1	Pressure, temperature and timing of mineralization of the
2	sedimentary rock-hosted orogenic gold deposit at Klipwal,
3	southeastern Kaapvaal Craton, South Africa
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20	Abstract
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22	Gold mineralization in the Klipwal Shear Zone (KSZ) at the Klipwal Gold
23	Mine is confined to laminated quartz-carbonate lodes, stringers and associated
24	alteration in sandstone and siltstone of the Delfkom Formation in the upper Mozaan
25	Group of the Mesoarchaean Pongola Supergroup. The moderately dipping brittle-
26	ductile KSZ strikes N-S with an oblique-reverse, sinistral sense of shear. The
27	deformational events that are recognized include an early compressional phase that
28	produced anastomosing shears defined by shear fabrics with numerous shear-parallel
29	laminated quartz-carbonate fault-fill veins and, in places, extensional quartz vein
30	stockworks, and a late brittle reactivation phase that produced fault breccias,

developed: the main R-reef constitutes the KSZ, while the J- and H- reefs represent
footwall splays. Alteration comprises chlorite, muscovite, epidote, feldspar and
carbonates along with pyrite, arsenopyrite and chalcopyrite ± pyrrhotite. An inner

displacing earlier extensional veins. Three closely spaced economic reefs (lodes) are

35 alteration zone is dominated by laminated quartz-carbonate veins with alternating 36 quartz-carbonate-rich and muscovite-chlorite-rich laminae whereas the proximal zone 37 is characterized by alteration halos of K-feldspar, albite, epidote, chlorite and 38 muscovite along with carbonates and associated quartz veins. Chlorite thermometry 39 from the inner and proximal zone yielded temperatures of 267 to 312 °C. 40 Arsenopyrite compositions provide temperatures in the same range, 255 to 318 °C. 41 Fluid inclusion microthermometry and Raman spectrometry of quartz veins in the 42 mineralized reefs reveal the presence of metamorphogenic aqueous-gaseous fluid with 43 an average salinity of 6.5 wt. % NaCl equiv. Fluid compositions and estimated P-T 44 range (1.1 to 2.5 kbar at 255 to 318 °C) are typical of orogenic gold deposits. 45 Devolatilization during the regional facies metamorphism of the Pongola Supergroup 46 is considered the likely fluid forming event with fluid flow focused into a 47 'compressional-jog' of the KSZ. Shear-induced pressure fluctuations generated phase 48 separation of the initial aqueous gaseous fluid producing a gaseous and low saline 49 aqueous fluid. This, together with fluid–rock interaction, and a decrease in fO_2 lead to 50 sulphide and gold precipitation at Klipwal. Re-Os data from six sulfide samples 51 constrain the age of sulfide precipitation and, by inference, gold mineralization, to 2563 ± 84 Ma, with an initial 187 Os/ 188 Os = 0.29 ± 0.08 (MSWD = 0.38). This age is 52 53 distinctly younger than the post-Pongola granites (2863-2721 Ma) ruling out the 54 association of granite emplacement with mineralization. This would suggest that 55 mineralization is linked to the regional D₃ folding event which reactivated the KSZ 56 after emplacement of the post-Pongola granites and that final brittle, post-57 mineralization reactivation is related to Karoo-age faulting. Low initial Os values 58 suggest that ore fluid interacted with mafic rocks, leaching non-radiogenic Os, the 59 likely source being the deeper-seated Nsuze Group volcanics and/or the greenstone 60 belts that underlie the Pongola Supergroup.

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Keywords: Orogenic gold deposits, Fluid inclusions, Klipwal, Kaapvaal craton, SouthAfrica

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65 Introduction
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67 Orogenic gold mineralization constitutes an important class of deposits, collectively 68 representing a significant world gold resource (about 25000 t gold; Goldfarb et al. 69 2005). Although these ores are associated with deformed metamorphic terrains of 70 almost all ages (Kerrich and Cassidy 1994) they mostly occur in the Archaean 71 greenstone belts of Australia, Canada, Africa, India and Brazil, the Proterozoic belts 72 of West Africa, and to a lesser extent, in Mesozoic and Cenozoic provinces (Kerrich 73 and Cassidy 1994; Goldfarb et al. 2001; Groves et al 2003; Hagemann and Brown 74 They are characterized by a strong structural control, distinct alteration 2000). 75 assemblages, ore mineralogy, ore fluid composition, and occur in a range of host 76 lithologies (Groves et al. 1998; McCuaig and Kerrich 1998; Goldfarb et al. 2001; 77 Groves et al. 2003). These include greenstones (metamorphosed from low greenschist 78 facies to granulite facies), banded iron-formations, ultramafic rocks, sedimentary 79 rocks and granitoids (McCuaig and Kerrich 1998, and references therein). Among all 80 these, sedimentary rock-hosted orogenic lode gold deposits form an important and 81 distinctive class (Bierlein and Crowe 2000).

82 Although the sedimentary rock-hosted deposits are well known from the 83 Phanerozoic there are only a handful from the Archaean, which in many ways are 84 similar in mineralization and structural style, alteration and ore mineralogy to the 85 Phanerozoic counterparts. These deposits, also called slate belt-hosted gold deposits, 86 are developed in mineralized shear zones in thick marine sedimentary sequences 87 commonly underlain by bimodal volcanics generated during spreading, arc formation, 88 plate collision and subduction (Goldfarb et al. 1998). They are associated with major 89 translithospheric structures or compressional to transpressional-transtensional shear 90 zones, similar to the Archaean orogenic gold deposits (Bierlein and Crowe 2000; 91 Lawley et al. 2013).

92 Sedimentary rock-hosted orogenic gold deposits are reported from a number 93 of Phanerozoic accretionary terrains, notably the Pacific rim: the North American 94 Cordillera, far east Russia, northeastern China, eastern Australia and New Zealand 95 (Fig. 1 in Goldfarb et al. 1998; Bierlein and Crowe 2000). Some of the major gold-96 bearing districts in the eastern and western part of the Pacific rim include the 97 Hodgkinson gold field in the Hodgkinson-Broken River Fold Belt and Ballarat in the 98 Lachlan Fold Belt from eastern Australia (Phillips and Hughes 1996; Peters et al 99 1990); the Alaska–Juneau, Treadwell, Kensington mines in the Juneau gold belt in the 100 North American Cordillera (Goldfarb et al. 1991; Miller et al. 1995), the Omchak 101 goldfield in the Yana-Kolyma belt in northeastern Russia (Nokleberg et al. 1994;

102 Nokleberg et al. 1996), and the Reefton goldfield in eastern New Zealand (Cooper and103 Tulloch 1992; Goldfarb et al. 1995).

104 The sedimentary rock-hosted lode gold deposits in eastern Australia are 105 similar in some aspects to the Klipwal gold deposit described here. Gold-bearing 106 quartz veins in the Hodgkinson gold field, Queensland Australia, are found in low-107 grade metasediments with restricted hydrothermal alteration halos. Mineralization is 108 concentrated in brittle to brittle-ductile shear zones that occur within reactivated 109 second-generation fold axial planes (Peters et al. 1990). Geological characteristics, 110 isotopic data, alteration mineralogy, and fluid inclusion studies show that upward 111 migrating homogeneous metamorphic or distal magmatic fluids were responsible for the gold mineralization (Peters et al. 1990). The Ballarat East gold field is located in 112 113 close proximity to the Avoca fault (Fairmaid et al. 2011) with gold hosted in large 114 fault-related quartz veins that are stacked in arrays associated with west-dipping 115 reverse faults. Evidence of mixing metamorphogenic fluid with sedimentary 116 formation waters is documented by Fairmaid et al. (2011). The sources of gold are 117 considered to include the underlying Cambrian volcanic rocks, surrounding Paleozoic 118 sediments and Proterozoic continental crust.

119 In South Africa, the Mesoarchaean volcano-sedimentary sequence of the Pongola Supergroup, exposed in the southeastern part of the Kaapvaal craton (Fig. 120 121 1a), is considered contemporaneous with the Witwatersrand Supergroup and similarly 122 contains paleo-placer Au-U conglomerate occurrences (Bullen et al. 1994). In 123 addition to Witwatersrand type paleo-placer deposits, the Pongola Supergroup hosts a 124 number of epigenetic, structurally controlled orogenic-style lode gold deposits (Fig. 125 1b). These include the Wonder Mine situated in the Bumbeni Shear Zone, the Klipwal 126 Gold Mine (KGM) on the Klipwal Shear Zone (KSZ) and Ngotshe Mine located a 127 few hundred meters east of the KSZ (Bullen et al. 1994). As opposed to the Pongola 128 paleo-placer deposits the lode gold deposits have until recently continuously produced 129 gold, especially from the Klipwal Gold Mine (Bullen et al. 1994; Gold 1993; Gold 130 2006). Presently the mine is not in operation.

Here we present a detailed description of the geology of the Klipwal gold deposit, its alteration mineralogy, P-T conditions of mineralization deduced from chlorite and arsenopyrite geothermometry and fluid inclusion studies on mineralized zones (reefs). In addition, we present Re-Os compositions of pyrite and arsenopyrite from the ore and report on the timing of gold mineralization. The Klipwal Gold Mine (KGM) is located in KwaZulu-Natal Province of South Africa, about 15 km south of the Swaziland border, between the towns of Piet Retief and Pongola (Figs. 1a, b). Mining operations commenced during the late nineteenth century with a total gold production of 5.7 tons by 2003. A total of 1.18 million tons of ore were mined between 1981 and 2003, and from these, at least 5.3 tons of gold were recovered at a grade of 4.5 g/t. Mining operations extended fifteen levels to a depth of 454 m. This study is based mainly on samples collected from levels six through to ten.

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144 **Regional geological setting**

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146 The Pongola Supergroup is preserved as two structural basins, the extensive Pongola 147 basin in the north and the smaller Nkandla basin in the south, separated by the 148 Babanango structural high within a stabilized segment of the southeastern Kaapvaal 149 craton (Matthews 1990; Gold 1993; Gold and Von Veh 1995; Gold 2006). The basins 150 comprise a lower volcano-sedimentary sequence, the Nsuze Group and an upper 151 dominantly sedimentary sequence, the Mozaan Group. The Nsuze Group is 152 characterized by 4.6 km of mafic and subordinate felsic volcanic rocks with minor 153 calcareous and siliciclastic sedimentary units. The overlying Mozaan Group, with a 154 maximum thickness of 5 km, comprises arenaceous and iron-rich argillaceous 155 sediments with minor banded ironstones. The Mozaan Group hosts both the placer and lode gold deposits, with the KSZ the most important, displacing interbedded 156 157 sandstone, mudstone, ferruginous siltstone and two diamictite units of the Delfkom 158 Formation of the Odwaleni Subgroup (Fig. 1c).

