THE GEOLOGY, PETROLOGY AND GEOCHEMISTRY OF THE MINERALIZATION AND HYDROTHERMAL ALTERATION AT ONGEAMA, ONGOMBO AND MATCHLESS WEST EXTENSION, NAMIBIA

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Declaration

All the work in this thesis is that of the writer, except where specific reference is made to the work of others.

Signed: Morilene Moroni

The Matchless Amphibolite Belt (Damara Orogen, Namibia) hosts several volcanogenic-exhalative, sediment-hosted stratiform cupriferous pyrite deposits. These are thought to be related to submarine volcanism during the early evolutionary stages of a narrow Damaran ocean, the Matchless Trough. The mineralized bodies examined (Ongeama, Ongombo and Matchless West Extension) are deformed and metamorphosed to lowmedium grade (greenschist-amphibolite facies). They are associated with metapelite and amphibolite country rocks, and crop out as prominent limonite-rich gossans. The elongated shape of the sulphide bodies suggests a structural control. The mineralization normally consists of a variably developed massive sulphide portion, either talc- or amphibole-bearing, and a stratigraphically quartz-, overlying, extensive horizon of sulphide- and baryte-bearing exhalite (magnetite quartzite and less common talc- and actinolite-bearing schists). Lateral and vertical mineralogical changes within the mineralization match with significant variations in the element distribution. A metamorphosed and deformed alteration pipe, indicating the position of the fluid conduit, can be recognized in association with some ore bodies. The formation of quartz-muscovite and chlorite alteration envelopes (Ongeama, Matchless West Extension) and the presence of subtle mineralogical changes (Ongombo) in the immediate wallrocks, accompanied by extensive redistribution, leaching and introduction of elements from outside, suggest the hydrothermal metasomatic origin of the alteration zones. Element zoning within the mineralized bodies can be related to the original position of the vent, possibly coinciding with the intersection of the axis of the alteration pipe with the sulphide body. Cu, Zn, Au (pro parte) and Mo are enriched proximal to the vent, whereas Pb, Ba, Mn, Ag, Au, Sn, Bi enrichment characterizes the distal facies of the and mineralization. In spite of the obliterating and disrupting effects of the regional dynamo-metamorphism, the element distribution within the mineralization and alteration zones examined is comparable with the geochemical trends observed in present-day mineralizing systems in Basin). early-stage oceanic environments (e.g. Guaymas During exploration for blind volcanogenic mineralization, the detection of hydrothermally altered rocks is fundamental in indicating the proximity to the mineralization. The localization of the alteration zone is also important in the interpretation of the regional geology of the explored area: in deformed terrains the assessment of the stratigraphic position of the alteration zone, relative to the mineralization, helps in establishing the polarity of the sequence.

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INTRODUCTION

The purpose of this study is to examine the geological, mineralogical and geochemical characteristics of metamorphosed and deformed sediment-hosted volcanogenic exhalative mineralization and associated alteration. The mineralization is represented by three Besshi-type sulphide deposits, Ongeama, Ongombo and Matchless West Extension, in the eastern sector of the Matchless Amphibolite Belt, Damara Orogen, Namibia.

Information was obtained from the regional geological framework of the Matchless Belt (Part A), detailed mapping, core logging and sampling of mineralization, alteration and country rocks (Part B), microscope examination of thin and polished sections (Part C), and geochemical analyses (Part D). This information was used with the following purposes:

a) gain a better understanding of the characteristics and nature of the mineralization hosted by the Matchless Belt;

 b) evaluate the effects of regional metamorphism and deformation on the mineralogy and geochemistry of mineralization and associated hydrothermal alteration;

c) obtain some guidelines and principles applicable to exploration for similar mineralization in a metamorphosed terrane.

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PART A

REGIONAL GEOLOGICAL FRAMEWORK

1 GEOLOGICAL AND STRUCTURAL EVOLUTION OF THE DAMARA OROGEN

The Damara Orogen represents a part of the Pan African orogenic network dissecting Africa (Fig. 1a). The Damara Orogen (Fig. 1b) consists of a NE-trending 400 km-wide intracontinental branch, which occupies most of Central and Northern Namibia, and of a NNW-SSEtrending, 150 km-wide coastal branch, whose southern extension is represented by the Gariep Belt. The intracontinental and coastal branches define a three-armed orogenic junction. This triple junction separates the overriding Congo Craton from the Kalahari Craton, and, along its intracontinental branch, it probably is connected to the Pan African Katanga Belt in Zambia across Botswana (Martin and Porada, 1977).

A brief account of the geology and the structure of the Damara Orogen, with special reference to the intracontinental branch and, within this, to the Southern Zone, is given below.

The main geological and tectonic features of the Damara Orogen were developed during middle-late Proterozoic to early Palaeozoic times. The nature and distribution of lithologies in the Damara sequence indicate their deposition in an extensional environment. The tensional regime induced the opening of the proto-South Atlantic Ocean along the coastal branch, whereas the intracontinental arm of the triple junction evolved in an aborted rift (Tankard et al., 1982).

The complex evolution of the intracontinental branch may have involved continental rifting followed by convergence and continental collision (Miller, 1983a). During Karoo-post Karoo times, new extensional tectonism developed coincident with the Gondwana breakup and resulted in the separation of Africa from South America.

According to Mason (1981), the early tectonic history of the Damara Orogen followed trends established by an earlier, Irumide rift system, the Koras-Sinclair-Ghanzi (KSG) Rift (after Borg, 1988) (Fig. 2a). Both KSG and Damaran rift systems might represent the results of a continuous tectonic process related to southward plate migration over



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Fig. 1 - a) Pan African belts and areas of thermal and tectono-thermal rejuvenation during the Pan African event (Porada, 1983). b) Schematic tectonic map of the Damara Orogen, Namibia, showing tectonic zones (modified after Tankard et al., 1982).

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Fig. 3 - Supposed position of rift systems during the early geosynclinal development of the Damara Orogen (modified after Porada, 1983).

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a stationary mantle plume (Fig. 2b).

The intracontinental belt is commonly divided into zones (i.e. Northern, Central, Southern and Southern Marginal zones) on the basis of geological, structural and metamorphic features. A simplified stratigraphy of the volcano-sedimentary Damara sequence across the belt is given in Table I. Briefly, an early rifting phase, represented by the volcano-sedimentary Nosib Group, is followed by the development of a carbonate shelf facies (Otavi Group) in the north, and a "eugeosynclinal" facies (Swakop Group) in the central and southern parts of the belt. Rifting is believed to have occurred in three main, approximately parallel NE-trending graben basins (B, C and D in Fig. 3) (Miller, 1983a; Porada, 1985). A shift of the rifting activity from north toward south is envisaged by Porada (1985). Of the three basins, only the southernmost, the Khomas Trough in the Southern Zone, experienced the most intense rifting process. In the Southern and Southern Marginal Zones several amphibolite units with transitionalwithin-plate to mid-ocean ridge basalt, or MORB, affinities reflect this process (Finnemore, 1978; Miller, 1983b). The latter amphibolite units are represented by the narrow and linear NE-striking Matchless Amphibolite Belt (see below). Important flysch sedimentation in the northern and southern rift basins is represented by the Kuiseb Formation, attaining a thickness of up to 10000m in the Khomas Trough. This formation includes pelagic sediments and interbedded continentally-derived terrigenous and carbonate-rich (marly) material deposited from turbidity currents (Miller et al., 1983). The carbonate component of the Kuiseb Formation is believed to have been partly derived from deep-sea debris fans (Tinkas turbidites) originating from the carbonate platform (Karibib Formation; see Table I) along the northern margin of the Khomas Trough (Fig. 4). On the basis of geophysical investigations, the Khomas Trough does not persist into the sand-covered extension of the Damara Belt into Botswana (Pirajno and Jacob, 1984).

Convergence and continental collision are responsible for polyphase deformation, metamorphism and, in the Central Zone, emplacement of



Table I - Simplified lithostratigraphy of the Damara sequence (Porada, 1983).

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Fig. 4 - a) Palaeo-geographic map of the Damara geosyncline showing the Karibib platform and the Khomas Trough at the time of deposition of the Tinkas turbidite sequence; b) hypothetical NW-SE section from the Karibib platform to the Khomas trough, showing the deposition of the Tinkas turbidites (Porada, 1985).

Fig. 5 - Reaction isograds in pelitic rocks in the Damara Orogen (Hoernes and Hoffer, 1979). Abbreviations: *chl*, chlorite; *ms*, muscovite; *qtz*, quartz; *an(pl)*, anorthite component of plagioclase; *pl*, plagioclase; *gr*, garnet; *ky*, kyanite; *ad*, andalusite; *si*, sillimanite; *cd*, cordierite; *kf*, K-feldspar; *L*, liquid; *v*, vapor. voluminous, almost exclusively S-type, collision-related granitoid intrusives (Pitcher, 1982; Miller, 1983a).

In general, two main deformation phases, D1 and D2, and a, less intense D₃ phase are responsible for the structural grain of the belt. Each deformation phase was accompanied by emplacement of granitic Different deformation styles characterize the central and rocks. southern parts of the belt. The Okahandja Lineament is an important tectonic boundary between the Central and Southern Zone (Downing and Coward, 1981). In contrast with the dome-and basin tectonic style induced by granitic intrusions in the Central Zone, structures in the Southern Zone are linear: they are upright close to the Okahandja Lineament, become less steep and southeast-vergent to the south, and culminate in intense southeastward thrusting of cover and pre-Damaran basement in the Southern Marginal Zone. Tight to isoclinal folding and intense axial-plane transposition characterize the tectonic style of the Southern Zone. Depleted-mantle, Alpine-type serpentinite bodies (Barnes, 1983) were emplaced along thrust planes into both the pre-Damaran basement inliers and the Damara sequence of the Southern Zone.

The distribution of the metamorphic reaction-isograds within the belt (Fig. 5) (Miller, 1981) shows progressive increase of temperature from the margins towards the centre of the belt, and the development of anatexis in the Swakopmund area (Central Zone). This may represent the original location of the triple point. High pressure-low temperature conditions are recorded in the Southern and Southern Marginal Zones (Kasch, 1983; Miller, 1983a). Metamorphic assemblages in the Central and Southern Zone indicate two main thermal peaks, which were contemporaneous and posterior to the main deformation phase (D_2) .

Post-Pan African extensional tectonics developed mostly by reactivation of Damaran trends. Faulting occurred along main NEtrending and along NNW- and NW-trending lineaments, the latter parallel to the opening of the South Atlantic Ocean. Bimodal fissure volcanism and intrusion of dolerite dike swarms and alkalineperalkaline rocks characterize the Karoo-post Karoo intraplate magmatic activity (Eales et al., 1984).

2 THE MATCHLESS AMPHIBOLITE BELT AND ITS SIGNIFICANCE IN THE GEODYNAMIC EVOLUTION OF THE DAMARA OROGEN

The Matchless Amphibolite Belt consists of a NE-trending, approximately 350 km-long and up to 3 km-wide, linear belt of mafic rocks (Matchless Member), included in the sedimentary fill of the Khomas Trough (Fig. 6). This belt shows evidence of folding at its western end, and it disappears beneath the Kalahari sand cover to the east. In the lower half of the Kuiseb Formation, the Matchless Member occurs as numerous layers, lenses and bands of medium- to finegrained amphibolite and chlorite-amphibole schist enveloped in the metasediments. Structures interpreted as deformed pillows in finegrained amphibolite units suggest the presence of original basaltic lava flows. Nevertheless, the formation of pillow structures is not diagnostic only of extrusive basalts, because similar structures can form along the prograding front of basaltic sills intruding wet basinal sediments (Yagi, 1969). Lenticular pods of metagabbro, locally with basal ultramafic differentiates, locally show foliated margins and almost massive cores. These features are suggestive of structural emplacement. Several Alpine-type serpentinite bodies (see above) are associated with the Matchless Member in the eastern sector of the belt.

