22.6	2.18	53	73
1937188	2.1	1.2	0.05	10.35	1	0.71	0.82	2.81	137	0.05	48.7	5.12	96	350
1937190	1.6	1.34	0.1	10.24	0.63	0.5	0.75	2.25	77	0.05	47.3	4.12	76	283
1937384	1.1	0.68	0.05	14.82	0.52	0.63	0.38	2.62	84	11	26.3	2.38	11	123
1937387	1.7	0.85	0.1	28.53	0.67	3.1	0.51	1.68	68	9	34.1	3.06	39	145
1937390	5.8	2.96	0.05	52.66	0.65	0.2	1.73	8.58	15	6	102	11.37	13	640
1937393	0.7	0.24	0.05	3.13	1.45	1.8	0.26	2.04	355	6	9.8	1.69	42	104
1937396	12.1	0.75	0.08	27.81	0.47	0.08	0.74	6.97	14	2	36.3	5.37	14	453
1937399	2.8	0.66	0.11	29.26	0.52	0.63	0.55	4.71	40	0.05	26.7	3.54	14	460
1937402	2.7	1.72	0.11	19.97	0.6	0.99	1.02	2.47	23	2	65.6	6.37	83	519
1937405	1.2	0.19	0.05	4.45	0.06	0.45	0.17	1.72	17	0.05	9.2	0.95	15	65
1937610	1.8	1.26	0.05	17.87	1.07	0.35	0.76	1.65	170	0.05	49.4	4.56	88	170
1937612	2.1	1.25	0.05	14.44	0.8	0.24	0.77	2.4	45	0.05	51.7	4.81	51	489
1946896	0.4	0.42	0.05	2.79	0.65	0.4	0.27	1.12	197	1	16.6	1.68	31	40
1946903	0.6	0.66	0.05	11.71	0.27	0.18	0.35	0.88	76	5	23.4	2.16	19	69
1946905	0.4	0.62	0.05	2.83	1.33	0.86	0.44	7.14	349	2	24.5	2.87	163	110
1946907	1.4	0.72	0.48	19.52	0.68	2.21	0.37	45.26	188	6	25.2	2.32	117	174
1946912	0.5	1.15	0.05	5.42	0.28	0.4	0.44	2.82	16	0.05	30.4	2.55	26	209
1947177	4.2	1.98	0.08	67.07	1.21	1.62	1.15	8.47	128	4	77.1	7.3	265	258
1947180	1.5	2.48	0.05	22.72	0.75	0.87	0.71	13.01	116	2	47.1	4.53	23	174
1949009	1.1	0.54	0.18	19.19	0.52	0.73	0.27	4.11	79	12	17.6	1.72	17	95
1949015	2.4	1.55	0.1	36.51	0.66	0.59	0.82	7.46	41	0.05	54.3	5.1	18	420
1949018	0.9	0.42	0.06	13.15	0.44	0.7	0.22	1.3	106	0.05	15.4	1.34	63	133
1949021	0.7	0.63	0.05	3.58	1.56	0.39	0.39	4.2	344	7	25.1	2.52	239	92
1949024	2.7	1.62	0.16	26.33	0.33	0.53	1.16	4.3	46	3	65.4	7.66	47	251
1949027	1.2	0.69	0.12	15.85	0.49	1.03	0.31	1.76	85	5	22.5	1.84	15	107
1949029	3.6	1.88	0.09	71.91	0.62	0.54	1.11	13.06	33	0.05	72	7.02	97	371
1949038	3.6	1.4	0.18	36.12	0.51	0.23	1.85	8.96	14	1	102.3	12.3	133	519
1949043	0.7	0.36	0.05	10.4	0.21	0.23	0.23	1.61	18	1	15.6	1.5	40	66
1949045	0.5	0.48	0.1	4.18	0.98	1.26	0.32	4.1	217	18	19.9	2	215	77