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160 Structure

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162 The Pongola Supergroup in the central and main Pongola basin is gently deformed 163 and typically metamorphosed under sub-greenschist to greenschist facies conditions, 164 with the exception of high-grade occurrences in Swaziland where granulite facies 165 conditions had been reached locally (Wilson and Jackson 1988; Gold and Von Veh 166 1995; Saggerson and Turner 1995, Mukasa et al. 2013; Horvath et al. 2014). Gold and 167 Von Veh (1995) in accordance with Matthews (1990) proposed three regional 168 deformational events $(D_1, D_2 \text{ and } D_3)$ affecting the main Pongola basin (Fig. 1). D_1 is 169 represented by early NNW-directed thrusts, reverse faults and shear zones, including

170 the KSZ (Fig. 1). During D_1 , ENE-trending F_1 folds developed contemporaneously 171 with NNW-directed thrusting. Following D_1 compressional tectonics, the area was 172 affected by NW-SE extension (D_2) associated with the emplacement of mafic dykes 173 and sills. This was followed by D₃ NE-SW-directed compression which produced 174 major northwesterly trending, open upright F_3 folds that deformed the early D_1 shears 175 and refolded the ENE-trending F_1 folds forming a dome-and-basin interference 176 pattern (Gold and Von Veh 1995). Between D_2 and D_3 a number of granitoid plutons 177 intruded into the Pongola Supergroup, collectively known as the post-Pongola 178 granites (Gold 2006).

The KSZ is interpreted by Gold (2006) as a D_1 structure. It forms a major shear zone extending approximately N-S for about 20 km, characterized by a moderately dipping brittle-ductile shear plane that displays oblique-reverse sinistral sense of shear. The shear zone shows anastomosing shear fabrics with numerous shear-parallel laminated quartz-carbonate veins and, in places, a quartz-vein stockwork. There is also evidence of late brittle reactivation of the shear plane (Gold 2006).

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187 Geochronology of granitoids

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189 The ages of deformation and low-grade regional metamorphism remain poorly constrained due to the lack of reliable age data, whereas the geochronology of 190 191 basement rocks to the Pongola Supergroup and the post-Pongola granites is well 192 established. The basement granitoid rocks of the Anhalt granitoid suite intruded into 193 3300 Ma old greenstone remnants (such as the Nondweni greenstone fragment) in the 194 southeastern Kaapvaal craton (Farrow et al. 1990; Hunter et al. 1992; Robb et al. 195 2006). These granitoids range in age from 3290 to 3028 Ma (Rb-Sr: Barton et al. 196 1983; Farrow et al. 1990; Matthews et al. 1989; U-Pb: Kamo and Davis 1994). The 197 pre-Pongola Tsawela gneiss on the northern side of the Pongola basin formed at 198 3428±22 Ma (U-Pb single zircon: Mukasa et al. 2013). The Nsuze Group volcanic 199 rocks, which non-conformably overlie the basement, were erupted between 2984±3 200 Ma (U-Pb single zircon: Hegner et al. 1993) and 2940±22 Ma (U-Pb: Hegner et al. 201 1984). On the basis of new U-Pb zircon age data, Mukasa et al. (2013) established the 202 period of deposition for the Pongola sequence rocks. Accordingly, the oldest Nsuze 203 group volcanic layers date at 2980 ± 10 Ma, which is similar to the previously

204 published ages, and the uppermost sedimentary layers of the Mozaan group have an 205 age of 2954±9 Ma. The post-Pongola granitoids (cf. Fig. 1b) were emplaced over a 206 period of about 150 Ma, between 2863 and 2721 Ma. The Godlwayo granite has been 207 dated at 2863±8 Ma (Reimold et al. 1993), the Nzimane granite in the Hlabisa area at 208 2739±3 Ma (Thomas et al. 1995) and the Spekboom granite at 2700-2730 Ma 209 (Reimold et al. 1993). Maphalala and Kröner (1993) obtained an age of 2722 ± 6 Ma 210 for the Kwetta Granite. All these ages were obtained from single zircon Pb-211 evaporation method. Mukasa et al. (2013) constrained the emplacement ages for 212 Kwetta and Mswati granites, using U-Pb zircon studies, at 2721 ± 10 Ma and 2723 ± 7 213 Ma respectively.

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215 Geology of the Klipwal Gold Mine

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217 Wall rocks, structures and distribution of the reefs

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219 The Klipwal Shear Zone cuts the Delfkom Formation of the upper Mozaan Group 220 which comprises interbedded sandstone, siltstone, mudstone, diamictite and 221 associated mafic and ultramafic intrusives, possibly related to the ca. 2.8 Ga 222 Usushwana intrusive event. Gold mineralization at the KGM is centered on a convex-223 westward flexure in the KSZ (Fig. 1c). At least three closely spaced economic reefs 224 are developed (Fig. 2a). The main R-reef constitutes the master shear zone while the 225 J- and H- reefs are lodes that occupy footwall splays of varying dip. The fourth Quartz 226 (Q)- reef is less extensive and developed as a footwall splay of the R-reef in the upper 227 levels (Fig. 2a). The Q-reef and a major portion of the R-reef are mostly mined out. 228 The R-reef is a curvy-planar structure dipping steeply (about 70°) in the upper levels, 229 while gradually changing to a gentle dip at deeper levels, to as little as 35°. The most 230 pervasive fabric observed within the KSZ is an N-S trending shear foliation which is 231 best observed underground as it is poorly exposed on the surface. The observed 232 structures constitute an early-formed set indicating compression, which includes the 233 main shear zone, its mylonitic wall rocks, fault-fill veins and sub-horizontal 234 extensional veins. A late phase of brittle faulting of the extensional veins produced a 235 fault breccia. Poles to the major shear foliation show a point maximum corresponding to a mean strike of 014° and a dip angle of 50°E. The bulk of the variation in shear 236 237 foliation orientation is within a dip direction interval from about E to ESE, and a dip

angle range between shallow and moderately steep angles. Some deviation of the shallow-dipping sections to more southerly dips and a corresponding clockwise rotation of the strike are evident in the plot of Figure 2b. The relatively well-defined pole maximum in Figure 2b, however, partly reflects a bias in the readings being taken at underground levels accessible for this study. This effectively reduces the spread of the dip and strike data expected over the entire fault structure.

244 Old mine excavations of the Q-reef at the surface show that the footwall 245 siltstone (quartz-chlorite-carbonate±muscovite-schist) comprises altered and 246 sandstone (metapsammite) in the hanging wall (Fig. 3a). Henceforth, these two rocks 247 are named chlorite-carbonate schist and metapsammite, respectively. Although the 248 shear fabric is not often observed on surface, reactivation of the shear plane is evident 249 from slickensides (Fig. 3b). Shear-related folds, with a NW trending axial planar 250 fabric, are observed in the siltstone (Fig. 3c). Strongly foliated sandstone occurs in 251 close proximity to the shear zone (Fig. 3d).

252 The R-reef comprises a 0.5–5.5 m wide, strongly foliated zone, containing 253 numerous shear-parallel fault-fill laminated quartz veins that range in thickness from 254 less than 1 cm to a few meters (Fig. 4a). The N-S trending foliation in the R-reef 255 forms mylonitic fabrics and locally preserves S-C fabrics indicating a horizontal offset 256 with a sinistral sense of shear (Figs. 4a, b). Shear lenses comprising smoky quartz 257 grains are observed in the mylonites (Fig. 4b). In places, an array of thin subhorizontal 258 extensional veins are observed in the R-reef (Fig. 4c). The contact between the R-reef 259 and its hanging-wall shows evidence of late brittle reactivation indicated by clay-rich 260 fault gouge and/or breccia (Fig. 4d). The fault breccia consists of fragments of fault-261 fill quartz veins as well as country rocks. Sub-horizontal extensional veins are 262 displaced by the later brittle faults (Fig. 4e).

263 The H-reefs form a number of footwall splays that developed from the R-reef, 264 linking across to a sequence of reefs known as the J-reefs (Fig. 2a). The H-reefs are 265 similar to the R-reef in that they are also characterized by brittle-ductile shears, but 266 were not reactivated to the same extent during later brittle faulting. The fabric in the 267 quartz veins is less pervasive, and fault breccia is absent (Fig. 4f). A crosscut at level 268 10 (~350m from shaft surface) from the main R-reef, provided an opportunity to 269 access the H- and J-reef shear zones and quartz veins underground (Fig. 5). These 270 reefs cut across a range of footwall lithologies such as metapsammite, least-altered

sandstone and chlorite-carbonate schist that strike sub-parallel to, and dip at shallowerangles than, the R-reef (Fig. 5).

J-reefs represent footwall mineralization of the H-reef shears (Hilliard 2007) where the H-reef shears cut and displace the contact between least-altered sandstone and chlorite–carbonate schist (Figs. 2a and 5). J-reef ores typically extend for a distance of between 10 and 50 m representing the amount of displacement along the H-reef. Juxtaposition of the sandstone against siltstone along the H-reef shears is considered a critical controlling factor in J-reef development.

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280 Alteration mineralogy

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Petrographic studies reveal an inner and a proximal alteration zones centered on the KSZ. The inner zone is dominated by laminated quartz-carbonate veins with alternating quartz-carbonate and muscovite-chlorite-rich layers. The proximal zone is characterized by an alteration halo of K-feldspar, albite and chlorite, along with carbonate and associated quartz veins surrounding the inner zone.

287 About 30 m away from the KSZ, in the upper level, sandstone in the least 288 altered zone comprises quartz grains with less K-feldspar, plagioclase, muscovite 289 (Fig. 6a) and minor heavy minerals such as zircon, ilmenite and titanite. The clastic 290 components are cemented by quartz. K-feldspar and plagioclase are unaltered (Fig. 291 6a). A weak S_1 foliation is defined by muscovite (Fig. 6a) that is considered to have 292 formed during the development of the KSZ. Siltstone in the least altered zone is 293 characteristically weakly foliated with the schistosity being defined by chlorite and 294 muscovite (Fig. 6b).

In the proximal zone the altered metapsammite consists of microcline, dolomite and magnesite, indicating potassic and carbonate alteration (Figs. 6c, d). Minor chlorite and quartz are also observed in the rock as alteration products, and albitic plagioclase is intensely altered to muscovite (Fig. 6d). Chlorite–carbonate schist in the proximal zone is derived from hydrothermal alteration and deformation of siltstone along the shear zone, producing a strong fabric defined by chlorite and muscovite (Figs. 6e, f). Other minerals in this rock include epidote and quartz.