The chemistry of the Matchless amphibolites suggests a general affinity with MORB-type tholeiites (Finnemore, 1974, 1978; Miller, 1983b; Schmidt and Wedepohl, 1983), although meaningful variations at a regional scale are detectable (Breitkopf and Maiden, 1988; see below).

The evolution of the Damara Orogen is still the object of debate. The interpretation of the features of the Matchless Belt is one of the factors constraining the formulation of models for the geodynamic evolution of the Damara Orogen. Existing geodynamic models are divided into two groups by Martin (1983), with the assuption of a mechanism of



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subduction (either continental or oceanic) as a common denominator (for details see Martin, 1983, and references therein):

 a) <u>continental collision models</u> ("aulacogen" and "delamination" models);

b) <u>oceanic crust subduction models</u> (subduction of a wide or narrow ocean). In this group the very recent models by Hoffmann (1990) and Kukla & Stanistreet(1990) maybeincluded, although these consider the formation of the Kuiseb schists as subduction-related clastic trench sedimentation and the emplacement of the Matchless amphibolites within an accretionary prism during subduction.

The first group of models denies the opening and subsequent closure of an oceanic basin, and, consequently, the origin of the Matchless Belt from oceanic crust. This denial is based on several factors, among which are the lack of a recognizable ophiolite sequence, the small volume of the mafic rocks in the Matchless Belt, the absence of volcanic and/or plutonic rocks typical of magmatic arcs over a subduction zone (I-type; Pitcher, 1982), and the involvement of basement inliers in all the major structural zones of the orogen.

These factors strongly affect the validity of the second group of models, which however take into account the mid-ocean-ridge-type geochemistry of the Matchless amphibolites, the tectonic emplacement of the serpentinites in that part of the orogenic belt, and the existence of major thrust sheets and nappe structures along the southern margin of the belt. According to Martin (1983), the formation of a narrow oceanic basin (1000 km wide; Miller, 1983b) and the involvement of stike-slip shear movements (Downing and Coward, 1981) appear to be interesting alternatives to the wide-ocean subduction model.

Smalley's (1988) and Breitkopf and Maiden's (1988) models for the formation of the Matchless Belt take into account the last two mentioned factors. Crustal thinning is believed to have occurred with local formation of a spreading centre in a subbasin, the sediment-

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Fig. 7 - Plot of TiO₂ versus FeO_{tot}/MgO. MORB=mid-ocean ridge basalt; CFB=continental floor basalt; triangles=Matchless Belt, western part; diamonds=Matchless Belt, central part; filled squares=Matchless Belt, eastern part (Breitkopf and Maiden, 1988).



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Fig. 8 - Map of the Gulf of California, showing the location of the Guaymas Basin (Peter and Scott, 1988).



Fig. 9 - Simplified block diagram and plan view of Guaymas Basin-type pull-apart basins in stage of early drifting with formation of sill-sediment complexes in spreading troughs (Einsele et al., 1980).
filled Matchless Trough (Smalley, 1988). This subbasin, along the axis of the Khomas Trough, would have originated as a pull-apart basin by simple crustal stretching and superimposed strike-slip shear movements (Breitkopf and Maiden, 1988).

A point of note is that geochemical variations along the axis of the Matchless Belt suggest that most crustal thinning , and eventual spreading, may have occurred in the eastern sector, where amphibolites with oceanic-ridge-type tholeiitic geochemical signature are present (Fig. 7). Geochemical data, however, indicate that a tectonic setting similar to that of fully developed ocean spreading centres is improbable. In the west, amphibolites with geochemical affinities to within-plate basalts suggest the original presence of thin continental crust (Breitkopf and Maiden, 1987, 1988).

The development of the Matchless Trough is comparable with that of modern counterparts, such as the Guaymas Basin (Gulf of California) (Fig. 8) (Einsele et al., 1980). The proximity to the continent determines a sedimentation rate approaching or surpassing the spreading rate. A similar situation The seismically activated sedimentary filling of the basin (turbidites) would generally preclude the extrusion of magma at the sediment-water interface and induce the intrusion of basaltic sheets within the soft, wet sediments. A particular stratigraphy of the oceanic crust is obtained with the formation of a transitional zone, i.e., a sill-sediment complex, between the overlying youngest sediments and the sheeted dike complex at depth (Fig. 9) (Einsele et al., 1980).

3 SULPHIDE MINERALIZATION AND GEOLOGY IN THE EASTERN SECTOR OF THE MATCHLESS AMPHIBOLITE BELT

Along its length, the Matchless Amphibolite Belt hosts several volcanogenic-exhalative, stratiform and strata-bound cupriferous sulphide deposits (Fig. 6), which display common features throughout the belt. These deposits range in size from 0.3 to 16 millions tons of ore averaging 2% Cu (Smalley, 1988; Breitkopf and Maiden, 1988). Details on the main deposits of the belt are found in Killick (1983) and Breitkopf and Maiden (1988) (Gorob), Adamson and Teichmann (1986) and Klemd et al. (1987) (Matchless), Goldberg (1976) and Thomson (1989) (Otjihase).

These deposits consist of massive to semi-massive mineralization, in which predominant pyrite and/or pyrrhotite are accompanied by subordinate chalcopyrite. In addition to Cu, the mineralization can carry variable amounts of Zn, Co, Ag, Au and Mo. Layers of exhalative oxide-silicate facies iron formation (Fox, 1984) are associated with the mineralization, which is typically capped by these layers. The sulphide bodies are hosted in the metaturbiditic Kuiseb schists and are closely associated with the amphibolites. Irregular schist bands and lenses mineralogically distinct from the normal Kuiseb schists and commonly spatially related to the mineralization are considered to represent original hydrothermally altered sediments.

During the orogenesis, the sulphide bodies were deformed into isolated or closely stacked pencil- or ruler-shaped shoots, and arranged according to the regional structural grain. The plunge of the ore bodies is variable at a regional scale according to an undulating pattern (Fig. 10).

The Matchless Belt ore deposits are considered to be examples of Besshi-type mineralization according to the definition given by Fox (1984), i.e., subaqueous exhalative mineralization formed in epicratonic rift environments and resulting from hydrothermal convection developing in the subjacent mafic volcanic-clastic sedimentary piles. Besshi-type mineralization occupies an intermediate position in a rift-related metallogenic spectrum, whose end-members

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Fig. 10 - Schematic sketch of the regional undulating plunge of the mineralized bodies along the Matchless Belt (Smalley, 1988).



Fig. 11 - Generalized tectonic map of the northern Pacific plate boundaries and the location of the Escanaba Trough (Kappel and Franklin, 1989).

are sedimentary-exhalative- and Cyprus-type deposits. The rate and degree of crustal extension is the discriminating genetic factor for the types of mineralization between the end members. Apart from the type-locality in Japan (Kanehira and Tatsumi, 1970), other examples of Besshi-type mineralization are those in the Blue Ridge Belt (Virginia, USA) (Gair and Slack, 1984), in the Norwegian Caledonides (Fox, 1984 and references therein), and the modern sulphide mounds in the Guaymas Basin and in the Escanaba Trough (Lonsdale and Becker, 1985; Kappel and Franklin, 1989) (Fig. 11).

The size of the sulphide deposits along the Matchless Belt seems to correlate positively with the assumed occurrence of oceanic crust towards the east. In fact, the eastern sector, or Windhoek sector, of the Matchless Belt (where the study areas are located; Fig. 12) hosts the economically significant mineralization: i.e. the largest deposits, at the Matchless Mine (recently closed), at the Otjihase Mine (currently operating), and other minor deposits currently at the prospect stage.

Accounts of the geological and tectonic features of the eastern sector of the Matchless Belt are given by Viljoen et al. (1975) and Smalley (1988). ERTS imagery of this sector of the belt (see interpretation in Fig. 6) illustrates the superposition on the NEoriented, Damaran regional trends by dominantly NNW-SSE later fault systems. Some of these faults displace the amphibolite belt in the region to the east of Windhoek to produce an "en echelon" pattern (Fig. 12). A number of less evident, older fault zones (wrench or thrust faults) occur, their traces being almost parallel to the regional trends. An example, in Fig. 13, is the wrench fault with sinistral lateral movement along the Otjihase River and extending between the Ongeama and Otjihase gossans (Viljoen et al., 1975).

Two structural features are of particular interest in the eastern sector of the Matchless Belt (Smalley, 1988):

a) the so-called Colvania Horizon, a cataclastic/mylonitic band (fault/shear zone or thrust), acutely convergent with the



Fig. 12 - Simplified regional map showing the distribution of the main geological features of the eastern sector (Windhoek sector) of the Matchless Belt (modif. after Smalley, 1988). Study areas are outlined.

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southernmost, structurally lowest amphibolite horizons. It extends both west and east of Windhoek, over a distance of 50 km (Fig. 11). Towards the east it is characterized by the presence of elongated bodies of magnetic metadiorite ;

b) the linear narrow zone of discontinuous serpentinite bodies, developed east of Windhoek and near the Otjihase Mine, and lying above the structurally uppermost amphibolite band (Fig. 12 and 22).

These features delimit the sequence of amphibolites northwards and southwards, respectively. West of Windhoek , the sequence includes two main bands of metalava, i.e., the structurally lower Matchless Mine Horizon Amphibolite and the overlying Friedenau Amphibolite Horizon (Smalley, 1988). A more complicated stratigraphy characterizes the sector east of Windhoek, in the Otjihase-Ongeama area. In spite of the dominant MORB character of the amphibolites in this sector of the belt (see above), geochemical variations across the amphibolite sequence west of Windhoek reflect an upward transition from within-plate (in the lower Matchless Mine Horizon Amphibolite) to incipient ocean spreading conditions (in the Friedenau Amphibolite) (Fig. 14).

The assessment of these geochemical variations is a significant tool in defining the polarity of the sequence in the deformed Matchless Belt, in which reliable and continuous markers are lacking. On the basis of the amphibolite geochemistry and of characteristics in the local mineralization, Smalley (1988, 1990) claims that the sequence in the western Windhoek sector is the right way up. An opposite interpretation is reported by Klemd et al. (1989). The present author's opinion is given in Part E. In the eastern Windhoek sector, in the Ongeama-Ongombo area, the sequence appears to be overturned (Hoffmann, 1976; Hallworth, 1989).

The Colvania Horizon and the ultramafic band are regarded by Smalley (1988) as the original southern and northern boundary graben faults, respectively, delimiting the Matchless Trough. Magmatic and exhalative activity is believed to have occurred only along the actively dominant side of the graben (i.e. the southern fault in the west, and the northern fault east of Windhoek). The mineralized bodies, in fact,



Fig. 13 - Composite geological and structural map from ERTS imagery of the area around the Otjihase deposit, eastern Windhoek sector, Matchless Belt (scale approx. 1:300000). Blank=Kuiseb Fm.; dot rows=amphibolite bands; horizontal lines=Auas Fm.; crosses=basement granites; 1=Otjihase gossan; 2=Ongeama gossan; 3=Ongombo gossan (Viljoen et al., 1975).