Kym Michael Custance Identifying IOCG pathfinder elements

Samp R#	DH_No	East	North	Depth from	Depth to	Strat_Name	Lith_type	Samp_type	Samp_from	Au ppb	Ag ppm	AI2O3%	As ppm	Ba ppm	Be ppm	Bi ppm	CaO%
1949049	23052	752291	6225060	123.7	124.7	Arthurton Granite	Granite	Drill core	B2	0.5	0.025	13.28	18.7	1317.4	3.5	0.03	0.62
1949051	23674	759856	6246503	126.5	127.5	Oorlano Metasomatite	Metasomatite	Drill core	B2	0.5	0.025	11.38	16.4	84.4	1	1.37	0.35
1949054	23730	775944	6248599	50.5	51.25	Mona Volcanics Member	Felsic volcanic	Drill core	B2	6	0.28	15.65	35.9	442	2.4	0.76	0.65
1949780	139488	759504	6267846	167.3	168.3	Oorlano Metasomatite	Metasomatite	Drill core	B2	0.5	0.06	11.43	16.7	14.6	3	0.05	4.54
1949783	23729	776876	6249028	98.2	99.2	New Cornwall Member	Carbonate	Drill core	B2	0.5	0.46	13.06	22.8	730.9	2.2	0.53	5.43
1949787	23723	777295	6249394	80	81	Mona Volcanics Member	Felsic volcanic	Drill core	B2	0.5	0.26	11.59	20.5	898.2	2.8	0.17	0.22
1949792	23048	770327	6233976	51.3	52.3	Unnamed GIS unit	Sediment	Drill core	B2	1	0.35	15.88	23.9	480.3	4.6	0.56	0.57
1949795	23701	741389	6240671	52.8	53.8	Weetulta Formation	Felsic volcanic	Drill core	B2	0.5	0.26	11.37	26.2	2220.1	1.4	0.005	0.25
1949798	30047	784258	6246312	348.9	349.9	Mona Volcanics Member	Felsic volcanic	Drill core	B2	0.5	0.025	19.13	91.7	588	3	0.63	1.99
1949800	23027	745198	6225492	27.4	28.4	Doora Member	Sediment	Drill core	B2	1	0.025	15.65	22.8	404	3.5	0.005	0.67
1949805	23672	752805	6242398	149.9	150.6	Wandearah Formation	Sediment	Drill core	B2	2	0.025	11.63	5	521	2.1	0.34	1.41
1949810	23046	768768	6233533	58.8	59.9	Wandearah Formation	Sediment	Drill core	B2	7	0.025	16.18	9	259.4	4.7	0.22	0.85
1949813	23669	751171	6243568	126.8	127.8	Oorlano Metasomatite	Metasomatite	Drill core	B2	2	0.025	9.08	8	233.1	1.4	0.005	13.35
1949818	23726	777294	6247803	63.7	64.7	Wandearah Formation	Sediment	Drill core	B2	1	0.4	14.52	14	868.3	3.1	1.18	2.77
1949820	30048	784207	6244524	436.3	437.3	Bute Metadolerite	Mafic intrusive	Drill core	B2	15	0.025	8.53	4	36.9	1.3	0.07	13.24
1949823	30050	781823	6257617	302	303	Wandearah Formation	Sediment	Drill core	B2	0.5	0.025	17.94	11	1567.1	4	0.19	1.35
1949826	23700	741294	6240971	60.6	61.6	Doora Member	Sediment	Drill core	B2	0.5	0.025	19.48	8	1071.5	4.9	0.05	0.37
1949828	23691	760919	6243044	57.7	58.7	Unnamed GIS unit	Sediment	Drill core	B2	0.5	0.025	7.46	3	1749.8	2.4	0.005	26.23
1960844	143972	744329	6163730	49.5	50.5	Unnamed GIS unit	Gneiss	Drill core	B2	0.5	0.025	14.93	4	1198.2	1.7	0.005	0.94
1960847	23028	733429	6208620	46.3	47.3	Wandearah Formation	Felsic volcanic	Drill core	B2	2	0.025	11.06	9	767.1	4.2	0.005	0.45
1960878	23047	769461	6233877	111.5	112.5	Unnamed GIS unit	Sediment	Drill core	B2	54	0.3	12.68	7	69.5	2.6	0.005	1.62
1960881	30065	778298	6242989	168.4	169.4	Wandearah Formation	Carbonate	Drill core	B2	3	0.5	2.43	53	225	1.2	1.75	23.05
1960884	30051	778309	6244444	179.8	180.8	Wandearah Formation	Carbonate	Drill core	B2	0.5	0.025	8.47	19	2293.6	2.6	0.13	12.51
1960887	23037	769374	6233729	63	64	Unnamed GIS unit	Sediment	Drill core	B2	3	0.025	14.44	9	566.1	3.7	0.26	0.42
1960892	23697	770244	6235564	235.7	236.7	Unnamed GIS unit	Mafic intrusive	Drill core	B2	5	0.025	11.55	11	202.1	4.6	0.11	3.04
1960895	23698	769811	6234551	281.1	282.1	Unnamed GIS unit	Sediment	Drill core	B2	6	0.025	15.5	8	745	4.3	0.02	1.85