The inner alteration zone comprises auriferous laminated quartz-carbonate veins.
 Alternating laminae in these veins consist predominantly of quartz-carbonate and
 muscovite-chlorite (Fig. 7a). Carbonates in the veins include dolomite, magnesite and

305 siderite. A sinistral sense of shear is observed at thin-section scale with the 306 development of an S-C fabric in the muscovite-chlorite lamina (Fig. 7b). The fine-307 grained texture and strong foliation indicate that the rocks experienced mylonitization 308 and dynamic recrystallization in the shear zone. Matrix quartz grains display bulging 309 and recrystallized grain boundaries, with some grains displaying subgrain rotation and 310 formation of core-and-mantle structures (Figs. 7a, b, c). These textures indicate plastic 311 deformation at low temperatures (about 300°C; Stipp et al. 2002) and correspond to 312 sub-greenschist facies conditions. Shear lenses consisting of quartz grains are 313 observed in the mylonitic matrix (Fig. 7d). Broken fragments of quartz vein clasts and 314 mylonitic clasts are very well preserved in the fault breccia (Fig. 7e). Antitaxial quartz 315 "strain fringes" within the mylonites (Fig. 7f, g) grew on rigid pyrite and arsenopyrite 316 grains. In addition to quartz, these strain fringes sometimes contain chlorite. The 317 'jigsaw-puzzle' type brittle fractures in the sulfide grains are evident of hydraulic 318 breccia and these fractures are filled with quartz and chlorite (Fig. 7f). Silicate 319 inclusion trails within sulfides occur parallel to the matrix shear foliation (Fig. 7g, h) 320 which suggest that either sulfides overgrowing an existing, unmodified early S_1 321 foliation in the D_3 event, or on a reactivated S_1 - S_3 foliation, potentially during D_3 . In 322 both cases, the sulfide growth is associated with syn- D_3 . As D_3 is a reactivation, 323 presumably makes small angle between S_1 and S_3 , it is difficult to distinguish these 324 two fabrics.

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326 Analytical techniques

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328 Selected thin sections were analyzed for chlorite compositions using the CAMECA 329 SX-100 electron probe micro-analyzer (EPMA) at the DST-EPMA National facility, 330 Department of Geology and Geophysics, IIT Kharagpur. Operating conditions for 331 chlorite analysis were 15 kV acceleration voltages with 20 nA beam currents. The 332 counting time was 20 to 30 s. The beam diameter was set at 1 µm. For analysis of 333 pyrite and arsenopyrite an acceleration potential of 20 kV was used. Beam currents of 334 200 nA and 20 nA for was used for pyrite and arsenopyrite respectively. The counting 335 time for Fe and S was 20s and for As 40s.-Appropriate natural and synthetic minerals 336 were used for standardization. Raw data were corrected with the help of PAP 337 correction program by Pouchou and Pichoir (1984). Back scattered electron (BSE)

imaging was acquired using JEOL JSM 6490 SEM at the Department of Geology andGeophysics, IIT Kharagpur.

340 Nine doubly polished wafers of ~200 µm thickness were prepared for fluid 341 inclusion petrographic study. Microthermometric runs were conducted on five 342 samples with the help of a Fluid Inc. adapted USGS gas flow microscopic heating-343 freezing stage, fitted on a Leica Laborlux D petrological microscope housed at 344 Geological Sciences, University of Kwazulu-Natal, Durban. The unit operates in the 345 temperature range of -195 °C to 700 °C, and is periodically calibrated using distilled 346 water-ice bath (0 °C) and pure CO₂ inclusions (-56.6 °C). Phase changes were 347 observed during heating. Fluid salinity and density values were calculated and 348 isochores were constructed using the FLUIDS software package (Bakker 2003). For 349 type-I inclusions the DENSITY program in package CLATHRATES (Bakker 1997) 350 was used. Equation of state (EOS) of Duan et al. (1992a, b) and Bakker (1999) was 351 used for salinity and density calculation and isochore construction respectively. For 352 type-II inclusions, EOS of Jacobs and Kerrich (1980) was used for calculating 353 density, and for isochore construction Belonoshko and Saxena (1991) was used. For 354 type-III inclusions Thiéry et al. (1994) and Duan et al (1992a) were used for calculation of density and isochore respectively. For type-IV inclusions Bodnar 355 356 (1993) was used to calculate salinity and Zhang and Frants (1987) for density. A 357 Renishaw RM1000B laser Raman probe, attached to a Leica microscope, at the 358 Department of Geology and Geophysics, IIT Kharagpur was used to analyze fluid 359 inclusions. The system is equipped with edge filters to block the Rayleigh lines, 360 confocal configuration, thermoelectrically cooled CCD detector, air-cooled laser, and 361 associated software to acquire and evaluate the spectral data. Irradiation was by the 514.5 nm line of a continuous wave Ar-ion laser, which delivered ~ 8 mW laser 362 power at the sample surface. The acquisition time was 60 seconds. The first order 363 Raman band of silicon at 520 cm⁻¹ was used for routine calibration. The 364 reproducibility of the Raman wave number was set up to be $\pm 1 \text{ cm}^{-1}$. Equation (2) of 365 Burke (2001) is used for quantitative analysis of gas species (X_{CO2} and X_{CH4}) from the 366 respective peak areas. Raman results were also compared to the graphical methods of 367 Thiéry et al. (1994). The observed maximum uncertainty in species composition is 368 369 below 5% between these two methods.

370 For Re-Os analysis, sulfide minerals (pyrite and arsenopyrite) were prepared 371 using traditional methods, crushing without metal contact, heavy liquids, FRANTZ 372 magnetic separation and hand picking. Re and Os abundance and isotope 373 compositions were determined using isotope dilution negative ion thermal mass 374 spectrometry (Selby et al. 2009). In brief, approximately ~400 mg of pyrite/arsenopyrite were dissolved with a known amount of mixed tracer solution 375 (¹⁸⁵Re, ¹⁹⁰Os) in 8 ml of inverse *aqua regia* (1:3 mix of HCl and HNO₃) in a carius 376 tube at 220°C for 48 hrs. Osmium was isolated and purified from the acid solution 377 378 using chloroform solvent extraction (CHCl₃) and micro-distillation methods. Rhenium 379 was isolated using solvent extraction (NaOH-acetone; Cumming et al. 2013) and anion chromatography. Full procedural blanks were 0.1 ± 0.1 and 6.2 ± 5.4 ppt (1 SD; 380 n = 2) for Os and Re, respectively, with an 187 Os/ 188 Os of 0.25 ± 0.02. 381 382 The in-house solution standards (Re std; DROsS) analyzed during the period of these 383 are 0.59773 ± 0.002 and 0.16093 ± 0.0002 (n = 2), respectively, which are identical to 384 those previously reported (Cumming et al. 2012 and references therein).

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386 Chlorite geothermometry

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388 Chlorite compositions have been used for temperature estimation and this was done 389 by: (1) empirical calibrations based on tetrahedral Al content and amount of 390 octahedral vacancy (Cathelineau and Nieva 1985; Zang and Fyfe 1995), and (2) 391 thermodynamic formulation, using intra-crystalline exchange reactions between the 392 chlorite phase components and their temperature-pressure dependence (Vidal et al. 393 2001; 2005). In the latter thermodynamic model four end members are considered: 394 clinochlore $[Si_3Al_2Mg_5O_{10}(OH)_8]$ daphnite $[Si_3Al_2Fe_5O_{10}(OH)_8]$, Mg-amesite 395 $[Si_2Al_4Mg_4O_{10}(OH)_8]$ and sudoite $[Si_3Al_4(Mg,Fe)_2 \Box O_{10}(OH)_8]$. These end members 396 are necessary to model the (i) tschermak (TK), (ii) Fe-Mg (FM), and (iii) 397 dioctahedral-trioctahedral (DT) substitutions in chlorite. Temperatures of chlorite 398 formation in alteration and ore zones (reefs) have been calculated from the equations 399 of state of three intra-crystalline equilibria (eqns. 1 through 3), at a pressure of 2 kbar. 400 The chosen pressure is justified from the fluid inclusion studies discussed below.

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402
$$16daph + 15sud = 20Fe-am + 6clin + 35qtz + 20H_2O$$
 (1)

$$403 \qquad 2clin + 3Mg-sud = 4Mg-am + 7qtz + 4H_2O \tag{2}$$

$$404 \qquad 4daph + 6Mg-sud = 5Fe-am + 3Mg-am + 14qtz + 8H_2O \tag{3}$$

406 Two samples each from the R- and J- reefs were selected for electron 407 microprobe analysis. Representative chlorite analyses, their structural formulae and 408 estimated temperatures are given in Table 1. Average temperatures, estimated using 409 the empirical calibrations fall in the range of 267–301°C. Similarly, average temperatures, computed using reactions (1) through (3) are in the ranges of 410 411 268-312°C. Hence, both approaches furnished comparable temperatures, with 412 reasonable standard deviations. However, it should be noted that whenever the estimated temperature values are high, the calculated X_{Fe}^{3+} values are too low. In 413 order to reduce the temperature, a_{H2O} is decreased to as low as 0.2. This is mainly 414 because a decrease of X_{Fe}^{3+} leads to a related decrease of octahedral Al and vacancy 415 with increase in octahedral summation, consequently leading to an increase in the 416 417 estimated temperature (Vidal et al. 2005; 2006). Similarly a decrease in a_{H2O} leads to 418 a decrease in the equilibrium constants of reactions (1) through (3) and therefore 419 decreases the estimated temperature. A decrease in fluid content in the rock suggest 420 periods of reduced fluid flow due to closure of the fractures during interseismic stage (Sibson, 2001). In the present case, reduced a_{H2O} values were used to make the 421 422 temperatures compatible with those obtained by empirical calibrations.

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424 **Ore mineralogy**

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426 In the inner zone, the dominant sulfide minerals are, in decreasing order of 427 abundance, pyrite and arsenopyrite, with minor chalcopyrite, pyrrhotite and galena. In 428 the proximal zone pyrite, arsenopyrite, chalcopyrite, pyrrhotite, and sphalerite occur. 429 Ilmenite, rutile, and titanite occur in both zones. Ore mineral aggregates are aligned 430 along the sheared and mylonitized fabric. Texturally, three types of pyrite and two 431 types of arsenopyrite are present (Figs. 8a and 9). Pyrite-I ranges in size between 432 100µm and >1mm. It forms porphyroblasts with the shear foliation wrapping around 433 them (Figs. 7g, h, 8a and 9). Euhedral grains of pyrite-I contain numerous randomly 434 to preferentially oriented silicate inclusions mimic matrix foliation (Figs. 7g, h and 435 8b). Sulfides such as chalcopyrite, pyrrhotite and galena and gold also occur as 436 inclusions (Figs. 8c, d). Pyrite-II, which overgrew pyrite-I, has also euhedral grain 437 boundaries and is almost free of silicate inclusions (Figs. 8a, d) but contains 438 inclusions of other sulfides such as chalcopyrite, arsenopyrite-II and pyrrhotite (Fig.