Fig. 14 - Y/Nb ratios for the Matchless Mine Horizon Amphibolite (MMA) and Friedenau Amphibolite (FA). TWPB=tholeiitic within-plate basalt; TOFB=tholeiitic ocean floor basalt; AOFB=alkalic ocean floor basalt (modif. after Smalley, 1988).

Fig. 15 - Schematic sketch with the assumed original location of the sulphide bodies relative to the graben-boundary and transverse (transform ?) faults before (a) and after deformation (b), according to Smalley (1988).





occur next to these assumed major graben faults. Certain overlap and additional magmatic-exhalative activity along transverse faults is thought to have contributed to the stratigraphy in the Otjihase-Ongeama area (Fig. 15) (Smalley, 1988), although in this area Viljoen et al. (1975) envisage a duplication of the amphibolite belt by thrust faulting (see above).

PART B

GEOLOGY OF THE MINERALIZATION AND COUNTRY ROCKS AT ONGEAMA, ONGOMBO AND MATCHLESS WEST EXTENSION

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1. INTRODUCTION

In this chapter an account is given of the geological setting of the sulphide bodies at the Ongeama and Ongombo Prospects, and Matchless West Extension, in the Windhoek sector of the Matchless Belt (for location see Fig. 12). Geological information was gained from previous and current investigations by mining companies, and from field work by the present author, between November 1989 and February 1990.

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The study of the sulphide bodies involved mapping, systematic lithogeochemical sampling of the weathered surface exposures of the mineralization, and detailed examination of drillcores. At Ongombo and Matchless West Extension maps of the outcrops were already available. In these areas, therefore, the task of the author has been the reexamination, as critically as possible, of the previous observations, concentrating on particular features related to the mineralization.

With regard to the sulphide deposits examined, additional areas, including specific facies (magnetite-rich schists, ultramafic schists) were also included for sampling purposes.

Note: in the text the abbreviations "h/w" and "f/w" are used for "hangingwall" and "footwall", respectively.

2. MINERALIZATION AT THE ONGEAMA AND ONGOMBO PROSPECTS

Ongeama and Ongombo are two copper prospects situated on the farms Ongeama and Ongombo Ost/West, about 20 and 30 km NE of Windhoek, respectively. They are currently held under prospecting grant 1623 (Ongeama claims and Hoffnung Grant) by Gold Fields Namibia. The two farms have a gently undulating topography and are traversed by the Otjihase and White Nossob Rivers, respectively. These rivers are dry for most of the year. Outcrops are often concealed by sand cover, soil and vegetation (thorn bushes and grass).

2.1 Brief review of previous and current investigations

The first geological investigations in the Ongeama and Ongombo areas

were carried out by B&O Minerals in the period 1971-1973, when the copper mineralization of the two stratiform sulphide bodies was outlined by drilling. Moderate metal grades, limited distribution of ore-grade mineralization, and subeconomic tonnage did not encourage further investigation.

In February 1974, Johannesburg Consolidated Investments (J.C.I.) undertook airborne magnetic and electro-magnetic surveys, at the scale 1:50000, over the sector of the Matchless Belt between Otijhase and Ongombo. This survey resulted in the identification of a number of well defined magnetic anomalies and conductors. In November 1975, a new prospecting programme was commenced in order to define the controls on the distribution of ore-grade mineralization. This programme included surface mapping, detailed examination of diamond drill cores, soil geochemistry and geophysics, as well as revision of previous data. Additional information on the geological setting and the geometry of the sulphide bodies at depth was obtained by drilling.

Soil geochemistry confirmed the presence of anomalous metal concentrations in the geochemical profile along the strike of the well exposed gossan at Ongeama. At Ongombo erratic exposures of outcrops influenced the quality of response. Nevertheless, an almost continuous profile of metal distribution along strike indicating a minimum strike length with detectable mineralization of about 4700m, was established. Geophysical prospecting in both areas involved a ground magnetic survey and a Turam electromagnetic (EM) survey.

Results of these surveys indicated a strong anomaly over the gossan outcrop at Ongeama, due to the presence of magnetite quartzite and magnetite in the surrounding schists. At Ongombo, erratic anomalies were obtained by ground magnetics, whereas the Turam EM survey yielded more precise responses and indicated a moderate anomaly in the eastern part of the ore body.

Hoffmann's report (1976) comments on the results of the JCI's prospecting. The results for Ongeama were assessed as follows: "limited potential for viable ore ... for any significant distance in depth". At Ongombo, although limited potential was' expected, further drilling was advised by Hoffmann in order to understand the nature of the mineralization and to verify its possible extent in



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Fig. 16 - Ongeama Prospect: schematic sketch-map of the magnetic anomalies (ground magnetic survey, Gold Fields Namibia Ltd., 1989). The location of the gossan outcrops and of the "I anomaly" area (in the frame) is given. - 24 -

depth.

In 1980, the Ongeama and Ongombo claims were taken over by Tsumeb Corporation Limited (TCL). TCL was subsequently incorporated into Gold Fields Namibia Ltd, which is the current holder of the grants.

In order to carry out mapping and geophysical surveys (ground magnetics and electromagnetic powerage, EMP), grids were established in both claims. Magnetic (Fig. 16) and time-domain EM surveys were completed in 1989 at Ongeama, outlining an elongated shoot at depth.

At Ongombo the EM survey outlined three trends or conductors, extending downplunge from the gossan. These correspond to two major conductors, and a secondary, eastward plunging ruler-shaped conductor (ore shoots) (see Fig. 18).

At Ongombo geological mapping was carried out by Schneeweiss (TCL) in 1984 and by McCormick from Natal University in 1988. The Gold Fields crew completed mapping in both claims in 1989. At Ongombo mapping was carried out in conjunction with a new drilling programme.

The geological data obtained by the author were integrated with information from previous work.

2.2 THE ONGEAMA PROSPECT

2.2.1 Geological setting

During this study the weathered outcrops of the sulphide mineralization (gossan) at the Ongeama Prospect were geologically mapped and sampled (see section 5.1 below) at the scale 1:1000 over an area of 1200m x 150m. The limits of this area were chosen according to the existing grid established by Gold Fields Namibia in 1989 and utilized during the mapping project by Hallworth (1989).

Mapping was carried out by the author along 20m-spaced traverses and limited to exposure of the ore zone and the enclosing schists. Details of the geological setting of the Ongeama area are mostly derived from Hallworth (1989) and Hoffmann (1976) and, in part, from the author's observations. A more detailed description of the gossan exposure is given in section 2.2.2 below.

The mineralization is exposed over a strike length of 915 m (Plan 1 in Appendix A). It consists of two discrete sinuous gossanous bands, which are here indicated as "major" and "minor" outcrops. Their strike length is 600m and 315m respectively, and their maximum width is about 20m. An estimate of the maximum thickness of the ore zone is approximately 2m in the centre of the major outcrop (Hoffmann, 1976). The gap offsetting the two gossan outcrops is roughly 200m wide.

The mineralization is conformable with the schists. Accordingly, the strike is from 050° to 078° , with dips varying between 15° and 20° to the north. The plunge of the ore body at $7^{\circ}-8^{\circ}$ is towards WSW.

The gossan is exposed on a shallow slope on which it stands out as a ridge in places. Only gossan and quartz float occur downslope, where outcrops are concealed by sand from a tributary of the Otjihase River. Country rocks surrounding the gossan outcrop are mainly upslope and in the dry river bed.

Hallworth (1989) mapped the geology of the area, including the gossan outcrop and the extension at depth SW of the shoot, as indicated by the magnetic survey.

The country rocks predominantly consist of Kuiseb metapelites and intercalated metapsammitic horizons, with transitional facies between the two rock types. Metapelites are represented by well foliated, planar to contorted, light-grey to greenish, garnetiferous muscovitebearing biotite-chlorite-quartz schist and chlorite-biotite-quartz schist. The schists often contain guartz pods, boudins ("sweat-outs") and veins of metamorphic segregation. The metapsammitic intercalations consist of a fine-grained massive to slightly foliated, light-grey quartz-biotite schists. The metapsammites locally garnetiferous include centimetre- to decimetre-thick layers made up of coarsegrained amphibole-bearing calc-silicate rock. Hallworth's mapping showed the impossibility of mapping the different schist facies as distinct units, because of their lenticular and interdigitating character. Graphitic schists occur as thin and discontinuous pods in the structural h/w of the major gossan outcrop (see Plan 1).

Amphibolites do not occur in the area mapped by the author. They outcrop, however, to the NW and SW, in the h/w of the mineralization. Amphibolites occur as three finely foliated metalava horizons associated with coarse-grained, sheared lenticular metagabbro plugs and metadolerite sills, in Hallworth's area and in its southwestward extension (not indicated in Fig. 17). The attitude of the mafic rocks is conformable with that of the enclosing schists.

The schists enclosing the mineralization display planar to wavy D2related S2-cleavage planes and remnants of the folded D1-related S1 foliation. A D3-related E-W- to NW-SE-trending crenulation cleavage is visible in the foliation planes of schists and metalavas. Late, barren crosscutting quartz veins strike 010° to 082°. The orientation of the veins is conformable with fracture and joint patterns observed in the gossanous schists in the h/w of the major gossan outcrop (see section 2.2.3 and Plan 1). The sinuous, lenticular and strongly elongate shape of the mineralization appears to be the result of stretching and boudinaging induced by the D₂ deformation. According to Hoffmann (1976), thickness variations of the ore shoot at depth are the result of an en-echelon arrangement of several thick lenses down the plunge. A D₂-related lineation, generated by the main deformation phase and typically indicating the plunge of the deformed sulphide body, is observed in the magnetite quartzite outcrop. The trend of this lineation varies from 111° to 099°, and suggests the occurrence of deformation induced by a sinistral shear movement.

There is geological and geophysical evidence for faulting at Ongeama. A general, simplified tectonic sketch from Hallworth (1989) (Fig 17) indicates several major lineaments, with strikes varying from N-S to E-W to ENE-WSW. Hallworth indicates a major 100° -striking dextral shear as the cause of the offset in the gossan outcrops (the large "bull quartz" vein outcrop at the grid reference 10660 E/10252N; Plan 1). In the western portion of the major outcrop, an E-Wstriking dextral shear fault may be the cause of the drag deformation recorded in the gossan (Plan 1). At the eastern end of the minor outcrop a N-S-trending lineament extending through the Otjihase River is possibly responsible for the abrupt interruption of the Fig. 17 - Ongeama Prospect: simplified geological-structural sketch-map with the position of the main air-photo lineaments, fault trends and amphibolite outcrops (hatched). Blank areas indicate sub-outcrops of Kuiseb schists (modified after Hallworth, 1989).



Fig. 17

mineralization towards E.

No evidence of major folding is recorded in the gossan or in the enclosing schists.

The overturning of the stratigraphic sequence at Ongeama was initially inferred from the arrangement of the different facies of the mineralized system (D.H. Corbett, 1989, pers. comm.), assuming that magnetite quartzite is typically located above a sulphide body. The present position of the magnetite quartzite is at the structural f/w of the sulphide body (see sect. 2.2.2).

A detailed description of the Ongeama mineralization revealed by drill cores is reported by Hoffmann (1976). Here a brief account of it is given.