Samp R#	Cd ppm	Ce ppm	Co ppm	Cr ppm	Cs ppm	Cu ppm	Dy ppm	Er ppm	Eu ppm	F ppm	Fe2O3%	Ga ppm	Gd ppm	Ge ppm	Hf ppm	Ho ppm	In ppm	к20%	La ppm	Li ppm	Lu ppm
1949049	0.01	335	7.9	10	1.49	20	11.14	6.52	2.95	1449	4.83	16.2	13.1	0.38	12.3	2.35	0.031	5.51	173	25.4	1
1949051	0.01	67.1	20.9	44	1.28	426	4.1	2.37	0.93	626	4.14	13.7	4.7	0.025	5.9	0.85	0.041	1.8	35.3	36.6	0.35
1949054	0.01	59.5	24.3	83	6.97	201	4.01	2.2	1	2344	12.78	22.5	4.3	0.74	3.7	0.8	0.03	2.95	28.3	56	0.33
1949780	0.04	37.8	10.9	58	0.44	34	8.87	5.64	1.28	859	12.59	18.9	8.12	0.79	5.8	1.92	0.202	0.1	12.3	3.4	0.9
1949783	0.01	59.8	19.1	65	2.53	154	4.61	2.74	0.74	870	7.74	19.8	5.05	0.84	4.3	0.94	0.164	2.97	24	36.5	0.4
1949787	0.03	178.4	6.6	10	2.34	9	13.62	8.57	2.68	453	4.16	19.1	13.25	0.77	12.9	2.97	0.023	5.77	90.5	21.5	1.39
1949792	0.11	43.9	16.9	76	16.51	34	3.36	2.15	0.59	2981	7.47	21.6	2.96	0.63	5.8	0.73	0.094	6	21.1	42.8	0.35
1949795	0.01	91.8	6.5	10	0.9	16	6.69	4.38	1.14	295	8.99	22.9	7.17	0.52	16.7	1.47	0.0025	8.19	45.1	7.4	0.9
1949798	0.01	112.1	48.5	108	3.9	22	6.64	3.62	1.62	979	6	28.3	7.38	0.55	3.6	1.32	0.088	5.29	55.5	43	0.48
1949800	0.02	38.8	23.1	84	3.31	306	3.98	2.47	0.89	2274	9.66	21.4	3.48	0.33	4.1	0.84	0.038	3.11	20.4	39.8	0.4
1949805	0.01	78.3	12.1	52	1.99	358	3.08	1.88	1.23	1135	4.21	13.2	4.2	0.7	7.4	0.65	0.05	7.04	53.4	40.5	0.34
1949810	0.01	73.4	32.8	88	7.59	110	5.12	3.16	0.89	2300	5.48	22.1	4.93	1	5	1.09	0.03	4.72	39.5	47.1	0.51
1949813	0.01	112.6	35.5	54	1.88	33	2.66	1.86	0.89	782	3.53	10.4	3.51	0.7	6.6	0.6	0.02	2.27	60.9	26.1	0.42
1949818	0.01	86.7	19.3	71	2.71	95	4.67	2.55	0.78	1227	5.77	20.8	5.53	0.6	3.8	0.94	0.07	4.32	43.8	39.7	0.4
1949820	0.01	25.6	48.8	105	6.12	148	6.48	3.4	1.73	332	13.59	16.1	6.44	0.9	1.9	1.28	0.12	0.34	10.9	91.3	0.44
1949823	0.01	44.8	18.3	80	5.01	33	4.89	2.87	0.7	1937	4.55	27.2	3.98	1	3.8	1.03	0.09	5.82	21.9	39.2	0.4
1949826	0.01	245.1	26.5	91	7.93	33	8.46	4.23	1.79	2302	10.22	22.5	12.29	0.9	3.4	1.58	0.03	11.25	113	31.8	0.6
1949828	0.1	52.2	15.4	48	2.94	62	3.38	1.72	0.73	1221	2.45	14.2	3.68	0.5	2.3	0.65	0.03	3.44	26.3	15.8	0.25
1960844	0.01	90.7	14.5	68	3.52	62	3.83	1.97	1.38	767	5.66	20.5	5.03	1	6.1	0.73	0.03	5.84	55.9	30.9	0.31
1960847	0.2	540.3	47.9	10	0.82	117	19.87	10.44	3.72	1730	12.67	16.2	27.2	1.9	17.3	3.86	0.02	5.34	280.7	24.3	1.63
1960878	0.01	36.9	22	42	3.66	173	3.42	2.03	0.7	1172	4.29	14.1	3.32	0.4	6.5	0.73	0.01	2.28	18.5	52.2	0.4
1960881	0.01	33.8	25	25	1.32	64	2.16	1.05	0.61	334	3.27	5.6	2.48	0.4	1.1	0.4	0.04	1.19	14.8	14.3	0.15
1960884	0.01	74.4	9.4	38	3.19	36	4.76	2.45	1.33	787	3.49	13	5.49	0.8	3.7	0.94	0.06	4.05	37.4	21	0.33
1960887	0.1	294.3	34.9	68	6.29	68	5.88	3.37	2.68	1652	6.15	19.6	9.44	1.3	5.8	1.19	0.03	6.3	153.6	74.5	0.47
1960892	0.01	155.7	93.4	10	11.54	78	12.36	6.7	4.06	2577	16.32	24.6	15.11	1.7	6.6	2.44	0.08	2.59	79.6	62	0.99
1960895	0.01	69	6.7	56	9.25	24	2.8	1.59	0.82	1715	4.22	24.3	3.04	0.7	3.5	0.56	0.09	7.07	35.5	45.1	0.25