439 8d). Pyrite-III has a characteristic irregular outline with or without silicate inclusions 440 and occurs as elongated grains aligned either along the shear foliation or overgrowing 441 this deformation fabric (Fig. 8a). Pyrite-I contains up to 2.8 wt % As and may be 442 referred to as arsenian pyrite (cf. Large et al. 2009). SEM-back scattered electron 443 (BSE) images show that these pyrites are weakly zoned with respect to As showing 444 As-poor cores (Fig. 8e). Pyrite-II shows concentric As-rich and As-poor zones (Figs. 445 8f, g) but also irregular As zoning patterns (Fig. 8h). Arsenopyrite-I is represented by 446 large euhedral grains with abundant randomly oriented silicate inclusions and may 447 contain inclusions of pyrite-II (Fig. 8b). Arsenopyrite-II is euhedral to subhedral in 448 shape, smaller in size and free of any silicate inclusions (Figs. 8c, d). Arsenopyrite-II 449 in places overgrows pyrite-I (Fig. 8c). Gold occurs mostly in association with sulfides and occasionally as free gold grains in quartz veins. Gold grains are generally 10-20 450 451 μm in size and occasionally as large as 100 μm. Gold occurs as

452 (1) inclusions within pyrite-I, at the interface with other sulfides like chalcopyrite453 with pyrite (Fig. 10a),

454 (2) inclusions within pyrite-I without other sulfide inclusions (Fig. 10b),

455 (3) inclusions within arsenopyrite-II (Fig. 10c), and

- 456 (4) free gold in the silicate matrix in close proximity to sulfides and chlorite (Fig.457 10d).
- 458

459 Arsenopyrite geothermometry

460

461 As described earlier, arsenopyrite- I and II occur in association with pyrite in the 462 proximal and inner alteration zones. The temperature of formation can be inferred 463 from arsenic contents of arsenopyrite as described in Kretschmar and Scott (1976) and 464 Sharp et al. (1985). In selected samples (KU10A and KU16) arsenopyrite grains were 465 analyzed by electron microprobe. Representative analytical data and the deduced 466 temperatures are summarized in Table 2. In sulfur rich assemblage (arsenopyrite with 467 pyrite and/or pyrrhotite) arsenopyrite may contain less than 30 atomic % arsenic due 468 to non-equilibrium feature reflecting the kinetics of growth of arsenopyrite and local fluctuations in f_{S2}/f_{As2} (Kretschmar and Scott 1976). Temperature values were inferred 469 up to ~28.4 atomic % arsenic. The estimated temperatures vary from 255 °C to 470 471 318°C, while the corresponding $\log f_{S2}$ falls in the range of -9.9 to -12, comparing 472 well to the temperatures obtained from chlorite geothermometry.

474 Fluid inclusion studies

475

476 Fluid inclusion studies, involving inclusion petrography, microthermometry and 477 Raman spectroscopy in quartz veins from all reefs were carried out. A total of nine 478 doubly polished wafer sections were prepared for examination and five samples (two 479 from R- and J-reefs and one sample from H-reef) were selected for 480 microthermometry. The choice of these samples was based on the availability of 481 workable inclusions within suitable Group of Synchronous Inclusions (GSI, cf. Touret 482 2001). Grains that showed indications of dynamic recrystallization and inclusions 483 showing stretching or leakage were carefully avoided.

484 Inclusions are generally small in size ranging from less than 2 μ m to 10 μ m. 485 Inclusions occurring as isolated, clustered and as intra-granular trails were considered 486 as primary and pseudo secondary, respectively, and selected for microthermometric 487 runs. Inclusions were classified on the basis of disposition and phase content in 488 ambient laboratory conditions, and grouped into four types. Type-I aqueous gaseous 489 inclusions contain a dark gas-rich bubble surrounded by aqueous liquid (Fig. 11a). 490 The volume percent of gaseous phase varies from 40 to 90 % (Table 3). At times 491 inclusion walls are decrepitated due to high internal pressure while heating the 492 inclusions. Type-II and type-III inclusions are monophase gaseous inclusions in which 493 type-II inclusions exclusively contain pure methane as a gas phase (Figs. 11b, c). 494 Type-III inclusions contain a gas mixture of CO₂ and CH₄ with varying proportions 495 (X_{CH4} up to 0.31; Table 3) (Figs. 11b, d). Type-IV inclusions are low salinity aqueous 496 biphase inclusions (Fig. 11e) that frequently occur within the same clusters as type-II 497 and III inclusions (Fig. 11f) and in places they also occur as intra-granular trails. 498 Vapor occupies 10 to 20 % of volume in type-IV inclusions.

499 For type-I inclusions complete microthermometry data could be obtained on 500 only 21 inclusions (12 from R-reef and 9 from J-reef; Table 3). The temperature of 501 CO_2 melting (T_{m,CO2}) varies from -57.8 to -63.2 °C and the temperature of CO_2 vapor 502 homogenization (T_{h,CO2}) varies from -15.5 to +7.8 °C. Clathrate melting (T_{m,Cl}) varies 503 from +7.8 to +14.8 C. Temperature of total homogenization (T_{h.tot}) varies from 272 to 504 367 °C. For some of the inclusions decrepitation temperatures were recorded and the 505 final homogenization was into a gaseous phase. The temperature of CH₄ vapor 506 homogenization (T_{h,CH4}) for type-II inclusions varies from -94.3 to -84.3 °C (Fig.

507 12a). The temperature of CH_4 melting $(T_{m,CH4})$ could not be measured because the stage only cools to -196 °C using liquid nitrogen as cooling agent. For type-III 508 509 inclusions the observed variation in $T_{m CO2}$: was -63.9 to -56.6 °C (Fig.12b). $T_{h CO2}$ 510 varied from -20.1 to +18.3 °C (Fig.12c). From Raman spectrometric analyses CH₄ 511 (Fig. 11d) is the only gas identified in these inclusions apart from CO_2 . For type-IV 512 inclusions temperatures of last ice melting ($T_{m,ice}$) varied from -16.1 to +0.2 °C (Fig. 513 12d) and final liquid vapor homogenization (T_{h.tot}) was into liquid phase with values 514 ranged from 115 to 302 °C (Fig. 12e).

515 The average calculated salinity for type-IV inclusions is 6.5±4.3 (wt% NaCl 516 equiv.). The isochore intersection geobarometric method described by Roedder and 517 Bodnar (1980) was adopted to estimate the entrapment P-T conditions of the 518 inclusions. Coexisting carbonic (type-III) and aqueous (type-IV) inclusions occurring 519 in the same GSI are considered as coeval inclusions (Fig. 11f). The assumption is that 520 these inclusions were entrapped simultaneously and no post-entrapment modifications 521 of the inclusions had taken place. Two intersection points, from the isochores of these 522 inclusion types, IS-1 (275 °C and 1.8 kbar) and IS-2 (287 °C and 1.5 kbar) furnished 523 P-T values (Fig 13). In addition, pressure values were further inferred from the 524 intersection of the isochores with chlorite (CT) and arsenopyrite (AT) 525 geothermometry (Fig 13). This intersection additionally furnished a pressure range of 526 1.1 to 2.5 kbar with the thermometry by the two above independent approaches yielding comparable P-T values. This further supports the use of the isochore 527 528 intersection method.

529

530 **Re–Os geochronology**

531

532 Two samples from each of the three mineralized reefs were selected for Re-Os 533 analysis to determine the age of sulfide and gold mineralization. The samples were 534 collected from the fault-fill laminated quartz veins in the inner zone that contains a 535 profuse amount of sulfides in very close association with the gold mineralization. The Re-Os data for the six samples are presented in Table 4. The pyrite and arsenopyrite 536 grains contain between ~0.2 and 4.7 ppb Re and 32 and 240 ppt Os. A significant 537 portion of the Os budget comprises ¹⁹²Os (7.7 to 45 ppt). The ¹⁸⁷Re/¹⁸⁸Os values are 538 low and range from ~19 to 210. These values positively correlate with the 187 Os/ 188 Os 539 540 from ~1.11 to 9.44. Regression of the Re-Os data including rho using isoplot v. 4.15

541 (Ludwig, 1980) and the ¹⁸⁷Re decay (1.666 e⁻¹¹ a⁻¹; Smoliar et al. 1996) yield a Model 542 1 Re-Os age of 2563 \pm 84 Ma (MSWD = 0.38), with a relatively unradiogenic initial 543 ¹⁸⁷Os/¹⁸⁸Os value of 0.29 \pm 0.08 (Fig. 14).

544

545 **Discussion and conclusions**

546

547 Regional and mine scale structures

548

549 Although three regional-scale deformation events $(D_1, D_2, and D_3)$ have been 550 described (Gold and Von Veh 1995 and Gold 2006), only an earlier set of ductile-551 brittle structures and a final brittle deformation are recognized in the mine. Since 552 mineralization postdates the emplacement of the post-Pongola granites which 553 occurred between D_2 and D_3 , the initial development of the shear zone is considered 554 to have developed during D_1 with mineralization occurring during D_3 involving 555 folding and shear zone reactivation. D₁ generated the initial KSZ geometry (R-reef) and footwall splay (H-reef) during NNW-oriented thrusting. The D₂ event, which is 556 557 regionally related to the emplacement of mafic dykes and sills and which was 558 followed by the emplacement of the post-Pongola granites, is not recognized in the 559 mine as forming any distinct meso- or microscale structures.

560 The area south-west of Swaziland, where the Mozaan Group is most 561 extensively developed and which includes the Klipwal mine, is characterized by the 562 lowest-grade metamorphic imprint of all the exposed Pongola Supergroup. This 563 region was classified as "unmetamorphosed" by Saggerson and Turner (1995), 564 although it should more appropriately be referred to as "very-low grade metamorphic" 565 with no evidence that regional metamorphism ever exceeded lowermost greenschist 566 facies during any of the deformational events. If the original KSZ formed during D_1 , under very-low grade metamorphic conditions, the subsequent reactivation of 567 568 structures and mineralization during the regional D₃ event occurred under similar 569 temperature conditions, making the two events difficult to distinguish. Nevertheless, 570 microstructural evidence confirms post-D₁ ductile deformation. As sulfide growth 571 generally postdates the early-formed (D_1) shear zone, the sulfides can be used as 572 microstructural markers with respect to ductile overprinting of D_1 structures during 573 D₃. Critical evidence includes the preservation of an early-formed foliation as 574 inclusion trails within sulfides, and sulfide-matrix relationships (strain fringes,

575 foliation wrapping around sulfides, while late-formed sufides overgrew the foliation; 576 Figs. 7g, h, 8a and 9). Thus, the period of sulfide growth overlaps to a large extent 577 with D_3 deformation.