The mineralization extends downplunge for a distance of at least 1650m (vertical depth of 230m) and with a maximum width of 500m. The ore shoot at depth consists of a single, well-defined zone. A small subsidiary mineralized lens occurs above the middle portion of the shoot. The styles of mineralization vary from unevenly disseminated to streaky, sub-massive and massive. Sulphides are represented by pyrite, pyrrhotite, chalcopyrite and sphalerite, with the last two being minor constituents. The best mineralization is hosted in quartz-chloritegarnet-magnetite schists and subordinate magnetite quartzite. The contact with the barren wall rocks is gradational along the margins of the shoot, but abrupt at the h/w and f/w. Pyrite forms a considerable proportion of the coarse-grained, massive to semimassive ore associated with the stratigraphically higher/structurally lower magnetite quartzite. Disseminated and banded chalcopyrite and pyrrhotite largely occur in the structurally higher schists, where pyrite becomes subordinate.

2.2.2 The gossan exposure

The two limonite-rich gossan outcrops (major and minor), represented in Plan 1, pinch and swell along strike and gradually thin towards the ends. The ore zone is represented by quartz-bearing gossan derived from massive sulphide mineralization (hereafter indicated as "massive sulphide gossan"), and magnetite quartzite. A gossan facies derived from amphibole-rich semi-massive mineralization (hereafter indicated as "semi-massive amphibole-rich gossan") is well developed in the centre of the major outcrop. Only in the major gossan outcrop is the ore zone overlain by an irregular band of gossanous magnetite-bearing chlorite and muscovite schists.

Ferruginous schists occur as lenses within the ore zone and in the h/w. The lithofacies of the ore zone are similar in the two separate gossan outcrops, although the gossan in the major outcrop displays a more complex rock association. An account of the distribution of the gossan facies and of their nature is given below.

The <u>minor outcrop</u> consists of a weathered massive sulphide lens in the east, which grades into a poorly exposed magnetite quartzite band towards the west. Non-magnetic ferruginous schists only occur along a shallow trench in the E in the h/w of the massive lens. Deformation is probably responsible for the repetition of the magnetite quartzite band in the f/w.

The <u>major outcrop</u> consists of a well developed pinch-and-swell structured gossan after a massive sulphide layer. This layer is underlain by a magnetite quartzite layer, and overlain by a band of magnetite-bearing schists (Plate 1). The magnetic schist band is best developed in the central part of the outcrop, whereas towards the margins only non-magnetic ferruginous schists are present.

The transition from magnetite quartzite to massive sulphide gossan is either gradational or due to tectonic interleaving. The blocky, banded magnetite quartzite is best developed in the eastern-central sector of the exposure, whereas in the west it crops out very discontinuously . The massive sulphide gossan facies shows considerable thickening at both ends and towards the centre of the sulphide body. A peculiarity of the central portion of the gossan is the lateral transition to a semimassive amphibole-rich quartz-poor gossan facies. This facies occurs in association with the maximum thickness of the magnetite-rich gossanous schists in the h/w of the sulphide body. Late mobilization of the mineralization is manifested by sulphide infilling of crosscutting metamorphogenic quartz veinlets.

The gap between the two gossan outcrops is mostly barren, with minor evidence of mineralization emerging from an isolated trench (grid reference 10720E; Plan 1).

The massive sulphide gossan consists of mainly massive siliceous Fehydroxide-rich bands which alternate with rusty quartzitic and/or thin phyllitic lenses. The limonitic bands may contain massive brownish limonitic jasper, crumbly or cellular siliceous limonite portions, or brown to black concretionary limonite crusts. Fine-grained crystalline quartz, of thin network texture, appears to be a common groundmass for the mineralization. In the cellular portions well preserved round and cubic boxwork structures after pyrite can be observed. Delicate ladder-like boxwork structures after chalcopyrite are less common. Pyrrhotite replicas were recognized only under the microscope (see Section 3.1.1 Part C). Magnetic massive sulphide gossan is rather common in the western central part of the major outcrop, although it may also occur elsewhere.

The <u>semimassive amphibole-rich gossan facies</u> is a slightly banded, relatively porous quartz-poor rock type. It consists of a fine network of yellowish fibrous anthophyllite aggregates (Andrew, 1980, and this study). Ladder-like boxwork structures are locally observed within the interstitial weathered sulphide patches. The transition from siliceous to amphibole-rich gossan is gradational. Isolated small amphibole-rich lenses are also noted in other parts of the massive gossan and magnetite quartzite, in both gossan outcrops. Lenses of ferruginous schist occur associated with the anthophyllite-enriched sector in the major outcrop (grid coordinates 10260E/10197N). This schist facies consists of a magnetic, strongly micaceous, schistose rock type, in which the mica-rich groundmass hosts limonitic boxwork probably pseudomorphing garnet-like porphyroblasts. <u>Magnetite quartzite</u> is a banded, rusty, chlorite-bearing quartzite, containing fine- to coarse-grained magnetite-rich and non-magnetic gossanous bands. Banding, thought to be the result of metamorphism, is mostly planar, although minor slump-like structures can also be observed (Plate 2). Cellular spongy and ladder-like boxworks probably after pyrite and chalcopyrite, respectively, occur in patches not totally invaded by massive or concretionary limonite. Cubic cavities remain after coarse euhedral pyrite grains.

2.2.3 The magnetic hanging wall schists

Detailed mapping carried out by the author outlined the distribution of the magnetite-bearing, gossanous chlorite and muscovite schist band in the structural h/w of the major gossan outcrop (Plates 1 and 3).

The strike of the magnetic schists compares well with that of the country rocks. Nevertheless, the h/w magnetic schists define a transverse, elongated zone of chlorite enrichment, and a continous muscovite-rich band parallel to the strike of the gossan outcrop (Plan 1). The transverse chlorite-rich zone converges with the sulphide body at an angle of about 20°.

Both chlorite and muscovite schists contain abundant coarse disseminated magnetite and are quartz-rich. The deep bluish-green chlorite-rich schists in the axial zone are pockmarked by unusually coarse-grained garnet and honey-brown staurolite porphyroblasts. The latter occur also as centimetre-scale clusters rimming quartz boudins. The magnetic muscovite-quartz schists have a mottled appearance imparted by the numerous and coarse euhedral magnetite crystals. The magnetic schists are moderately gossanous, due to the original presence of small streaks, lenses and disseminations of sulphides. This schist type laterally grades into a fine-grained non-magnetic ferruginous sericite schist, similar to the phyllitic intercalations within the massive gossan. The transition from one rock type to the other can be either abrupt, gradational or characterized by tectonic interleaving.

2.3 THE ONGOMBO PROSPECT

2.3.1 General geological setting

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In the Ongombo Ost/West farm areas sulphide mineralization crops out as a gossanous magnetite quartzite for a total strike length of 4,9 km (Fig 18). The estimated thickness of the mineralization at the surface is in the range of 1 to 2 m.

The mineralization crops out almost continuously from the White Nossob River (Plate 4) towards the SW. East of the river it is found mainly as suboutcrops beneath the sand cover. The gossan outcrop is thickest in the central western portion, and thins towards both ends. It strikes NE-SW (average 039°) with a shallow ($15^{\circ}-20^{\circ}$) NW dip. As at Ongeama, the attitude of the mineralized body is conformable with the regional trend of the enclosing Kuiseb schists. However, the plunge of the ore shoot, as shown by the EMP trends, is towards the NE (Fig. 18). Statistical evaluation of strike readings of the D₂-related lineation in the magnetite quartzite outcrops confirm the EMP trends (Schneeweiss, 1984).

Outcrops in the area around the ore zone (Fig. 18) are poorly exposed. The subdued surface expression of the ore-hosting schists prevents the recognition of any sign of alteration related to the mineralization. Only the more resistant mafic rocks crop out locally in the gently sloping to flat-lying topography of the area.

Previous geological mapping indicated the presence of two main bands of amphibolites positioned above and below the ore body and intercalated within the metasediments. Amphibolites occur as thin, medium-grained, sheet-like bodies in the metasediments, and as coarsegrained boudinaged, lens-shaped metagabbro bodies with sheared margins. Metalava layers do not outcrop. As at Ongeama, the bulk of the stratigraphic sequence is represented by metapelites and metapsammites of the Kuiseb Formation. The Ongombo metapelites are generally garnetiferous and staurolite-rich, unlike the staurolite-



devoid country rocks at Ongeama.

The style of deformation at Ongombo is similar to that observed in the Ongeama area. Minor fold structures are preserved in coarse metagabbro outcrops and occasionally in banded magnetite quartzite. Faulting is revealed by moderate displacement in the gossan outcrop and in the amphibolite zones (Fig. 18). Late veining is widespread throughout the gossan and in the country rocks. Two crosscutting vein sets are recognized by McCormick (1988): WNW-ESE-trending banded tourmalinite veins and ENE-WSW-trending quartz veins. Tourmalinite veins are well developed towards the western end of the gossan outcrop. Tectonic stretching during the main phase of deformation is considered to be responsible for the variations in thickness of the quartzitic sulphide body, which behaved as a rigid and competent unit.

2.3.2 The gossan exposure

The surface exposure of the Ongombo mineralization does not show the complexity shown by the main gossan outcrop at Ongeama.

The Ongombo gossan consists of a moderately gossanous, locally chloritic, banded magnetite quartzite. Subordinate non-magnetic gossanous schists are unevenly intercalated as thin lenses and bands within this magnetic quartzite. In the deeply leached outcrop, copper oxide staining is only confined to the excellent exposure along the White Nossob River and to two other small outcrops (grid reference: 11740E/10225N and 7920E/10000N) (Fig. 18). In general the magnetite quartzite is coarse- to medium-grained, with a sugary texture. Magnetite is clustered in bands or finely disseminated in the granular quartz matrix.

Weathered sulphide mineralization occurs:

a) as disseminations, bands, lenses and stringers along the whole strike extension of the magnetite quartzite;

b) as disseminations in the schists;

c) as impregnations in discordant, fine-grained amphibole-rich patches (possibly anthophyllite);

d) as remobilised, semimassive quartz-vein infillings.

In Fig. 18 three main sectors of the gossan have been highlighted. They correspond to the surface intersection of the two main EMP trends which coincide with the two major downplunging ore shoots. From E to W, the three sectors are the Eastern Shoot, the Central Shoot and the western extension of the Central Shoot. This subdivision of the gossan was devised (see Section 5.1 below) in order to record possible significant changes in the areas approaching the intersection of the shoots with the surface. The evaluation of the lithologic changes along the strike of the mineralized outcrop is dependent on the conditions of the exposure. In general, the approach to the zone of intersection of the EMP trends with the surface is characterized by an increase in the proportion of sulphides within magnetite quartzite and crosscutting quartz veins. This is particularly evident in the better exposed central-western portion of the gossan, towards the Central Shoot. This is less evident towards east due to the poorer outcrop conditions. The two thicker portions of the Central Shoot outcrop coincide with a relative increase in the gossanous sulphide content. An interesting feature in the outcrop of the central-western sector, is the presence of irregular amphibole-rich patches. These patches start to be abundant from within the Central Shoot exposure towards its western extension. The amphibole-rich patches show a fibrousradiating texture and are similar to those at Ongeama. Amphibole enrichment was observed neither in the Eastern Shoot exposure nor in the corresponding cores examined.

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2.3.3 The ore zone and the enclosing schists in drill cores

Drill cores of a total length of approximately 240m were examined in

detail in order to examine the relationships between the mineralization and enclosing schists, and to study the mineralization in unweathered samples. The drill cores studied are part of a drilling programme by Gold Fields Namibia, currently exploring the Eastern Shoot.