Samp R#	MgO%	MnO%	Mo ppm	Na2O%	Nb ppm	Nd ppm	Ni ppm	P2O5%	Pb ppm	Pd ppb	Pr ppm	Pt ppb	Rb ppm	Re ppm	S ppm	Sb ppm	Sc ppm	Se ppm	SiO2%	Sm ppm	Sn ppm	Sr ppm
1949049	2.07	0.05	0.6	2.71	26.56	116.2	2	0.24	2.5	0.05	34.8	0.05	223.4	0.002	274	0.22	13	2.2	67.93	18.3	5	80.2
1949051	2.62	0.1	0.8	2.7	10.57	29.1	15	0.28	2.5	0.05	7.64	0.05	79.2	0.002	108	0.13	10	1.4	70.98	5.51	2	20.2
1949054	3.85	0.06	0.7	3.47	13.61	25.1	26	0.16	2.5	2	6.61	4	173.3	0.003	894	0.29	16	3.6	56.17	5.07	2	45.6
1949780	3.2	0.15	0.2	6.75	18.19	28.7	24	0.31	17	0.05	6.17	0.05	11.5	0.002	25	0.18	10	1.6	58.88	7.67	12	15.6
1949783	3.64	0.14	2.3	2.1	14.06	30.4	24	0.31	7	1	7.63	1	75.6	0.009	4965	0.58	12	0.025	56.7	6.21	3	29.1
1949787	0.84	0.02	0.3	1.92	41.71	77.9	5	0.11	6	0.05	20.29	0.05	125	0.003	1008	0.67	15	2.8	73.37	15.04	3	16.9
1949792	3.82	0.06	1.2	0.56	16.7	18.4	37	0.28	2.5	0.05	5.07	0.05	346.3	0.003	6315	0.69	17	8	60.27	3.61	6	24.1
1949795	0.45	0.01	0.4	1.05	37.06	41.9	5	0.2	2.5	0.05	10.79	0.05	217.6	0.002	25	0.23	10	3.4	68.34	8.43	2	68.3
1949798	2.69	0.04	0.2	0.13	7.7	46.9	34	0.17	2.5	0.05	12.67	0.05	218.3	0.002	152	0.67	21	0.025	58.87	8.9	5	12.3
1949800	2.69	0.01	5.2	4.98	8.13	16.1	21	0.15	2.5	0.05	4.36	0.05	182.6	0.002	65	0.05	21	1.2	60.86	3.41	13	51.7
1949805	2.69	0.03	0.6	0.21	6.88	34.6	13	0.27	2.5	0.05	9.74	0.05	221.3	0.002	112	0.14	1	0.025	69.98	5.74	4	19.1
1949810	4.89	0.06	1.2	0.9	10.16	29.4	49	0.21	6	0.05	8.23	0.05	226	0.002	207	0.26	18	0.025	62.34	5.36	2	22.6
1949813	3.7	0.17	0.8	3.21	8.09	37	10	0.19	10	0.05	11.24	0.05	103.2	0.002	954	0.34	1	0.025	53.45	5.49	3	66.4
1949818	3.48	0.1	2.9	0.5	15.22	36.8	53	0.16	7	2	10.08	1	173.6	0.002	273	0.45	14	0.025	61.92	6.93	4	15.8
1949820	8.52	0.45	0.5	0.04	3.89	16.4	110	0.08	5	0.05	3.42	9	15.6	0.002	438	1.6	31	0.025	39.86	5.17	2	110.2
1949823	2.55	0.11	2.1	0.9	10.97	18.8	49	0.15	6	0.05	4.95	6	329.3	0.002	3098	1.26	17	0.025	62.23	3.78	6	13.4
1949826	2.99	0.01	0.5	0.34	27.63	101.3	55	0.25	9	0.05	27.28	2	296.3	0.002	158	0.37	18	0.025	52.17	16.36	2	30.9
1949828	2.9	1.19	1.6	0.72	2.59	20.8	58	0.09	12	0.05	5.71	2	136	0.002	629	0.42	1	0.025	32.59	4.05	2	68.3
1960844	2.41	0.04	0.4	2.8	15.85	40.1	49	0.14	17	0.05	11.17	2	240.8	0.002	165	0.13	10	0.025	64.95	6.51	3	162.8
1960847	1.02	0.03	6.1	2.33	57.19	230.6	100	0.15	10	2	65.96	2	109.5	0.002	57	0.52	1	0.025	65.57	37.7	4	23.9
1960878	3.62	0.06	0.3	2.4	3.65	15.2	27	0.16	2.5	5	4.06	3	127.3	0.002	316	0.18	1	0.025	68.75	3.23	2	13
1960881	14.42	0.78	3.6	0.11	2.81	11.5	49	0.09	7	0.05	3.01	1	39.1	0.002	7199	1.14	1	0.025	19.21	2.48	0.05	55.5
1960884	8.69	0.53	1.1	0.09	6.95	32.3	38	0.13	9	0.05	8.66	0.05	171.6	0.002	1447	0.91	1	0.025	41.01	6.41	2	25.5
1960887	4.47	0.04	0.7	0.07	12.18	98.2	65	0.12	9	2	28.71	1	343.3	0.002	151	0.29	13	0.025	63.47	15.98	2	60.2
1960892	4.07	0.06	0.5	1.57	10.95	74.2	2	1.22	2.5	0.05	18.35	0.05	213.6	0.002	7086	0.31	42	0.025	52.99	15.7	7	30.9
1960895	2.95	0.04	0.4	0.13	12.04	26.2	36	0.14	8	1	7.48	1	288.2	0.002	775	0.47	16	0.025	62.34	4.33	6	62.3