578 The thermobarometric data extracted from the mineralized rocks, as well as 579 the related structures, are considered to represent the metamorphic-structural D_3 580 overprint as supported by the consistency of the data between silicate equilibria and 581 sulfides. Mineralization was related to a substantial influx of fluids during D₃. Fluid 582 pathways exploited the D₃-modified and reactivated KSZ and its related vein systems 583 as well as lithological contacts, becoming sites of sulfide-gold precipitation. The 584 actual origin of the mineralizing fluid remains to be discussed, but late fluid 585 infiltration can be recognized on a regional scale. Saggerson and Turner (1995) note 586 that post-peak hydrous alteration of metamorphic mineral assemblages is widespread 587 in the Pongola Supergroup. The final (post-D₃) brittle effects on the shear zone 588 produced fault breccia comprising fragments of quartz veins, mineralized domains 589 and mylonitic wall rocks (Fig. 4d). This event is most likely part of the regional Karoo 590 extensional faulting related to the breakup of Gondwana.

591 The J-reef with higher gold grades was developed at the contact between 592 metapsammite and chlorite-carbonate schist (Figs. 2a and 5). Hence the original 593 lithological discontinuity provided a fluid conduit for J-reef mineralization. On the 594 regional scale, the KSZ strikes approximately N-S. However, at the KGM there is a 595 change in direction from the N–S to a NNE–SSW orientation (Fig.1c). This change in 596 the shear zone geometry, is considered to be a result of the regional D_3 folding event 597 which produced a convex westward flexure in the KSZ and generated a 598 'compressional jog' through which regional fluid flow was driven (cf. Cox et al. 2001; 599 Sibson, 2001).

600 The shear zone contains profuse laminated quartz veins that are developed as 601 fault-fill veins (Fig 4a) due to formation of microfractures along the shear planes. 602 Microfracturing, along grain boundaries, is the dominant mechanism for the formation 603 of grain scale porosities in the ductile regime (Knipe and McCaig 1994; McCaig 604 1997; Mancktelow et al. 1998; Kolb et al. 2004). These microfractures generate 605 fracture porosity and increase fluid permeability along the shear zone (Cox et al. 606 2001), increasing pervasive fluid flow through the shear zone. Fractures provided the 607 open space for rapid flow of the gold-bearing ore fluids. The 'jigsaw-puzzle' type 608 brittle fracturing with angular fragments observed in sulfide grains (Fig. 8f) are

609 interpreted to represent hydraulic breccias which point to hydraulic fracturing, as a610 mechanism for fracture formation, as described by Robert et al. (1995); Kisters et al.

- 611 (2000) and Kolb et al. (2004).
- 612

613 Extensional veins and pressure fluctuation

614

615 Oblique or sub-horizontal extensional vein arrays are also observed in association with the shear zone and fault-fill veins (Fig. 4c). These veins represent hydraulic 616 617 extension fractures, which opened during vein filling episodes by fluid pressures (P_f) 618 in excess of the lithostatic pressure (i.e., $P_f \ge \sigma_3 + T$, where T = tensile strength of the 619 rock) (Robert and Brown 1986). Extensional fractures typically develop parallel to σ_1 620 when $\sigma_{3}' (\sigma_{3}' = \sigma_{3} - P_{f})$ equals or exceeds the tensile strength of the rock. This situation is only possible under conditions of low differential stress. Hence, 621 622 extensional fracturing (possible at negative values of σ_3) can only be attainable by 623 elevated fluid pressure in the inferred compressional environments of formation of 624 orogenic gold deposits (Sibson et al. 1988; Sibson 2001; Robert and Poulsen 2001).

625 Kolb et al. (2004) demonstrated the effect of change in shear zone geometry 626 for economic gold mineralization in the world class Hutti gold mine in the Hutti-627 Muski greenstone belt, eastern Dharwar craton, India. Mishra and Pal (2008) reported 628 oblique sigmoidal extensional veins, similar to the subhorizontal extensional veins in 629 the KSZ in the Hira-Buddini mine from the same Hutti-Muski greenstone belt. These 630 veins were formed by hydraulic fracturing during brittle-ductile shearing. At the Val 631 d'Or lode gold deposits at Quebec Canada, Bouillier and Robert (1992) established 632 that successive cycles of opening and collapse in subhorizontal extension veins 633 correlated with opening and slip on high-angle shear veins. They interpreted these 634 observations to be a result of fluid pressure fluctuations in successive coseismic-635 interseismic cycles (McCuig and Kerrich 1998). Formation of laminated quartz veins 636 requires episodes in which fluid pressure exceeds the local normal stress on the fault. 637 Hence, a crack-seal and/or a fault valve mechanism are interpreted to have operated at 638 KGM during the mineralization event producing the laminated veins.

639 Fluid inclusion density variation further supports pressure fluctuation during 640 gold mineralization. The wide distribution in $T_{h,CO2}$ values of Type-III inclusions 641 indicates a significant variation in density (Table 3), which is attributed to fluctuation 642 in fluid composition and/or pressure. In order to determine the exclusive effect of 643 pressure, $T_{m,CO2}$ – $T_{h,CO2}$ plots (Fig. 12f) were prepared for inclusions with maximum 644 lowering in $T_{m,CO2}$ up to -57 °C, i.e., pure CO₂. The plot shows significant variation 645 of $T_{h,CO2}$ for a near-constant $T_{m,CO2}$, implying fluid pressure fluctuations at the time of 646 entrapment (cf. Dugdale and Hagemann, 2001).

647 Crystal plastic deformation microstructures are observed in the quartz grains 648 but are absent in feldspar grains, which suggests that the deformation occurred at 649 greenschist facies conditions (Scholz, 1988). Dynamic recrystallization structures of 650 quartz grains shows bulged and recrystallized grain boundaries, subgrain rotation and 651 core-mantle structures indicative of pressure solution and intracrystalline plastic 652 deformation at temperatures of about 300°C (Stipp et al. 2002). The alteration mineral 653 assemblage consisting of chlorite-muscovite-carbonates±epidote-quartz-pyrite-654 arsenopyrite surrounding the shear zone in the host rocks is characteristic of low to 655 sub-greenschist facies conditions. Estimates from fluid inclusion isochore 656 intersections coupled with the chlorite and arsenopyrite thermometry (Fig. 13) further confirm that the P-T conditions (255-318°C and 1.1 to 2.5 kbar) were at sub-657 658 greenschist facies. These temperatures and the observed alteration mineralogy in the 659 proximal and inner zone are consistent with the typical mesozonal orogenic gold deposits elsewhere (McCuaig and Kerrich 1998; Groves et al. 1998; Goldfarb et al. 660 661 2001; Groves et al. 2003; Elmer et al. 2006).

662

663 Ore fluid composition

664

665 Irrespective of the host rock, metamorphism and age, the observed fluid composition 666 in orogenic gold deposits shows a very narrow range which is in general aqueous-667 gaseous, low saline metamorphic and/or distant magmatic (Mikcuki 1998; McCuiag 668 and Kerrich 1998). For the Klipwal deposit, fluid inclusion studies reveal that the 669 original mineralizing fluid composition is H₂O-CO₂-CH₄-low salinity (~6 wt.% NaCl 670 equiv.). This is comparable with orogenic gold deposits around the world (Table 5) in 671 general, and sedimentary rock-hosted, Phanerozoic counter-parts in the Pacific Rim 672 (North American Cordellera, Paleozoic-Mesozoic orogenic belts in Asia and in 673 eastern New Zealand) in particular (Bierlein and Crowe 2000).

674

675 Mechanisms of mineralization

677 At Klipwal, gold mineralization is localized in a brittle-ductile, oblique-reverse, 678 sinistral shear zone and associated fault splays. Gold is confined to laminated quartz 679 veins within the sheared host rocks and also occurs in the alteration halo. It is 680 postulated that the initial, auriferous, sulfur-rich, low salinity H₂O-CO₂-CH₄ fluid was 681 transported to near-surface levels via a deep-rooted brittle-ductile shear zone, located 682 at the contact of contrasting lithologic units and at a westward flexure of the KSZ, 683 which acted as the favorable site for high-volume fluid flow. Pressure cycling (Robert 684 et al. 1995) or the fault-valve mechanism (Sibson et al. 1988; Sibson 2001) is 685 indicated by the presence of coexisting aqueous and carbonic inclusions that show 686 wide variations in density (Fig. 12f). Although P-T values obtained by inclusion 687 thermobarometry coupled with chlorite and arsenopyrite geothermometry (1.1 to 2.5 688 kbar and 265 to 315°C) are comparable with the P-T window of the orogenic gold 689 deposits (Table 5), there is convincing evidence of near-isothermal pressure 690 fluctuation (about 1.4 kbar). Pressure fluctuation facilitated phase separation of 691 gaseous and aqueous fluid (Wilkinson and Johnston, 1996; Mikucki, 1998) resulting 692 in a decrease in total sulfur content of the ore fluid, and leading to precipitation of free 693 gold in quartz veins along with chlorite. Fluid-wall rock interaction, on the other 694 hand, was responsible for the association of gold with sulfides where a decrease in fO_2 695 occurred (cf. Mikucki 1998). The presence of type-II pure CH_4 bearing inclusions and 696 type-III CH₄-rich inclusions in close association with CO₂-rich carbonic inclusions is 697 evidence for a decrease or fluctuation in ambient fO_2 conditions.

698

699 Timing of mineralization and source of ore fluid

700

701 The relative timing of mineralization is important for identifying the source of ore 702 fluid and gold. Two possible models have been proposed for the source of gold-703 bearing hydrothermal fluid. These include (1) prograde metamorphic devolatilization 704 of host rocks; (2) magmatic fluid originating from extensive regional or specific 705 granitic intrusions (Hagemann and Cassidy 2000 and references therein; Tomkins 706 2013). Archaean orogenic lode gold mineralization, in general, is formed at a late 707 stage in the tectono-magmatic evolution of the host terrane. Most of these gold-quartz 708 lode veins are formed after peak metamorphism of the immediately surrounding host 709 rocks, in greenschist facies deposits (Groves et al. 1998; Ridley and Diamond 2000; 710 Goldfarb et al. 2001; Groves et al. 2003).