In the cores, lithologic changes (chloritization and magnetite dissemination) were recognised within the schists overlying the ore zone. These changes were originally interpreted as representing a zone of ore-related hydrothermal alteration (D.H. Corbett, pers. comm., 1989). The overturning of the sequence was inferred from the position of this chloritic alteration zone in the structural h/w of the mineralization. A complication to this preliminary assumption is provided by the presence of mafic units near the ore zone in the h/w. Therefore, one of the aims of the detailed study of cores was to find ways of discriminating between possible contact metasomatism related to the mafic intrusions and ore-related hydrothermal alteration.

Appendix B shows the logs of the four boreholes examined (Nos. OGB 151A, OGB 152A, OGB 153A and OGB 154A). These boreholes were re-logged from the position of the first appearance of chloritization downwards for about 45m. The first appearance of chloritization coincides with the first occurrence of the mafic units in the sequence. Several metres of schists in the immediate f/w of the mineralized intersection, down to the end of the hole, were considered as examples of unaltered schists. The f/w schists also include magnetite-bearing metapsammitic layers apparently not directly related to the mineralization. In these layers magnetite occurs as small grains, generally less than 0.5 mm in size.

The h/w schist sequence (see Appendix B and Appendix J) consists of alternating, planar to contorted, metapelitic staurolite-bearing and garnetiferous quartz-biotite-muscovite-chlorite schist and massive metapsammitic, locally garnetiferous quartz-biotite schist. In the detailed logs in Appendix J, the two schist end-members and intermediate schist types are schematically indicated by symbols. These symbols are commonly used during the current drilling programme to indicate schematically the proportion of phyllosilicates.

The mafic units occur as few-cm- to 3-m-thick intercalations. Augenamphibolite characterizes textured feldspathic the thickest intercalations. Coarse-grained, spotted chlorite-actinolite-biotite schist layers are well developed along the h/w margins of the amphibolite horizons. They can attain a thickness of approximately 80 cm and are in sharp contact with the underlying mafic unit. They commonly occur as an asymmetrically developed margin to the mafic units. Narrow layers of chlorite enrichment within the amphibolite units appear to be related to localized shearing. Thin chloriteactinolite-bearing bands are often intercalated with guartz-rich sediments in well defined horizons. The numerous mafic layers occurring in the h/w schist sequence can be grouped into two main bands separated by a 15m- to 25 m-thick schist sequence.

The chloritization observed in the schists is detectable as a weak, but perceivable change in colour towards a green hue in the usual light grey quartzites and phyllites. This change is more evident in the vicinity of both thin and thick mafic units, although it can also be observed elsewhere. The intensity of the chloritization is variable. Where most intense, it can be accompanied by disseminations of magnetite and locally of minute sulphide specks. The magnetite grains have an average size of 1 mm and rarely exceed 4 mm. In general, no increase in magnetite is observed in the immediate h/w contact of the ore zone, although magnetite enrichment is locally found proximal to the ore zone.

Garnet irregularly decreases in abundance in the h/w rocks towards the ore zone. This irregular garnet decrease appears to be accompanied by a concomitant staurolite enrichment.

The simplified stratigraphy of the ore zone is schematically represented in Fig. 19.

The ore zone is delimited by well-developed bands of coarse-grained brownish staurolite-rich biotite schists, or "biotite selvedges". These selvedges contain finely to coarsely disseminated lens-shaped sulphide grains (mostly pyrite) (see Plate 25). Within the ore zone



Fig. 19 - Schematic stratigraphy of the "ore zone" at Ongombo, in the thickest mineralized intersections (boreholes No. OGB 151A and OGB 153A). In the column zones of remobilized ore in veins are not indicated. Details of the hangingwall and footwall sequence are in the logs in Appendix B.

the mineralization occurs as disseminated, streaky, banded and locally semimassive coarse granular sulphide aggregates (pyrite + chalcopyrite and minor pyrrhotite) in both non-magnetic and magnetite-bearing guartzitic groundmass. Semimassive mineralized bands are intercalated with barren biotite- and staurolite-rich schists. In the schists sulphides occur in guartz lenses and streaks, and subordinately as disseminations. Remobilized ore in guartz veins and boudins occurs as semimassive sulphide blebs and patches, which commonly consist predominantly of chalcopyrite. Laminated magnetite quartzite, with disseminated and semimassive mineralized bands, occurs as two or more. thin, decimetre-thick units separated by zones of weaker mineralization. The lowest part of the ore zone is generally characterized by a poorly mineralized magnetite quartzite layer, up to 1 m thick, separated from the barren f/w schists by a biotite selvedge zone.

3 MATCHLESS WEST EXTENSION: SURFACE GEOLOGY OF THE MINERALIZATION

The Matchless West Extension sulphide deposit is situated about 25 km SW of Windhoek in proximity to the Matchless Mine. Both deposits are hosted in a schist sequence included between the "Colvania Horizon" and the mafic rocks of the Matchless Mine Horizon (Fig. 12 and 20) (see also Section 3 Part A).

Matchless West Extension is a small sulphide body with a maximum strike extension of about 240m. It is conformable with the enclosing schists which dip between 38° and 64°. Resistant rocks such as quartzite crop out as ridges, whereas pits expose part of the mineralized body in places. The deposit was mapped in detail, at a scale 1:500, by T. Smalley in 1984 under contract to Tsumeb Corporation Limited (TCL). That mapping was part of an exploration programme which covered the area between the deposit and the Colvania Horizon. This programme also included geophysical (electro-magnetic) surveys.



Fig. 20 - Stratigraphic section of the Matchless Belt in the Matchless Mine area, showing the position of the Colvania Horizon (CH), the Matchless Mine Horizon amphibolite (MMH) and the Friedenau Amphibolite Horizon (FAH) (Smalley, 1988). - 41 -

The map shown in Fig. 21 is a slightly modified version of Smalley's map. The latter was mainly used as a reference for the sampling of the deposit carried out by the present author.

The surface geology of the deposit is characterized by remarkable lateral lithological changes from E to W. Gossanous sericite schists (sericite zone) are well developed in the eastern sector of the area. These schists contain abundant fine-grained pyrite as disseminations or thin stringers parallel to the foliation. Sericite schists occur along most of the f/w of the mineralized body. In the eastern sector a distinct symmetrical zonation is observed in the exposure of the A fine-grained sugary-textured guartzitic facies sericite zone. (sericite quartzite) defines an elongated axial portion orientated at an angle of 12° relative to the strike of the mineralized body. This transverse band is symmetrically bordered by a more schistose, sericite-rich, locally chlorite-bearing facies. The sericite zone is overlain by a poorly exposed porous massive gossan in the east, and by a well developed gossanous, sulphide-rich magnetite guartzite band in the centre. The transition between these two lithologies appears to be gradational. The massive gossan displays a guartz-rich fine-grained groundmass, but it locally contains clots of minute phyllosilicatelike material. Unlike Ongeama and Ongombo, the magnetite quartzite has a very fine-grained quartz-rich matrix, in which small magnetite grains are distributed as dust-like disseminations and as well defined bands.

Talc- and actinolite-bearing schists occur as bands and lenses in the central sector. Locally, in this sector, they appear to be finely intercalated with quartzitic material or in direct contact with the gossanous sericite schists. The best development of talc- and actinolite-bearing schists is in the western sector, where a thick zone of these rocks is situated between two magnetite quartzite layers. This composite ultramafic layer consists of alternating, almost monomineralic, decimetre- to metre-thick, schistose to massive bands, in which either actinolite or talc is the dominant component. The weathering of rhombic carbonate porphyroblasts, commonly hosted in the talc-rich facies, leaves pitted rock surfaces. The gossanous



Fig. 21 - Geological sketch-map of the Matchless West Extension sulphide deposit (modified after Smalley, TCL, 1984).

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appearance of these ultramafic units is due to copper staining and intercalations of weathered sulphide-rich quartzitic bands. The spatial association of these rock types with magnetite quartzite suggests an exhalative origin (see Section 3.2.3. Part C).

The two magnetite quartzite layers, including the ultramafic band in display different the western sector. characteristics. The stratigraphically lower layer is massive, poorly banded, very dark due to the exceptional abundance of finely disseminated magnetite, and it often contains malachite-stained, limonitized sulphide streaks. A faint compositional layering is suggested by these gossanous streaks, by the localized denser magnetite concentrations and by the presence of thin amphibole-bearing lenses. Local black staining and crusts suggest Mn-enrichments. The upper magnetite quartzite layer is distinctly non-gossanous and well banded. The thin amphibole-rich lenses rapidly decrease towards the h/w.

Beyond the river on the western side a white marble-like rock, with scarce disseminated pyrite, occurs next to the gossanous magnetite quartzite exposure.

Wall rocks are represented by garnetiferous quartz-biotite schists of the Kuiseb Formation in the f/w, and by massive to sheared amphibolites (metagabbro) and chlorite-amphibole schists in the h/w.

An outstanding deformation-related feature observed by T. Smalley (1989, pers. comm.) is the development of an "ore cusp", or sulphide injection structure in the central-eastern portion of the exposure (see Fig. 21). The cause may have been competency contrast between the sulphide mineralization, the overlying rigid metagabbro and the underlying quartz-sericite schists, during the main deformation phase (D_2) . Similar cuspate ore-wall rock interfaces were observed also at the Matchless Mine (Maiden et al., 1986), where they are responsible for localized thickening of the ore zone. The later, less severe, D3 deformation phase induced only small-scale crenulation cleavage in the finely laminated guartz-sericite schists.

4 THE SAMPLING PROGRAMME

A detailed and systematic lithogeochemical sampling programme was carried out by the author on mineralized and barren outcrops and on cores in the areas of study. A total of 487 samples was collected and submitted for chemical analysis. Appendix C contains a list of the samples, their location and the rock type.

4.1 Sampling procedure on the mineralization

At Ongeama, the gossan exposure has been systematically sampled along a 20m-spaced grid. This grid was also used for geological mapping and was based on the existing 50m-spaced grid established by Gold Fields. Composite chip samples, with the approximate weight of 1 kg, were collected at lithological changes considered significant on the scale of the mapping. Due to poor outcrops, samples of barren schists had to be collected outside the area studied, but, nevertheless, within the main grid.

At Matchless West Extension sampling was carried out along a 10mspaced grid. The new grid was superimposed on an earlier grid established by TCL in the early 80's.

At Ongombo the sampling of the 4.7km-long mineralized outcrop was organized in a different way. The assumed zones of intersection of the EMP trends (i.e., ore shoots) with the surface were defined. Within these zones, samples were collected every 20m, whereas, in the remaining portions, samples were collected every 50m. One or two samples only were collected for small, isolated outcrops along strike. The sampling was done according to the existing grid established by Gold Fields Namibia (Fig. 18).

Drillcores examined from the Ongombo drilling programme (borehole No. OGB 151A, OGB 152A, OGB 153A and OGB 154A), were sampled from approximately 16-20 m in the h/w, across the "ore zone", to 6-10m in the f/w. The whole core was sampled, except in the "ore zone". In this portion the core was guartered and only one guarter sampled.

4.2 Sampling of the "I anomaly" area at Ongeama

At Ongeama the "I anomaly" is located approximately 800m NW of the gossan outcrop (Fig. 16). The "I anomaly" is an area characterized by a positive magnetic anomaly in the form of clustered peaks. In the field these magnetic peaks were found to coincide with metapsammitic, and subordinate calc-silicate, units containing fine-grained disseminated magnetite grains. Only in one case (sample OG 138, grid ref. 9905E/11058N) does coarse-grained magnetite occur in a few cmthick chlorite-rich selvedge wrapping around a quartz pod or vein.