Raw whole rock geochemical data: Central Yorke Peninsula														
Samp R#	R# Ta ppm Tb ppm Te ppm Th ppm TiO2% Tl ppm Tm p			Tm ppm	U ppm	V ppm	W ppm	Y ppm	Yb ppm	Zn ppm	Zr ppm			
1949049	2.9	1.8	0.05	28.55	0.71	0.59	1.05	6.23	45	1	66.9	6.7	55	492
1949051	1	0.64	0.06	17.76	0.41	0.6	0.35	4.47	53	5	23.9	2.27	212	225
1949054	1.3	0.63	0.09	14.05	0.63	0.9	0.33	2.88	116	4	21.7	2.06	75	134
1949780	1.3	1.29	0.08	35.83	0.51	0.01	0.85	7.94	123	1	55.3	5.28	17	212
1949783	1.4	0.72	0.12	19.19	0.61	0.49	0.42	6.58	100	2	26.7	2.71	64	152
1949787	3.9	2.13	0.1	33.78	0.48	0.54	1.38	7.09	12	4	83	8.86	21	479
1949792	1.7	0.5	0.23	20.85	0.66	1.5	0.35	4.24	113	2	20.8	2.25	133	184
1949795	4.1	1.04	0.14	28.48	0.68	0.37	0.72	6.46	27	0.05	46.2	4.98	10	636
1949798	1.3	1.08	0.26	19.02	0.72	0.5	0.54	3.12	162	7	39.1	3.19	29	122
1949800	1.2	0.57	0.11	18.3	0.64	0.44	0.39	4.19	72	5	26.1	2.47	36	138
1949805	1.4	0.53	0.05	16.24	0.47	0.53	0.31	2.33	63	5	18.3	1.97	48	261
1949810	1.5	0.73	0.4	15.85	0.64	0.51	0.49	2.36	104	6	32.1	2.94	57	159
1949813	1.2	0.45	0.05	16.06	0.39	0.25	0.31	5.1	40	0.05	17.8	2.3	37	245
1949818	1.4	0.78	0.1	16.1	0.57	0.56	0.38	3.52	144	4	25.7	2.49	32	130
1949820	0.4	1.03	0.05	1.53	0.88	0.06	0.48	1.81	240	6	36.7	2.94	66	62
1949823	1.8	0.67	0.05	20.04	0.69	0.85	0.44	1.98	93	4	29.1	2.73	20	126
1949826	1	1.48	0.05	13.73	0.68	0.49	0.55	2.52	83	0.05	48.4	3.69	61	118
1949828	0.7	0.53	0.05	9.65	0.32	0.36	0.21	5.37	45	4	19.5	1.58	94	78
1960844	1.1	0.65	0.05	17.28	0.63	0.96	0.24	2.21	70	0.05	21.6	1.8	37	217
1960847	3.8	3.45	0.2	27.12	0.78	0.14	1.56	21.09	52	1	102.4	10.26	79	633
1960878	1.3	0.5	1.2	20.17	0.55	0.22	0.28	3.05	78	13	21.6	2.09	54	243
1960881	0.4	0.36	0.05	3.84	0.15	2.16	0.11	2.7	18	3	12.6	0.95	32	41
1960884	0.9	0.76	0.05	12.02	0.38	0.44	0.3	1.74	45	5	25.7	2.13	24	134
1960887	1.4	1.05	0.2	20.33	0.58	0.98	0.47	3.97	136	13	34	3.13	113	198
1960892	1.8	2.03	0.05	15.76	3.04	0.59	0.94	3.14	82	2	69	6.08	69	227
1960895	1.3	0.46	0.05	18.49	0.56	1.01	0.21	2.62	165	7	16.9	1.62	36	116