711 At Klipwal, the pyrite and arsenopyrite Re-Os data provide the timing of gold mineralization at 2563 \pm 84 Ma. The initial Os isotope composition (¹⁸⁷Os/¹⁸⁸Os_i) from 712 713 these sulfide grains can be used to infer the source of sulfides and by inference ore 714 fluid and gold. In general, the mantle has relatively low amounts of Re, when 715 compared with crustal rocks, with respect to Os concentration (Kirk et al. 2002). This 716 is because crustal rocks are the products of partial melting of the mantle and 717 potentially re-melted products of previously formed crust. During partial melting Re 718 partitions more readily into the melt and as a result crustal rocks have higher Re/Os values and thus rapidly evolve to develop elevated ¹⁸⁷Os/¹⁸⁸Os ratios (Kirk et al. 2002; 719 Kirk et al. 2003). The low non-radiogenic initial 187 Os/ 188 Os_i (0.29 ± 0.08) value 720 721 determined from the Re-Os data (Fig. 14) suggests that the sulfides and, by inference, 722 gold were originally derived from more primitive sources (cf. Reisberg et al 1991; 723 Selby 2007; Moreli et al. 2007), the most likely being mafic volcanic rocks in the 724 underlying Nsuze Group or mafic-ultramafic greenstone belts, such as the Nodweni 725 greenstone belt, which released non-radiogenic Os.

The post-Pongola granitic intrusions (between 2863 and 2721 Ma) had been considered as a potential source of the ore fluid for the Klipwal mineralization. However, fluid inclusion studies provided no evidence of magmatic fluid components and Re-Os sulfide geochronology, even with the large error attached to the age date (2563±84 Ma) shows that mineralization is much younger than the post-Pongola granitic intrusions. Hence, there is no basis for relating these granites to gold mineralization.

733 Instead, we propose that devolatilization during metamorphism of the Pongola 734 volcano-sedimentary sequence and perhaps deeper-seated greenstones of the 735 Kaapvaal basement, previously metamorphosed at sub-greenschist to greenschist 736 facies conditions, generated the required amounts of fluid. At sufficiently deep levels, 737 a second phase of metamorphism at higher temperatures would cause dehydration of 738 these rocks, with fluids migrating upwards and overprinting successively higher-level 739 rocks, whatever their original metamorphic grade. These fluids may have mixed with 740 mantle components or interacted with mafic to ultramafic rocks at depth, carrying 741 metals to shallow crustal levels with fluid flow locally focused into the Klipwal shear 742 zone, precipitating gold-quartz and carbonate veins.

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- 744

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746

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Fig. 1. (a) Locality map showing position of the main Pongola basin in South Africa.
(b) Regional geological map of the Pongola basin showing major structural features
(modified after Gold 2006). (c) Geological map of the Klipwal Gold Mine showing
the Klipwal Shear Zone within rocks of the Mozaan Group (modified after Gold and
Von Veh 1995). PR: Piet Retief; P: Pongola.

1132

Fig. 2. (a) Subsurface geological cross section across the Klipwal Shear Zone showing
disposition of various reefs and alteration halos (modified after Hilliard 2007). (b)
Poles to the shear zone foliation; equal area lower hemisphere projection. See text for
discussion.

Fig. 3. Field photographs from Klipwal mine area (a) showing old mine excavation
with siltstone in footwall and sandstone in the hanging wall. (b) Slickensides (dashed
lines) on the siltstone surfaces. (c) Shear related fold with NW trending axial planar
fabric (broken line). (d) Sheared and mylonitized fabric in sandstone.

1142

Fig. 4. Underground mine photographs showing different features of shear zone,
laminated fault-fill quartz veins and alteration zones. (a) Laminated quartz vein in the
inner zone preserving S-C fabric with sinistral sense of shear. (b) Oblique S-C fabric
with sinistral sense of shear in the wall rock mylonites along with quartz shear lenses.

(c) Sub-horizontal extension vein arrays associated with fault-fill vein in the shear
zone. (d) Fault gouge in the R-reef showing fault breccia with fragments of host rocks
and quartz veins and mylonites. (e) Extensional veins are displaced by the later faults.
(f) H-reef fault-fill quartz vein and mylonitic shear zone contact with absence of fault
breccia.

1152

Fig. 5. Geological map along a crosscut from the main drive (R-reef) at level 10 which cuts across H- and J- reefs. Stars denote the sample locations. The map shows underground geology, alteration halos around the reefs and the attitude of shear foliation in the quartz veins.

1157

Fig. 6. Photomicrographs showing mineralogical and structural features of least
altered and altered host rocks in the proximal zone. Photographs a to d, f: + Pol; e:
plane-polarized light. See text for discussion. Mineral abbreviations are after Kretz
(1983). Car: carbonates.

1162

1163 Fig. 7. Photomicrographs showing the mineralogical and structural features of the auriferous laminated quartz veins in the inner zone. (a) Alternating laminae consisting 1164 of quartz + dolomite and muscovite + chlorite. (b) S-C fabric within the lamina 1165 showing sinistral sense of shear. (c) Plastic deformation structures in quartz grains. (d) 1166 1167 Quartz shear lens within the mylonites. (e) Fragments of quartz vein with brittle fractures and mylonite clasts in fault breccia. (f) Brittle fractures in the arsenopyrite 1168 1169 and pyrite grains are filled with quartz and chlorite; quartz strain fringes with chlorite. 1170 (g) Silicate inclusion trails within sulfides showing relationship between sulfide growth and shear foliation. (h) Sketch of (g) for clear illustration. All photographs are 1171 1172 taken under + Pol, except (e and g). See text for discussion. Mineral abbreviations are 1173 after Kretz (1983).

1174

Fig. 8. Reflected light photomicrographs and BSE images showing different types of 1175 1176 pyrites. (a) Three textural types of pyrite. (b) Coexisting pyrite-I and arsenopyrite-I along with pyrite-II inclusions within arsenopyrite-I. (c) Gold (Au) and chalcopyrite 1177 1178 inclusions within pyrite-I with arsenopyrite-II at pyrite-I boundary. (d) Chalcopyrite, 1179 pyrrhotite and arsenopyrite-II inclusions within pyrite-II. (e) Euhedral pyrite-I grains with silicate inclusions and As-poor zone (dotted line). (f) Pyrite-II overgrows on 1180 pyrite-I (dotted line). (g) Concentric As-poor and As-rich zones in pyrite-II. (h) 1181 1182 Irregular As zoning in pyrite-II. All photographs were taken in plane-polarized light. 1183 Mineral abbreviations are after Kretz (1983).

1184

Fig. 9. Summary of different textural types of pyrite and arsenopyrite, in the inner and
proximal alteration zone, with schematic sketches. Mineral abbreviations are after
Kretz (1983)

1188

Fig. 10. Reflected light photomicrographs in plane-polarized light showing occurrence of gold in the Klipwal deposit. (a) Gold (Au) inclusions within pyrite-I at the contact of chalcopyrite grain. (b) Gold as isolated inclusion within pyrite-I. (c) Inclusion of gold grain within arsenopyrite-II. (d) Free gold grain in quartz matrix, sharing the grain boundary with chlorite (Chl). Mineral abbreviations are after Kretz (1983).

Fig. 11. Representative photomicrographs and Raman spectra of different fluid inclusion types. (a) Isolated type-I inclusion. (b) Occurrence of type-II and type-III inclusions in the same 3-dimenstioanl cluster. (c) Representative Raman spectrum of type-II inclusions showing CH_4 peak at 2912 cm⁻¹. (d) Raman spectrum of type-III inclusions showing CO_2 doublets (at 1283 and 1388 cm⁻¹) and presence of CH_4 (at 2913 cm⁻¹). (e) Intra-granular trail of type-IV inclusions. (f) Coexisting type-III (C) and type-IV (A) inclusions in one group of synchronous inclusions (GSI).

1202

Fig. 12. Histogram plots of fluid inclusion microthermometric data. (a) Temperatures of CH_4 vapor homogenization of type-II inclusions, (b) Temperatures CO_2 ice melting and (c) Homogenization of type-III inclusions, (d) Last ice melting and (e) Liquid vapor homogenization temperatures of type-IV inclusions, (f) $T_{h,CO2}$ versus $T_{m,CO2}$ plot for pure CO_2 type-III inclusions showing density variation.

1208

Fig. 13. Isochore plot constructed using the minimum and maximum densities of each type of inclusions. Figures indicate the density values (in g/cm³) for respective inclusions. P-T estimation by (i) intersecting isochores of type-III (solid lines) and type-IV (dashed lines) inclusions (ii) intersection of chlorite thermometer (CT) and (iii) arsenopyrite thermometer (AT) with the isochores. The box illustrates the P-T domain of gold mineralization at Klipwal. IS-1 and IS-2 denotes isochore intersection.

1215

1216 Fig. 14. ¹⁸⁷Re/¹⁸⁸Os vs. ¹⁸⁷Os/¹⁸⁸Os isochron plot for pyrite and arsenopyrite grains 1217 from the ore zone. See text for discussion.





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Textural type	Description	Schematic sketch of texture
Arsenopyrite-I	Large euhedral grains with abundant randomly oriented silicate inclusions and may contain inclusions of pyrite-II	Apy-I Silicates Py-II
Pyrite-I	Euhedral in shape with numerous randomly to preferentially oriented silicates inclusions and may contain inclusions of sulfides and gold	Silicates
Pyrite-II	Euhedral pyrite, almost devoid of silicate inclusions and generally overgrow pyrite-I	Silicates
Arsenopyrite-II	Euhedral to subhedral in shape, smaller in size than arsenopyrite-I, free of any silicate inclusions and may overgrow pyrite-I	Apy-II Au Au (or) Py-I Apy-II
Pyrite-III	Irregular grain boundaries and overgrows the shear foliation.	Py-III Py-II Py-I