Samples have been collected from each area corresponding to a positive peak. According to what was observed in the south-easterly gossan exposure, the sampling of the "I anomaly" was done in order to check whether these magnetite disseminations might be associated with an ore-forming hydrothermal system. However, no positive evidence of hydrothermal alteration was observed in the field. Mafic rocks (metagabbros and chlorite-amphibole schists) are present towards the northwestern boundary of the "I anomaly" area.

Comments on the trace element content of the "I anomaly" samples are given in section 2.5 Part D.

4.3 Sampling of ultramafic rocks

Two ultramafic bodies outcropping in the Eastern Windhoek sector of the Matchless Belt were sampled. In section 4 Part D, the geochemistry of these bodies is compared with that of the ultramafic schists in the ore zone at Matchless West Extension and with that of the chloriteactinolite margins of the metagabbroic horizons in the Ongombo cores.

The ultramafic bodies sampled are located NNE and WNW of the Otjihase Mine, in the Van Francois Ost 60 and Elisenheim 68 farm areas (for location see Fig. 22 and 12). The ultramafic outcrop at Elisenheim Farm was previously studied by Finnemore (1974). The ultramafic bodies considered belong to the group of Damaran Alpine-type serpentinites studied by Barnes (1983).

These ultramafic bodies consist essentially of talc- and actinoliterich lithologies, crenulated to massive talc-carbonate schists with varying amounts of chlorite and magnetite and coarse actinolite-



chlorite schists. In the Van Francois Ost outcrop, the actinolitechlorite schists occur as a peripheral band around a talc-carbonate magnetite-rich core (Fig. 22). Such concentric zonation is similar to that reported by Sanford (1982) for the metasomatic reaction zones developed at the contacts between ultramafic bodies and country rocks in greenschist- to amphibolite-facies metamorphism.

5 DISCUSSION

The most significant result of the detailed work at Ongeama and at Matchless West Extension is the recognition of some schist facies, which are exclusively associated with the mineralization, and quite distinct from the country rocks. Based on field relationships, it is the opinion of the present author that these schist facies suggest the occurrence of pre-metamorphic ore-related hydrothermal alteration zones. The discordant nature of the lithologic changes characterizing the schists within the assumed alteration zones, does not appear to be the result of deformation and/or metamorphism. The distribution of the rock types reflects the geometry of typical pipe-like alteration zones associated with volcanogenic sulphide deposits. In the alteration pipe, facies changes may occur both laterally and vertically.

At Ongeama the magnetite-staurolite-garnet-bearing chlorite schist facies may represent the metamorphism of an aluminous and chloritic axial zone of an alteration pipe. The occurrence of abundant of the Fe-rich environment. staurolite is an indication The marginal magnetic muscovite schist band seems to associated, correspond to an outer sericitic envelope. Sulphides in fine disseminations and small lenses in the altered schists can be disseminated interpreted as an original and stringer-type mineralization developed along the hydrothermal fluid conduit. The outcrops of these schist facies, compared to the more subdued exposures of the country rocks, may, in fact, reflect an original silicification along the pipe. The outermost non-magnetic ferruginous schists may represent the product of the interaction of waning ore-
forming fluids with the sediments. An alternative interpretation is that these gossanous schists may be the metamorphosed equivalent of chemical precipitates (i.e. montmorillonite-kaolinite, silica gel and sulphides) deposited as a hydrothermal apron on the sea-floor contemporaneously with the massive mineralization (McLeod and Stanton, 1984).

The position of the assumed alteration pipe in the h/w of the mineralization at Ongeama suggests the overturning of the stratigraphic sequence. The attitude of the pipe is similar to the plunge direction of the ore shoot, as indicated by the corresponding magnetic anomaly (Fig. 16), and by the lineation in the magnetite quartzite outcrops (Plan 1). A sinistral shear movement is inferred from the flattening undergone by the pipe (Fig. 23).

Similar considerations can be made for Matchless West Extension. On the basis of field observations, the author agrees with Smalley's (1990) opinion that the sericite zone in the f/w of the mineralized horizon may represent the metamorphosed equivalent of an original alteration pipe-like zone (Fig. 24). This discordant alteration pipe essentially of a quartz-sericite-pyrite consists alteration assemblage. The symmetric lateral zonation, from the silicified axis to the more mica-rich margins, is not obliterated by deformation and metamorphism. A sinistral shear deformation is evident also at Matchless West Extension, as it is at Ongeama. The presence of this unusual lithological association in the f/w of the mineralized body should imply that the ore-hosting sequence is the right way up. Additional field evidence confirming the above statement can be seen in the two magnetite quartzite horizons in the western sector of the outcrop. In these two horizons, the gradually upward decreasing sulphide abundance seems to reflect the normal transition from an underlying sulphide-richer portion of the mineralization to an overlying sulphide-poor oxide cap, as observed in similar, undeformed volcanogenic sulphide deposits (Lydon, 1988).

The distribution patterns of trace elements along the alteration pipes at Ongeama and Matchless West Extension provide further data on the



Fig. 23 - Reconstruction of the possible original mineralized system at Ongeama (a) on the basis of its present attitude (b) (modified after Sangster and Scott, 1976).



Fig. 24 - Schematic plan with the lithological zoning at Matchless West Extension (modified after Smalley, 1988).

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nature of these pipe-like features (see Part D).

A further confirmation of the hydrothermal nature of these lithologic associations is provided by their greatest development in correspondence to significant facies changes of the mineralized body. For example, at Ongeama the intersection between the assumed alteration pipe and the sulphide body is characterized by the development of the amphibole-rich facies within the massive sulphide portion. At Matchless West Extension the top of the alteration pipe coincides with the occurrence of a richly mineralized portion of the ore horizon, which, in the field, is represented by massive sulphide gossan and gossanous, sulphide-rich magnetite quartzite.

The above considerations cannot be extended to Ongombo with certainty at this stage. The areally localized lithological changes (e.g. amphibole enrichment), observed in the gossan, may be a primary feature of ore deposition, and may suggest the position of a possible alteration pipe. However, they cannot be related to any rock association suggestive of hydrothermal alteration in the field because of insufficient outcrop.

The association of amphibole and sulphide enrichment with the thickening of the mineralized body in the Central Shoot may, alternatively, be interpreted as a deformation-related feature. As a matter of fact, the inflection of that part of the outcrop is considered by Hoffmann (1976) to have been induced by a major monoclinal fold structure. This feature stands out from the minor structural undulations present throughout the gossan outcrop (Fig. 18).

In the Ongombo cores, chloritization is the only major (although weak) lithologic change in the h/w schists. Magnetite disseminations are not evenly distributed. Chloritization can be seen at distances exceeding 40m from the ore zone and coincident with the first occurrence of mafic units in the drill cores. In consideration of the extent of deformation, this distance appears to be excessive for the chloritization to be considered as a product of hydrothermal

(RHODES UNIVERSITY LIBRARY alteration related to the ore zone. Moreover, no significant chlorite increase or change is observed in proximity of the ore zone. However, visual observation alone of the cores is not sufficient to evaluate the obliterating effects of metamorphism on hydrothermally altered schists, or to discriminate the effects of ore-forming fluids, and of sill intrusions into the sediments. A discussion about the significance of the chloritization in the cores at Ongombo is, therefore, reported at the end of the next chapter, in which details of the mineralogical and textural characters of mineralization and related alteration are discussed.

PART C

PETROLOGY OF MINERALIZED AND BARREN ROCKS

1

1. INTRODUCTION

The mineralogical and textural features of the metamorphosed volcanosedimentary sequence, of the enclosed mineralized bodies and of the associated zones of pre-metamorphic hydrothermal alteration in the areas of study are described and discussed in this chapter.

A list of the thin and polished sections examined and relative locations is given in Appendix D.

2. PETROLOGY OF THE COUNTRY ROCKS

2.1 Petrology of the metasedimentary Kuiseb Formation and of the intercalated mafic rocks of the Matchless Member

2.1.1 Schists of the Kuiseb formation

In the area of study the bulk of the Kuiseb Formation is represented by strongly micaceous metapelitic schists and subordinate metapsammitic, quartzitic schists with locally associated "calcsilicate" intercalations. Gradations between the two main schist types are observed both on the scale of the hand specimen and thin section.

The main mineralogical components of the <u>metapelites</u> (OG 755, 756 and sections from OGB 153A) are biotite, chlorite, muscovite, quartz, plagioclase (oligoclase), staurolite and garnet. Epidote (clinozoisite), apatite, opaque minerals, zircon and tourmaline occur as common accessories. Variations in the relative abundance of biotite and chlorite result in the occurrence of more biotitic or more chloritic facies. The minerals are arranged in alternating granoblastic quartz-plagioclase-rich microlithons and lepidoblastic cleavage stripes, in which the fine- to medium-grained accessories are variably distributed.

Biotite occurs in two distinct associations:

a) a chlorite-biotite association, intergrown with muscovite in the cleavage planes, is characteristic of greenschist facies metamorphism (Winkler, 1979) and occurs as well oriented fibro-lamellar aggregates

which define the foliation;

b) a staurolite-biotite association, which probably derives from the breakdown of chlorite in the presence of muscovite at the boundary between greenschist and lower amphibolite facies metamorphism (Winkler, op. cit.). Such association is characterized by lenticular aggregates of biotite and euhedral staurolite porphyroblasts grown discordant to the foliation. Biotite-staurolite layers alternate with the chlorite-biotite laminae, indicating the development of prograde metamorphic reaction in a "lit par lit" fashion.

Poikiloblastic garnet porphyroblasts are generally associated with staurolite and may display an atoll-like growth pattern. Muscovite lamellae often rim staurolite grains.

Retrograde metamorphic replacement of biotite by chlorite is common and can be recognized by preserved biotite inclusions in poikilitic sulphide blebs growing on chlorite plates (OG 756).

In the area of study the metamorphic paragenesis in the metapelites is comparable with that characterizing the staurolite + chlorite + biotite zone recognized by Hoffer (1983) in the phyllites just W and S of Windhoek. Hoffer estimates the temperature and pressure conditions for the biotite + staurolite association at 590°C and 5 kb. In the metapelites, as well as in the other rock types examined, a pronounced metamorphic recrystallization is indicated by the granoblastic polygonal texture of the quartz aggregates.

In the thin sections examined at least two metamorphic foliations, S₀₋₁ and S₂, and a D₃-related crenulation are recognized. The tight isoclinal D₂-related folding of the S₁ foliation is shown by the disposition of mica flakes and tabular opaque minerals along fold hinges. Layers with remnant fold structures, as pod-like fold closures, alternate with layers with a composite S₀-S₁-S₂ foliation. In these layers a strong tectonic transposition of the folded S₁ has been enhanced by the further tightening of folding. Quartz boudins are syntectonic to S₂. The D₃-related crenulation is shown by local gentle to narrow folding and kink banding in the S₂-oriented micaceous laminae. The common helicitic texture, defined by the rectilinear to sygmoidal and spiralled disposition of the inclusions, indicates the syn-late timing of the garnet nucleation relative to the S₂-generating deformation phase. The euhedral staurolite and associated biotite porphyroblasts growing along S₂ are not deformed. The arrangement of the inclusions in staurolite parallel to S₂ suggests a static blastesis and consequently the development of the metamorphic peak (lower amphibolite facies) after the D₂ deformation phase.