# APPENDIX B: SEM IMAGES ORDER: (MAP 1 - 1949789) - (MAP 2,3,4 1949023) - (MAP 5,6 - 1960847)











## APPENDIX C: RAW LA-CP-MS DATA FROM SULPHIDES ANALYSED

#### Raw LA-P-MS sulphide data

Element		Na23	S33	V51	Cr52	Mn55	Fe57	Co59	Ni60	Cu65	Zn66	Ga69	As75	Se82	Mo95	Ag107	Cd111	Sb121	Te125	Ba137	W184	Au197	Pb208	Bi209
923-1-01	Ру	22.63	496005	1.21	0	19.17	465500	232.61	245.37	9704	13.36	0.201	126.49	18.79	11.74	15.48	0	8.28	10.21	6.57	0	0	527.98	149.79
923-1-02	Ру	34.15	508063	0.455	0	555.76	465500	929.13	144.04	3735	41.36	0.08	149.2	15.09	15.86	14.97	0	6.41	14.65	23.23	0	0.054	429.88	184.73
923-1-03	Ру	28.47	549158	1.11	0	35.05	465500	208.51	130.51	5743	13.89	0.138	83.53	0	48.7	10.76	0	5.36	12.36	8.25	0	0	298.69	170.28
923-1-04	Ру	19.97	532322	1.37	0	31.27	465500	81.28	171.58	8651	6.73	0.23	55.76	13.94	41.29	6.15	0	3.71	5.64	2.3	0	0	262.73	133.05
923-1-05	Ру	0	523434	0.213	0	72.71	465500	98.73	154.62	2146	29.83	0	122.84	0	391.48	7.01	0	8.16	5.32	6.63	0	0	274.81	136.1
923-1-06	Ру	0	466954	0.451	0	38.43	465500	69.74	119.13	531	12.72	0	144.2	0	59.58	17.94	0	10.49	0	0	0	0.074	0	130.89
923-1-07	Ру	0	551877	1.38	1.24	198.07	465500	137.42	158.33	5523	39641.85	2.11	125.46	0	613.54	20.96	33.49	35.23	10.09	2.07	0	0.058	295.15	153.49
923-1-08	Ру	0	538132	0	0	6.55	465500	57.11	78.08	15978	2125.47	0	140.11	0	423.97	16.26	42.69	5.04	4.36	0	0	0	733.72	313.01
923-1-09	Ру	63.7	526381	7.54	1.51	44.55	465500	1186.59	141.93	1325	66.18	1.43	157.37	0	15.95	23.3	0	10.41	18.24	19.69	0	0	934.81	237.6
923-1-10	Ру	123.4	466541	10.97	0	88.7	465500	1546.72	100.28	1767	89.97	1.61	715.04	19.31	34.73	20.8	0	63.03	21.84	14.53	0	0	0	247.7
923-2-01	Сру	0	498031	0	0	0	304300	0	0	349077	16.17	0	0	65.15	0	4.71	0	0	0	0	0	0	0.4	0.176
923-2-02	Сру	0	435278	0	0	0	304300	0	0	347645	14.56	0	0	0	0	5.35	0	0	0	0	0	0	0.254	0.209
923-2-03	Сру	0	447556	0	0	0	304300	0	0	349084	20.83	0	0	54.7	0	4.43	0	0	0	0	0	0	0.4	0.59
923-2-04	Сру	0	456491	177.26	23.48	41.54	304300	0	5.52	330786	39.55	1.14	5.08	61.17	0	6.23	1.43	4.21	0	75.47	0.58	0.2	36.04	16.88
923-2-05	Ру	32.58	496725	2.93	2.18	17.01	465500	848.57	352.21	153	76.57	0.259	181.18	0	0.19	0.79	0.43	0.382	1.9	43.18	0.163	0	72.86	23.92