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Table

							R-reef						
Sample No							KU4B						
Analysis No	mat-1	mat-2	mat-4	mat-5	py-6	py-7	py-8	py-9	py-10	mat-11	mat-12	mat-13	mat-16
SiO ₂	25.62	25.25	25.38	25.51	25.08	26.04	25.21	25.38	25.52	25.78	25.48	26.65	26.82
Al ₂ O ₃	20.88	20.49	20.89	20.17	20.59	20.56	20.48	20.55	20.42	20.75	20.64	18.96	19.42
FeO	27.85	29.21	27.81	27.12	27.74	28.80	28.33	28.97	27.34	28.11	28.10	28.20	28.53
MnO	0.10	0.10	0.11	0.20	0.14	0.00	0.17	0.17	0.13	0.15	0.05	0.11	0.07
MgO	13.63	12.92	13.02	13.47	12.70	13.39	13.17	13.28	13.65	13.65	13.29	14.13	14.04
Total	88.08	87.97	87.21	86.47	86.25	88.79	87.36	88.35	87.06	88.44	87.56	88.05	88.88
Cations													
Si	2.70	2.67	2.72	2.69	2.67	2.69	2.67	2.67	2.69	2.70	2.67	2.73	2.74
Al	2.60	2.55	2.64	2.51	2.59	2.50	2.56	2.54	2.53	2.56	2.55	2.29	2.34
Fe(tot.)	2.46	2.58	2.49	2.39	2.47	2.48	2.51	2.54	2.41	2.46	2.47	2.42	2.43
Mn	0.01	0.01	0.01	0.02	0.01	0.00	0.02	0.02	0.01	0.01	0.00	0.01	0.01
Mg	2.14	2.03	2.08	2.12	2.02	2.06	2.08	2.08	2.14	2.13	2.08	2.16	2.14
Total	9.91	9.84	9.93	9.73	9.76	9.73	9.83	9.85	9.78	9.87	9.78	9.61	9.65
T1 (tetrahedral													
Al: CN85)	293	300	290	295	300	297	300	301	296	293	299	287	286
12 (octahedral	270	263	273	250	253	249	262	263	256	266	256	236	240
T3 (tetrahedral	270	203	215	250	233	24)	202	205	250	200	250	230	240
Al & Fe/(Fe+Mg): ZF95)	276	269	276	279	276	285	297	272	287	279	282	278	300
Avg. T °C	280	278	280	275	276	277	286	279	280	280	279	267	276
SD	12	20	9	23	23	25	21	20	21	14	22	27	31
Eq. (1)	286	300	287	304	308	312	300	309	300	287	309	300	278
Eq. (2)	317	303	328	292	300	299	306	312	299	308	301	279	269
Eq. (3)	300	300	305	299	306	306	302	310	300	297	306	291	276
Avg.T °C (T4) V01&05	301	301	307	298	305	306	302	310	300	297	305	290	274
SD	16	2	20	6	4	7	3	1	1	11	4	10	5
a(H ₂ O)	0.3	0.5	0.2	0.9	0.8	1.0	0.5	0.5	0.8	0.5	0.8	0.6	1.0

	R-reef														
Sample No								KU4G							
Analysis No	voin 1	vision 2	vain 2	voin 4	vain 5	voin 6	voin 7	vain 9	vain 0	vain 10	mot 11	mot 12	mot 12	mat-	mat-
Affalysis No	24.25	vein-2	vein-5	vein-4	vein-5	24.19	vein-/	Vein-8	vein-9	24.50	mat-11	mat-12	mat-15	14	15
	24.25	24.23	24.33	24.16	24.23	24.18	24.27	24.49	24.70	24.50	24.37	24.85	24.60	24.38	22.93
Al ₂ O ₃	18.46	18.81	18.82	18.35	18.8/	18.72	18.82	18.61	18.51	18.37	18.96	19.33	18.38	18.93	18.03
FeO	36.32	36.69	36.91	36.25	36.49	36.50	36.50	36.32	36.45	36.62	36.68	36.78	37.15	36.44	35.33
MnO	0.13	0.04	0.01	0.08	0.09	0.08	0.11	0.12	0.09	0.19	0.06	0.16	0.05	0.00	0.33
MgO	7.78	7.62	7.56	7.79	7.74	7.67	7.68	7.97	8.06	7.90	7.75	8.26	7.59	7.90	7.36
Total	86.94	87.39	87.63	86.63	87.42	87.15	87.38	87.51	87.81	87.58	87.82	89.38	87.77	87.65	83.98
Cations															
Si	2.67	2.67	2.67	2.68	2.66	2.67	2.67	2.67	2.68	2.68	2.67	2.66	2.68	2.67	2.67
Al	2.39	2.44	2.44	2.40	2.45	2.44	2.44	2.39	2.37	2.37	2.45	2.44	2.36	2.45	2.47
Fe(tot.)	3.34	3.38	3.39	3.37	3.36	3.37	3.36	3.32	3.31	3.35	3.36	3.30	3.39	3.34	3.44
Mn	0.01	0.00	0.00	0.01	0.01	0.01	0.01	0.01	0.01	0.02	0.01	0.01	0.00	0.00	0.03
Mg	1.28	1.25	1.24	1.29	1.27	1.26	1.26	1.30	1.31	1.29	1.27	1.32	1.23	1.29	1.28
Total	9.69	9.74	9.74	9.75	9.74	9.75	9.74	9.69	9.68	9.69	9.75	9.74	9.68	9.75	9.89
T1 (tetrahedral															
Al: CN85)	301	300	299	297	301	300	300	299	297	299	300	301	297	300	300
T2 (octahedral	215	251	251	2.52	251	2.52	0.51	244	244	2.15	2.52	251	244		a co
vacancy: CN85)	245	251	251	252	251	252	251	246	244	245	252	251	244	252	269
Al& Fe/(Fe+Mg): ZF95)	302	303	304	305	306	307	308	309	310	311	312	313	314	315	316
Avg. T °C	283	285	285	285	286	286	286	285	284	285	288	288	285	289	295
SD	33	29	30	29	30	30	31	34	35	35	32	33	37	33	24
Eq. (1)	314	314	306	301	320	317	314	321	311	316	308	312	310	311	286
Eq. (2)	286	298	292	292	302	300	299	292	290	292	292	298	288	296	301
Eq. (3)	301	306	300	298	313	311	307	308	301	306	301	306	300	305	292
Avg.T °C (T4) V01&05	301	306	299	297	312	309	307	307	300	305	300	305	299	304	293
SD	14	8	7	5	9	9	8	14	10	12	8	7	11	8	8
a(H ₂ O)	0.8	1.0	1.0	1.0	1.0	1.0	1.0	0.9	1.0	0.9	0.9	0.9	1.0	1.0	0.3

			J reef		
Sample No			KU12B		
Analysis No	1	2	3	5	7
SiO ₂	25.49	23.94	25.64	24.69	24.89
Al ₂ O ₃	18.44	21.41	18.52	20.78	19.80
FeO	33.87	33.70	33.09	34.06	34.24
MnO	0.00	0.02	0.13	0.00	0.00
MgO	9.40	7.97	9.65	7.75	8.90
Total	87.20	87.04	87.03	87.28	87.83
Cations					
Si	2.74	2.66	2.84	2.68	2.68
Al	2.33	2.80	2.41	2.65	2.52
Fe(tot.)	3.04	3.13	3.06	3.09	3.09
Mn	0.00	0.00	0.01	0.00	0.00
Mg	1.50	1.32	1.59	1.25	1.43
Total	9.62	9.92	9.92	9.67	9.72
T1 (tetrahedral A1: CN85)	286	302	265	299	297
T2 (octahedral	200	302	203	277	271
vacancy: CN85)	236	272	271	242	248
T3 (tetrahedral	210	210	220	221	222
Alæ Fe/(Fe+Mg): ZF95)	318	319	320	321	322
Avg. 1 C	280	298	280	288	289
SD E (1)	41	24	30	41	38
Eq. (1)	279	277	247	299	291
Eq. (2)	264	296	323	274	279
Eq. (3)	273	284	277	288	285
Avg.T °C (T4) V01&05	272	286	282	287	285
SD	7	9	38	13	6
a(H ₂ O)	1.0	0.2	1.0	1.0	1.0

	J reef												
Sample No						KU.	[1						
Analysis No	1	2	3	4	5	6	7	8	9	10	11	12	
SiO ₂	26.03	25.86	25.94	25.55	26.59	26.39	26.30	25.81	26.01	25.94	25.77	25.99	
Al ₂ O ₃	20.51	20.71	20.65	20.60	20.35	20.01	20.03	20.77	21.40	20.15	21.18	21.03	
FeO	24.81	25.42	26.21	26.40	26.07	25.89	26.43	25.18	25.88	25.99	26.07	25.26	
MnO	0.00	0.05	0.05	0.05	0.00	0.06	0.11	0.00	0.00	0.16	0.04	0.00	
MgO	15.11	15.06	15.10	14.61	15.31	15.38	15.17	15.22	15.11	15.26	15.06	15.20	
Total	86.46	87.10	87.95	87.21	88.32	87.73	88.04	86.98	88.40	87.50	88.12	87.48	
Cations													
Si	2.70	2.68	2.72	2.68	2.71	2.70	2.70	2.71	2.70	2.72	2.69	2.69	
Al	2.51	2.53	2.55	2.55	2.45	2.41	2.42	2.57	2.62	2.49	2.61	2.57	
Fe(tot.)	2.15	2.21	2.30	2.32	2.22	2.22	2.27	2.21	2.25	2.28	2.28	2.19	
Mn	0.00	0.00	0.00	0.00	0.00	0.01	0.01	0.00	0.00	0.01	0.00	0.00	
Mg	2.34	2.33	2.36	2.28	2.33	2.35	2.32	2.38	2.34	2.39	2.35	2.35	
Total	9.70	9.76	9.93	9.83	9.71	9.69	9.72	9.88	9.92	9.90	9.93	9.79	
T1 (tetrahedral Al: CN85)	293	297	290	298	291	293	294	291	293	289	295	295	
T2 (octahedral vacancy: CN85)	247	253	273	262	247	245	249	267	271	270	273	257	
T3 (tetrahedral Al& Fe/(Fe+Mg): ZF95)	324	325	326	327	328	329	330	331	332	333	334	335	
Avg. T °C	288	292	296	296	289	289	291	297	299	297	301	296	
SD	39	36	27	33	41	43	41	33	31	33	31	39	
Eq. (1)	300	278	287	283	269	281	277	282	297	262	299	274	
Eq. (2)	287	276	325	296	268	268	275	307	323	299	327	279	
Eq. (3)	295	277	304	289	268	276	276	292	307	277	311	276	
Avg.T °C (T4) V01&05	294	277	305	289	268	275	276	294	309	279	312	276	
SD	6	1	19	6	1	6	1	13	13	19	14	3	
a(H ₂ O)	1.0	0.6	0.2	0.4	0.8	0.7	0.7	0.5	0.3	0.3	0.3	0.2	

Note: CN85: Cathelineau and Nieva (1985); ZF95: Zang and Fyfe (1995); V01&05: Vidal et al. (2001) and (2005)