The <u>metapsammites</u> (OG 693 and sections from OGB 153A) are fine-grained quartzitic schists with poor to indistinct compositional layering. Their mineral assemblage consists of quartz, plagioclase, biotite, epidote, apatite, opaques and zircon. Garnet occurs locally. Short biotite lamellae grow both isolated and in discontinuous laminae in the granoblastic quartzose groundmass. They impart the slight foliation to the rock type. A single foliation is recognizable. Retrograde metamorphic chlorite replaces biotite.

The <u>calc-silicate</u> layers (OG 757) essentially consist of a mica-free hornblende-epidote-rich, plagioclase-bearing quartzite with a distinctive grano-porphyroblastic texture. The poikiloblastic porphyroblasts of greeny-blue hornblende are mostly arranged parallel, locally transgressive, to the foliation. Inclusions consist of coarsegrained subhedral epidote and quartz and plagioclase from the groundmass. The assemblage might be suggestive of an indirect contribution from mafic volcanic activity.

2.1.2 Magnetite-bearing schists

A few sections from magnetite-bearing schist units have been examined in order to characterize the mode of occurrence of the magnetite disseminations in the sequence, and to verify its possible association with ore-related hydrothermal activity. The schists occur both in the f/w and in the h/w of the ore zone at Ongombo.

The selected sections represent four different modes of magnetite occurrence, namely:

a) apparently not related to hydrothermal alteration, i.e. distant from the ore zone or in the f/w of the same (OG 688);

b) in the relative vicinity of (sulphide-magnetite-bearing) quartz veins but not proximal to the ore zone (OG 689A, 690);

c) in the h/w of the ore zone, within a maximum distance of 50-60 m from the ore zone itself, and spatially related to thin to coarse mafic units (OG 692A-B, 697 and 699);

d) in the immediate vicinity of the ore zone (within a few m in the h/w) although close to mafic units (sections from core OGB 153A).

Schist types containing disseminated magnetite are represented by both quartz- and mica-rich facies. The euhedral magnetite crystals or aggregates mostly cross-cut the foliation, and only locally (OG 688) deform the surrounding micaceous laminae during growth. The grain size of the crystals varies from less than 1 mm to 1.5 mm. Magnetite is coarser in the occurrences c) and d).

Fine-grained massive quartz-biotite schists seem to be more frequently carriers of disseminated magnetite away from the ore zone or from the mafic rocks. In these schists magnetite occurs in modes a) and b) described above. No remarkable mineralogical change distinguishes the schists of these sections from those devoid of magnetite (see section 2.1.1 above). The only distinguishing feature is the moderately poikiloblastic nature of some magnetite grains in schists intersected by guartz veins, and the the partial chloritization of biotite (in OG 690). The former indicates progressive grain growth, while the latter might merely be related to retrograde metamorphism.

Disseminations of magnetite in proximity to mafic units (occurrence c)) generally occur in both chlorite-rich metapsammitic and metapelitic schist facies (OG 682B, 697, 699), and together with sieve-textured actinolite porphyroblasts. The latter are frequently conformable with the foliation and possibly indicate the proximity of mafic units (OG 682A-B). Minute sulphide blebs are disseminated.

In parts of the h/w schists above the ore zone (core OGB 153A) (occurrence d), magnetite can be fairly abundant, and it occurs as isolated or clustered grains in moderately chloritized quartz-biotite schists.

2.1.3 Mafic rocks and their relationship to the enclosing schists

Metalavas (MAB 261, 264, OTJ 682 and OG 135) are characterized by a fine to very fine-grained and finely foliated fabric. A fine-grained fibrous aggregate of acicular, pale-green to bluish actinolite with nematoblastic texture is intergrown, or alternates, with minor granoblastic, fine-grained plagioclase-quartz lenticular aggregates. The zoned plagioclase grains are oligoclase in composition. Chlorite is either not present or, if so, only in minimal guantities as remnant flakes. Prismatic to tabular. fineto coarse-grained clinozoisite/epidote is a common constituent. It occurs as regular disseminations in the amphibole-rich groundmass; as coarse-grained knot-like aggregates; in crosscutting contorted and boudinaged monomineralic veinlets (MAB 261). Apatite is a common and rather abundant accessory, together with sphene. Rutile is the Ti-bearing phase in OTJ 682.

The foliation is commonly planar. Wavy foliation wraps around epidote knots (MAB 261) and particularly around lensoidal, stretched nodules consisting of coarse quartz, epidote and amphibole. Such centimetre-scale nodular features, observed in OTJ 682, occur in the Otjihase metalava horizon, where structures interpreted as deformed pillows are found. The nodules are characterized by a pronounced grain size increase, by increase in the quartz content, by the chaotic arrangement of the few amphiboles and by dissemination of sulphide blebs.

The mineralogy of the examined <u>metagabbroic horizons</u> from Ongombo is only slightly different from that observed in the metalavas outcropping SW of Ongombo.

The main components are hornblende and plagioclase (andesine), with

minor quartz, biotite, chlorite, apatite and epidote. The colour and shape of the hornblende crystals suggest that this mineral might be intermediate towards an actinolite composition. The minerals are arranged in medium-grained grano-nematoblastic aggregates which have undergone intense shearing. Tabular ilmenite, sulphide blebs and locally abundant magnetite (OG 700) are disseminated in the groundmass. The characteristic spotted, augen-like appearance of this rock type is imparted by the flattened granular plagioclase aggregates. Plagioclase grains are poikiloblastic, zoned and rarely twinned.

The mineral assemblage observed in the metamorphosed mafic rocks of the Matchless Member in the areas of study compares well with that marking the greenschist-amphibolite metamorphic facies boundary (Moody et al., 1983). This boundary is characterized by:

a) the growth of low-Al, -Ti and -Na amphibole (actinolite or actinolitic hornblende) coupled with a decrease in the amount of chlorite present as main constituent in the greenschist assemblage;

b) change in the plagioclase composition from albite- to anorthitericher;

c) breakdown of sphene and formation of ilmenite;

d) changing composition of the coexisting ilmenite-magnetite-rutile assemblage with increasing temperature.

These changes are temperature- (Fig. 25) and f_{02} -dependent. The pressure conditions attained during these changes are indicated by the occurrence of staurolite in the metapelites.

Although a single observation is not sufficient, the mineral assemblage in the Ongombo amphibolites seems to point to a slightly higher metamorphic grade than in the Ongeama and Matchless Mine area. This is, however, against the regional trend of increasing metamorphic grade towards the WSW, as reported by Miller (1983) for the Damara

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Fig. 25 - Metamorphic mineral assemblages, as functions of total fluid pressure and temperature, in greenstones at the transition greenschist-amphibolite grade. Curve (a) marks the chlorite-decreasing amphibole-increasing boundary and sphene-out boundary; curve (d) marks the epidote-in boundary (modified after Moody et al., 1983).



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Fig. 26 - Sketch illustrating the mineralogical zonation in the hangingwall margins of the amphibolite units in the drillcores from Ongombo.

Orogen.

The <u>h/w margins of the meta-sills</u> (OG 695, 712) consist of coarse, foliated to massive, talc- and carbonate-bearing actinolite schists. Abundant talc fibres, biotite and/or chlorite plates and very little quartz are interstitial to the actinolite-rich groundmass. Carbonate (calcite?) is in finely granular aggregates, locally replacing amphiboles along the cleavage planes. Disseminated, medium-grained rutile is common.

A rapid increase in the mica content and a concomitant decrease of the amphibole mark the transition to the chilled margin of the sill. This chilled margin is characterized by foliated, compact fibro-lamellar aggregates of coarse-grained chlorite and/or biotite with very subordinate interstitial guartz. Biotite often overgrows chlorite. Rutile, isolated or in clusters, and apatite can be unusually abundant. Coalescing coarse chlorite in radiating and fan-like aggregates, accompanied by euhedral actinolite porphyroblasts, give way to a well developed chlorite-actinolite selvedge zone along the margin of the sill. Elongated prismatic actinolite crystals freely grow within quartz veins, parallel to the chloritic sill margin (OG 694, 695). Fig. 26 illustrates the mineralogical and textural changes from amphibolite across the ultramafic layer to the chilled margin and the schists at the contact. The presence of coarse-grained chlorite and the growth of the chlorite "fans" suggest activity of aqueous fluid as a result of the interaction between wet sediments and the intrusive mafic sill.

The <u>schists at the contact with the mafic units</u> (OG 692A-B, 694, 696, 697, 698 and 699) record several mineralogical changes:

a) a general decrease in biotite, an increase in the abundance of chlorite and an increase in its grain size. This might mean an original chloritization of the sediments, preserved during metamorphism. This feature is evident in OG 692A-B and 694; b) dissemination of sulphides, as fine- to medium-grained anhedral blebs and flattened aggregates, and medium-grained euhedral magnetite (OG 692A-B, 697, 699);

c) increase of abundance and grain size of apatite and, locally, of rutile (OG 694 and 695);

d) coarsening of the quartzose groundmass;

e) local dissemination of pale green actinolite porphyroblasts parallel to the foliation (OG 682A-B).

Poikiloblastic, often epidote-rich green-blue hornblende porphyroblast enrichment, occurs in the quartzitic schists at the contact with mafic units (OG 689A-B). Because of their similarity to calc-silicate lenses far from mafic units, these may not be related to the effect of the mafic sill intrusion. They probably represent a calc-silicate metapsammitic facies.

3 PETROLOGY OF THE MINERALIZATION AND ASSOCIATED ALTERATION

Because of the lack of unweathered samples, microscopic studies of the ore zone at Ongeama were limited to a qualitative evaluation of the gossan textures by means of thin and polished sections.

In Matchless West Extension attention is focussed on aspects of the alteration zone and of the well developed magnetite quartzite and talc-actinolite-carbonate facies. The latter are mineralogically compared with samples of Damaran serpentinites.

The nature and texture of the Ongombo "ore zone" and enclosing schists will be considered in this Section; in this case, the term "ore zone " is the name given to the sulphide-bearing portion of the sequence, inclusive of the alteration zone. 3.1 The Ongeama sulphide deposit

3.1.1 The gossan

The petrology of the three main facies recognized in the gossan exposure at Ongeama (i.e. massive sulphide gossan, amphibole-rich semimassive gossan and magnetite quartzite) and of the ferruginous schist pockets associated with the amphibole-rich gossan facies, is discussed in this Section.

The strongly leached character of the surface gossan is such that no residual sulphide grain is preserved within the existing boxwork structures. Therefore, replica textures, when not obliterated by massive or concretionary exotic limonite, have been tentatively interpreted according to the criteria established by Blanchard (1966) and Blaine and Andrew (1977), and with the aid of Hoffmann's description of the mineralization in cores (see also Sect. 2.2.1 Part B).

Limonite is the main constituent of the porous <u>massive sulphide gossan</u> <u>facies</u> (OG 15, 38 and 47A-B). Quartz occurs as recrystallized, coarseto medium-grained aggregates, probably representing the groundmass of the mineralization. Muscovite, fibrous amphibole (anthophyllite) and fragmentary garnet occur as occasional accessories, isolated in limonite crusts. A faint banding is only recognizable in OG 15, where few mm-wide porous limonite bands irregularly grade to limoniteimpregnated quartz-rich layers. When present (OG 47A-B), the tiny fibrous anthophyllite aggregates occur within the quartz groundmass or along the boundaries between the former and the limonitic spongy patches.

Different Fe-hydroxide species are distinguished by their different colour and reflectivity in thin and polished sections, respectively. The term "limonite" is here used as a general name for Fe-hydroxides.