923-2-06	Ру	59.75	567155	0	0	7.71	465500	1136.22	7905.12	933	10.75	0	360.32	47.61	0	3.89	0	0.258	5.64	1.77	0	0	27.74	29.99
923-2-07	Ру	0	490173	0.101	0	1.65	465500	443.59	5108.23	261	2.95	0	50.22	27.62	0	0.69	0	0.049	0	0	0	0	13.94	14.97
923-2-08	Ру	0	551443	5.19	2.61	2.68	465500	914.19	6587.67	281	7.72	0.187	671.21	33.08	1.67	0.97	0	2.26	2.04	31.69	0.095	0	41.59	35.39
923-2-09	Сру	0	497334	0	0	0	304300	0	0	344485	9.31	0	3.78	0	0	3.8	1.33	0.22	0	0	0	0	0.29	0.275
923-2-10	Сру	0	467421	0	0	0	304300	0	0	343024	35.2	0	0	0	0	4.46	0	0	0	5.35	0	0	0.35	0.261
923-2-11	Сру	0	455710	0	0	4.04	304300	0	0	349645	12.19	0	0	0	0	4.41	0	0.29	0	0	0	0	0.57	0.301
923-3-01	Pyr	20.65	532537	5.44	2.45	16.02	465500	2215.17	9184.69	4169	47.71	0.91	3103.26	27.56	0.193	3.49	0	27.8	1.77	46.04	0.098	0	415.9	35.77
923-3-02	Pyr	80.93	538525	25.6	5.82	142.86	465500	1108.26	21029.67	24644	117.21	3.7	4490.07	36.23	0.61	11.97	0	33.71	2.47	722.8	1.32	0	305.08	111.67
923-3-03	Pyr	0	527272	0	0	0.76	465500	770.96	7781.59	1375	4.81	0	61.6	25.27	0	1.74	0	0.91	1.49	0	0	0	31.35	22.08
923-3-04	Сру	0	406647	0	0	0	304300	0	0	347278	16.56	0.22	0	0	0	2.55	0	0.173	0	0	0	0	1.22	1.12
923-3-05	Сру	0	433321	0	0	0	304300	0	0	340959	16.39	0	0	0	0	2.66	0	0	0	0	0	0	0.53	0.99
923-3-06	Сру	0	459540	0	0	0	304300	0	0	341722	13.66	0	0	0	0	3.28	0	0	0	0	0	0	0.41	0.9
923-3-07	Сру	0	452565	1.18	0	0	304300	0	4.51	342046	74.6	0	150.49	0	0	3.38	1.86	4.29	0	8.86	0	0	161.48	2.4
923-3-08	Сру	0	479619	0	0	3.33	304300	6.14	12.05	261745	20.66	0	0	36.27	0	3.22	0	0.46	0	0	0	0	4.88	2.62
923-3-09	Сру	0	384434	0.6	0	0	304300	0	0	344156	25.1	0	0	0	0	3.04	2.32	0.25	0	0	0	0	0.86	1.56
923-3-10	Сру	0	445795	0	0	0	304300	0	0	349103	14.63	0	0	0	0	2.26	0	0	0	0	0	0	0.211	0.461
923-3-11	Ру	0	529582	0.18	0	2.24	465501	1512.32	10235.39	19019	7747.34	0.189	165.1	28.52	0.37	4.57	0	9.47	2.12	3.39	0	0.037	54.45	30.27
923-3-12	Сру	0	450336	0	0	0	304300	0	0	351987	13.67	0	0	0	0	2.33	0	0	0	0	0	0	0.45	0.286
923-3-13	Сру	0	435063	0	0	0	304300	0	0	349861	9.55	0	0	85.54	0	2.59	0	0	0	4.49	0	0	0.47	0.424
923-3-14	Сру	0	438269	0	0	7.97	304300	0	0	347482	118.63	0	9.34	0	0.66	3.82	0	6.21	0	0	0	0	15.54	1.17
923-3-15	Сру	115.06	493632	0	0	0	304300	0	0	341049	13.51	0	0	72.93	0	2.53	0	0	0	0	0	0	0.69	0.78