Sample No										KU10A									
Analysis No	2	3	4	8	11	12	13	14	16	17	18	22	23	25	26	27	28	30	31
Fe	36.12	36.26	36.47	36.63	36.65	36.77	36.37	36.26	36.35	36.47	36.82	36.55	36.55	36.72	36.14	36.67	36.82	36.33	36.57
As	41.46	40.85	41.06	41.08	41.55	41.20	41.77	40.85	42.10	42.09	41.05	41.28	41.75	41.61	42.45	41.29	41.10	41.16	40.99
S	22.72	22.68	22.90	22.80	22.52	22.59	22.40	22.90	22.34	22.22	22.74	22.55	22.39	22.45	21.97	22.86	22.95	22.75	22.84
Со	0.04	0.00	0.00	0.04	0.02	0.02	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.01	0.00	0.00	0.00	0.01
Ni	0.11	0.00	0.00	0.00	0.11	0.00	0.01	0.00	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
Cu	0.00	0.00	0.00	0.00	0.00	0.00	0.02	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.00
Zn	0.00	0.00	0.00	0.00	0.00	0.00	0.03	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00
Ga	0.00	0.00	0.01	0.00	0.03	0.01	0.00	0.01	0.03	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
Se	0.16	0.18	0.10	0.18	0.18	0.15	0.12	0.17	0.18	0.13	0.13	0.14	0.14	0.15	0.13	0.22	0.15	0.17	0.15
Ag	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.01	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.00
Au	0.00	0.00	0.00	0.07	0.00	0.00	0.00	0.00	0.01	0.00	0.01	0.05	0.00	0.00	0.00	0.00	0.00	0.00	0.04
Total	100.61	99.98	100.54	100.78	101.05	100.73	100.73	100.19	101.03	100.92	100.76	100.58	100.83	100.92	100.69	101.05	101.03	100.42	100.61
As %	28.9	28.6	28.6	28.6	28.9	28.7	29.2	28.5	29.4	29.4	28.6	28.8	29.1	29.0	29.8	28.6	28.5	28.7	28.5
T (°C)	277	265	265	265	277	267	291	260	300	300	265	272	286	282	318	265	260	267	260

TableTable. 2. Selected electron probe microanalytical data of arsenopyrite from alteration zones along with results of arsenopyrite thermometry after Kretschmar and Scot (1976) and Sharp et al. (1985).

Sample No						KU16					
Analysis No	3	4	5	6	8	10	11	13	16	17	19
Fe	36.02	35.91	36.23	35.86	36.35	36.31	36.34	35.54	36.24	36.14	36.43
As	41.40	41.11	41.10	41.20	41.19	40.85	40.89	40.74	41.54	40.91	40.85
S	22.29	22.57	22.59	22.51	22.54	22.75	22.54	22.71	22.13	22.69	22.64
Со	0.00	0.02	0.00	0.01	0.00	0.01	0.01	0.01	0.00	0.00	0.00
Ni	0.00	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
Cu	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
Zn	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
Ga	0.00	0.00	0.00	0.00	0.00	0.00	0.02	0.00	0.00	0.02	0.04
Se	0.13	0.15	0.13	0.17	0.15	0.17	0.13	0.09	0.17	0.16	0.16
Ag	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
Au	0.00	0.00	0.04	0.00	0.00	0.03	0.00	0.00	0.00	0.04	0.03
Total	99.84	99.77	100.09	99.75	100.23	100.12	99.93	99.08	100.08	99.97	100.14
As %	29.0	28.8	28.7	28.8	28.7	28.4	28.5	28.6	29.1	28.5	28.4
T (°C)	282	272	262	272	262	255	260	265	286	260	255

Sample No (Reef)	Туре	n	$T_{m,CO2}(^{\circ}C)$	$T_{h,CO2}(^{\circ}C)$	$T_{h,CH4}(^{\circ}C)$	$T_{m,Cl}(^{\circ}C)$	$T_{m,ice}(^{\circ}C)$	T _h (°C)	$T_{h,tot}(^{\circ}C)$	X _(CO2)	X _(CH4)	V _g (%)	Density (g/cm ³)
Klip-1	Ι	12	-62.5 to -57.8	-12.6 to 7.8		9.2 to 13.5			299 to 367	0.25 to 0.75	0.03 to 0.25	40 to 90	0.356 to 0.805
(R-reef)	II	31			-92.8 to -84.3					0.00	1.00		0.221 to 0.275
	III	33	-62.8 to -56.6	-13.1 to 9.1						0.73 to 1.0	0.27 to 0.0		0.663 to 0.971
	IV	22					-16.1 to -0.2	115 to 292					0.799 to 1.048
KU16	Π	20			-94.3 to -85.2								0.230 to 0.281
(R-reef)	III	25	-63.5 to-56.8	-15.5 to 18.3						0.69 to 1	0 to 0.31		0.690 to 1.010
	IV	19					-4.6 to -1.9	172 to 299					0.794 to 0.947
KU10A	Ι	9	-63.2 to -58.2	-15.2 to 0.5		7.8 to 14.8			272 to 345	0.71 to 0.95	0.05 to 0.29	70 to 90	0.352 to 0.713
(J-reef)	II	12			-91.9 to -84.8								0.226 to 0.271
	III	17	-63.9 to -60.0	-20.1 to 0.2									0.642 to 0.831
	IV	15					-10.3 to -0.3	205 to 302					0.803 to 0.935
KU12D	Π	25			-94.2 to -85.4								0.232 to 0.280
(J-reef)	III	19	-60.2 to -56.7	-13.5 to 13.4						0.86 to 1.0	0.0 to 0.14		0.437 to 1.001
	IV	15					-12.2 to -2.0	157 to 222					0.901 to 0.982
KU9C	II	15			-92.5 to -84.9								0.227 to 0.273
(H-reet)	III	20	-63.0 to -57.2	-14.8 to 13.0						0.72 to 0.99	0.01 to 0.29		0.645 to 0.972
	IV	17					-12.5 to -0.5	130 to 250					0.831 to 1.022

Sample No	Location	Mineral	Re (ppb)	±	Os (ppt)	±	¹⁹² Os (ppt)	±	¹⁸⁷ Re/ ¹⁸⁸ Os	±	¹⁸⁷ Os/ ¹⁸⁸ Os	±	rho
RO432-2_KUJI_Apy	J-Reef	Arsenopyrite	0.9	0.1	73.4	2.4	20.3	1.4	84.1	8.1	3.9337	0.3161	0.626
RO432-3_KUJI_Py	J-Reef	Pyrite	0.2	0.1	56.6	1.7	20.7	1.5	18.7	5.7	1.1147	0.0905	0.193
RO458-1_KU11_Py	H-Reef	Pyrite	0.4	0.0	106.6	2.2	38.3	1.6	22.4	1.2	1.2645	0.0725	0.552
RO458-2_KU11_Apy	H-Reef	Arsenopyrite	1.6	0.0	154.1	3.9	44.4	1.8	70.6	3.0	3.4446	0.1967	0.692
RO458-3_KU16_Py	R-Reef	Pyrite	0.5	0.0	32.5	1.0	7.7	0.4	128.1	8.1	5.7947	0.3978	0.715
RO458-4_KU16_Apy	R-Reef	Arsenopyrite	4.7	0.0	240.4	8.0	44.8	1.8	210.4	8.6	9.4437	0.5389	0.708

Note: Uncertainties are reported at the 2σ level. ¹⁸⁷Os/¹⁸⁸Os uncertainties are at 2SE. All data are blank corrected, blanks for Os and Re were 0.1 ± 0.1 and 6.2 ± 5.4 ppt respectively, with an average ¹⁸⁷Os/¹⁸⁸Os value of 0.25 ± 0.02 (1 SD, n = 2).

Age	Gold fields/ deposits	Cratons/regions	Fluid composition	Salinity (wt.% NaCl equiv.)	Mineralizing P-T conditions	References
Early Paleocene – Early Eocene	Alaska–Juneau, Treadwell Kensington	Juneau gold belt, SE Alaska	H ₂ O-CO ₂ -	NA	250 to 350°C and 0.75 to 3 kbar	Goldfarb et al. (1993)
Siluro- Devonian	Hodgkinson gold field	NE Austraila,	CH ₄ -NaCl	3 to 11	270 to 355°C and ~1 kbar	Peters et al. (1990); Bierlein and Crowe (2000)
Cambro- Orodovician	Lachlan gold field	SE Australia		NA	250 to 350°C	Gao and Kwak (1995a, b); Ramsay et al. (1998)
	Willuna gold camp		H ₂ O-CO ₂ -	2.9±2.1	300±30°C and 0.55 to 1.4 kbar	Hagemann et al. (1994, 1996)
-	Kalgoorlie camp, Golden Mile	Yilgarn Craton, western Australia	with varying	NA	264 to 360°C and 1.5 to 2.3 kbar.	Ho (1987); Ho et al. (1990); Hagemann and Cassidy (2000)
	Mount Charlotte deposit		ratio	≤5		Hagemann and Cassidy (2000); Mernagh (1996)
	Hollinger- McIntyre deposit	Abitibi sub	H ₂ O-CO ₂ - CH ₄ -NaCl	NA	277±48°C	Smith et al. (1984); Spooner et al. (1987)
	Sigma-Lamaque, Val d'Or	province at Timmins district, Canada	$\begin{array}{c} H_2O\text{-}CO_2\text{-}\\ CH_4\text{-}NaCl\\ with 15-30\\ mole \%\\ CO_2 \end{array}$	<10	1.8 to 2.6 kbar	Hagemann and Brown (1996); Robert and Kelly (1987)
Late – Middle	Barberton greenstone belt	Kappval Craton, South Africa	H ₂ O-CO ₂ - CH ₄ -NaCl	NA	~300°C and 1 kbar.	de Ronde et al. (1992)
Archaean	Kolar gold field		H ₂ O-CO ₂ - NaCl	~7	205 to 280°C and 0.7 to 1.8 kbar	Mishra and Panigraghi (1999)
	Hutti gold mine, Hutti-Muski greenstone belt		H ₂ O-CO ₂ -	3.9–13.5	280–320°C and 1.0 to 1.7 kbar	Pal and Mishra (2002); Mishra and Pal (2008)
	Hira-Buddini mine, Hutti- Muski greenstone belt	Eastern Dharwar Craton, India	CH ₄ -NaCl	0.5 to 22.7	550°C ¹ 320°C ²	¹ Krienitz et al. (2008) ² Mishra and Pal (2008)
	Ramagiri gold field		Dominantly carbonic	low-salinity	1.45 kbar/240°C to 1.7 kbar/267°C;	Sinha (1997)

fluids

H₂O-CO₂-

CH₄-NaCl

~5

Saravanan et al. (2009);

Chinnasamy and Mishra

(2013)

263 - 323°C and

1.4 to 2.5 kbar

Table. 5. Summery of ore fluid compositions and P-T conditions of gold mineralization in major orogenic gold deposits around the world.

Note: NA= Not available

Jonnagiri

Deposit,

Jonnagiri

greenstone belt