## **APPENDIX D: PATHFINDER ELEMENT CRITERIA**

### Pathfinder criteria 1 – histogram and probability plots

Normalised histogram and probability plots were used to assess the distribution of data within each element. Elements needed to show three potential populations of data (back ground – interesting – potentially anomalous), display a relatively evenly distributed bell curve set of data and show high population outliers. Example given in (Fig.18-a) *Pathfinder criteria* 2 – *High element populations from histogram need to be anomalous.* Element concentrations needed to be assessed to determine if their high population outliers are in fact anomalous and above background concentrations. Lack of sample points within each selected lithological unit meant that creating a true background for the Wallaroo Basement rocks on central Yorke Peninsula was beyond the scope of this thesis. However, a medium must be used for comparing data as to determine anomalous concentrations. (Fabris et al. 2013b)work on characterising alteration in the Eastern Gawler Craton on identifying an IOCG index also used a 10x average crustal abundance anomalous indicator and utilized an average upper crustal abundance by (Rudnick and Gao 2003) as the datum point for measuring anomalous concentrations against. As both study areas are within the Olympic Copper Gold Province and areas share similar characteristics in alteration styles the same table by (Rudnick and Gao 2003) was used for this project. Table given in (appendix.aa). Example of criteria in (Fig.18-b) Pathfinder criterion 3 - Elements with high populations must be high enough above detection limits to not impede the data, yet with preference given to elements that are already associated with IOCGS

As detection limit issues are a factor for each individual element, elements had to be assessed to determine if the data may be influenced by detections limits for that element. Yet, if an element is considered to be a known interesting pathfinder and vectors towards known mineralisation in the area it may still be further looked at. Detection limits examples for elements given in (Fig.18-c).

Pathfinder criterion 4 – Elements should show vectoring towards areas of known or potential IOCG mineralisation.

Three areas have been identified as prospective areas for mineralisation. The already known mineralised area of Moota-Wallaroo, the Alford prospect and potential mineralised area north of Maitland (Fig. 19.d). Elements that passed pervious criteria had to vector towards these areas and show correlation with other potential pathfinder elements.