In thin section, dark brown-blackish limonite corresponds to the "autochtonous" product of weathering of the sulphides, whereas the reddish limonite clearly shows its exotic nature. Brown limonite constitutes areas of preserved boxwork textures and compact, rounded remnants of clustered sulphide grains. Limonite pseudomorphs after sulphides are surrounded by exotic concretionary limonite. These limonite crusts also coat fractures and form large colloform patches in which every previous structure is obliterated.

The brown limonite-bearing portions are dominated by a thin- to thick-walled sponge-like texture containing recognizable sulphide replicas. Thin-walled limonitic "sponges" consist of medium-grained round to elongate cells reflecting the granular nature of the recrystallized ore. Progressive thickening of the cell walls frequently occurs by precipitation of brown and of late reddish limonite. Square boxwork replaces pyrite grains (Plate 5). Within the thin-walled cellular portions the fine-grained limonite crusts which often coat the inner walls of the cells, suggest pyrrhotite-derived textures (see Blanchard, 1966). By contrast, smooth-walled cells are more indicative of pyrite. Limonite boxwork faithfully mimics concentric shrinkage cracks, typical of bird's eye weathering of pyrrhotite in OG 38 and 47B (Plate 6). In these sections delicate ladder-like patterns consisting of thicker longitudinal walls and thinner transversal septa, which might be chalcopyrite replicas, can be seen (Plate 5).

The <u>amphibole-bearing semimassive gossan facies</u>(in OG 63) is a porous rock type deficient in quartz. Anthophyllite and limonite are the only constituents. Anthophyllite occurs in densely clustered rosettes and variably oriented fibrous aggregates, which give way to a network-like pattern. The amphibole aggregates are impregnated with reddish limonite along cleavage and transverse cracks (Plate 7 and 8). In places where the amphiboles were dissolved during weathering, a delicate and elongate ladder-like boxwork similar to that after chalcopyrite, and a cellular texture with minute rhombic cells are observed. Round blackish limonite patches probably replace original sulphides. Occasional cubic cells are possibly after pyrite. As in the massive gossan, cells coated with limonite crusts probably replace pyrrhotite grains.

In the <u>magnetite quartzite</u> (OG 17, 19A-B), the recrystallized granular quartz groundmass contains patches and lenticular aggregates of finegrained euhedral magnetite which form a discontinuous banding at the scale of the thin section. Gossanous sulphide patches and limonitized anthophyllite aggregates are contained in the groundmass. Coarse actinolite is occasionally present in the anthophyllite aggregates. Magnetite grains are mostly unweathered, but are sometimes surrounded by crusts of exotic limonite. Cellular gossan is probably indicative of pyrite.

The ferruginous schist (OG 64A-B) associated with the semimassive amphibole-rich gossan facies is composed of yellowish mica, magnetite and limonite. The strongly schistose fabric is caused by the dominant lamellar, lepidoblastic micaceous groundmass, which shows evidence of folding. Where the limonite staining is minimal, the mica lamellae are slightly but distinctly pleochroic from pale yellowish to pale brown, and their interference colors are similar to that of biotite. Although the identity of this mica cannot be clearly established under the microscope, it does resemble phlogopite mica. Magnetite occurs as abundant medium to coarse-grained crystals, either isolated or more often in clusters, which seem to replace the micaceous groundmass. An unknown mineral occurs in places. This mineral is fractured. It forms roughly round grains, including few mica flakes, and is strictly associated with magnetite clusters. Black limonite films penetrate its fractures. The optical properties of this mineral, i.e. high relief, transparent color with greenish tinge, isotropy, might be indicative of a garnet or a spinel. In places a faint double set of almost perpendicular cleavage suggests a spinel: either the Mg-Fe-bearing pleonaste or the Zn-bearing gahnite (Plate 9).

3.1.2 The hangingwall schists

In this section the petrology of the unusual schists in the h/w of the Ongeama sulphide body is described from the central part of the outcrop towards the margins.

The <u>chloritic schists</u> in the axial portion of the discordant pipelike feature described in Section 2.2.3 (OG 77A-B-C), show a markedly foliated fabric and a distinctive porphyroblastic texture. The mineral assemblage includes quartz, chlorite, garnet, chloritoid, staurolite, magnetite and limonite. Quartz and chlorite form the schistose groundmass. The groundmass shows two foliations: one is planar to crenulated S_2 and the other is a tightly folded S_1 .

Both prismatic staurolite and tabular pale blue-violet chloritoid occur as coarse skeletal porphyroblasts. They locally display characteristic penetration and polysynthetic twinning, respectively. They are often associated with chlorite and carry disseminations of tiny sulphide blebs. The growth of staurolite seems to occur at the expense of chloritoid. Lamellar chlorite aggregates appear to form a reaction rim between staurolite and chloritoid. They are considered as a localized and incipient retrograde metamorphic breakdown of chloritoid. Chloritoid is also found in contact with both garnet and staurolite (Plate 10) or overgrown by garnet alone. Garnet (almandine) occurs as coarse, partially limonitized skeletal porphyroblasts, whose S2-synkinematic growth is indicated by the helicitic spiralled texture (Plate 11).

Magnetite occurs as porphyroblastic aggregates of coarse subhedral grains. The generation of magnetite may be ascribed to the reaction (Deer et al., 1982):

 ${chloritoid + qtz + 0_2} = {staur + almand + magnetite + H_20}$

Fig. 27 illustrates the relationships between staurolite and chloritoid at the transition between upper greenschist and lower amphibolite facies metamorphism in moderate to high pressure conditions.

The reaction

{Al-silicate (pyrophyllite or kaolinite) + Fe-rich chlorite} =
{chloritoid + qtz}

is considered responsible for the first appearance of chloritoid in metapelites under conditions approximating the greenschist facies (Winkler, 1979). Chloritoid is also formed through the reaction



Fig. 27 - Metamorphic reactions in pelitic rocks as function of total pressure and temperature (Winkler, 1979). Note: curve (5) marks the chloritoid-out-staurolite-in boundary.

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{Fe-oxide + Fe-rich chlorite} = {chloritoid + magnetite + quartz},

which might also be responsible for an initial nucleation of magnetite. Chloritoid formation has also been related to zones of shearing stress (Deer et al., 1982).

The rusty, spotted <u>magnetite-muscovite schist</u> (OG 62) occurs as a sheath to the axial chlorite zone. This schist (Plate 12) consists of abundant quartz, muscovite, magnetite, staurolite, garnet and minor chlorite. The rusty appearance is due to limonite derived from weathering of disseminated sulphides. Garnet is limonitized along cracks. Muscovite laminae define the planar to wavy S₂ foliation, where syn-kinematic skeletal staurolite and flattened garnet are grown. Magnetite porphyroblasts crosscut the foliation. Chlorite is present in some weathered lamellae isolated in the quartzose groundmass or interleaved with muscovite.

The <u>transition</u> from the chloritic axial zone to the muscovite sheath, as described in Sect. 2.2.3 Part B, is more obvious in the field. Nevertheless, on the scale of the thin section (OG 78) it can be noticed as an increase of the chlorite content within the muscovite laminae.

The <u>non-magnetic ferruginous schists</u> in the peripheral portion of the muscovite zone are strongly gossanous and show a very simple mineralogy. They consist of a grano-lepidoblastic aggregate of quartz and muscovite with disseminated, weathered sulphide blebs, and accessory tourmaline. Dark brown limonite replaces sulphides and impregnates the groundmass, whereas reddish limonite concretions encrust fractures and cavities.

3.2 The Matchless West Extension sulphide body

The petrological study of the Matchless West Extension was mainly

focussed on the alteration and exhalite lithologies. Only one thin section of gossan material was examined.

3.2.1 Gossan and alteration zone

Section MWE 207 represents a <u>talc-rich facies of the porous massive</u> <u>gossan</u> outcropping just above the alteration pipe. Talc occurs in a band within the thin section. It occurs as fibrous-flaky aggregates and coarse bent lamellae, which grow on a coarse, irregularly granular quartzose groundmass (Plate 13). Brown limonite impregnates the talc and quartz aggregates, and forms coarse, irregular thickwalled sponge-like and massive patches, where boxwork structures are no longer preserved. An unknown secondary mineral occurs as concretionary and botryoidal crusts. It is colourless, with moderate relief and isotropic.

The <u>marginal facies of the alteration pipe</u> is characterized by a finely foliated muscovite-chlorite schist (MWE 204). Muscovite and chlorite occur as parallel lamellar bands, locally associated with phlogopite lamellae. Coarse muscovite plates locally replace chlorite aggregates, and they may belong to a second generation of muscovite. Tiny sulphide blebs and prismatic rutile are present as disseminations.

The mineralogy of the <u>axial part of the alteration pipe</u> is fairly simple. It consists of variable proportions of quartz, muscovite, limonitized pyrite and accessory tourmaline (MWE 205, 217, 219) (Photo 14). Pyrite is disseminated as tiny cubic weathered crystals in the schistose grano-lepidoblastic quartz and muscovite groundmass. Limonite impregnates quartz and mica as thin films and curiously also as very dense disseminations of black and red spherulites (MWE 205).

The foliation is planar to crenulated. Complex kink-banding in the micas is visible in some sections, where the quartzose laminae also show evidence of strain. A new foliation, well displayed in hand specimens, is formed by recrystallization of mica plates along a cleavage plane inclined approximately 30° relative to the old

foliation.

3.2.2 Exhalites

The exhalites at Matchless West Extension include at least two rock types: magnetite quartzite and ultramafic schists. In the field the two rock types appear to grade into one another. Two samples were examined from this transition zone.

Sections of MWE 241A-B and MWE 253 represent the two different <u>magnetite quartzite</u> layers characterizing the western portion of the outcrop (see Sect. 3 Part B).

The upper magnetite quartzite band (Plate 15) is massive, well banded and non-gossanous. It contains exclusively quartz, magnetite and baryte. Euhedral fine- to medium-grained magnetite is densely clustered in bands in the recrystallized quartz matrix. Medium- to fine-grained, round to amoeboidal baryte grains occur both as interstitial material in the quartz groundmass and as inclusions in quartz grains.

The gossanous, lower magnetite quartzite band (Plate 16) is characterized by a remarkable increase in magnetite, that almost obliterates the banding. Quartz, magnetite and baryte are accompanied by actinolite and subordinate limonite and malachite. Actinolite is in coarse-grained poikiloblastic prismatic porphyroblasts containing inclusions of quartz and magnetite. They are grouped in lenses parallel to the banding. Baryte blebs, irregular patches of weathered sulphides and encrusting emerald-green malachite aggregates are commonly associated with the actinolite-rich bands.

The ultramafic schists display the following fabrics:

a) strongly foliated, especially in some talc-rich facies, as in MWE 246B, and in the actinolite schist in MWE 245B;

b) massive, as in the monomineralic actinolitite in MWE 239 or in the talc-phlogopite schist in MWE 246A (Plates 17 and 18);

c) porphyroblastic, particularly where carbonate porphyroblasts occur, as in MWE 245A.

The foliated actinolite schists (MWE 245B) are essentially monomineralic, being composed of a medium- to coarse-grained nematoblastic actinolite aggregate. Phlogopite and interstitial talc and quartz are rare.

The massive "actinolitite" is outstanding for its coarse grain size. The interstices of this amphibole-rich, felted aggregate are occupied by fine-grained granular yellow-greenish patches of an unknown secondary mineral, limonite and rare quartz grains.

The massive fibrous talc schist in MWE 246A consists of an irregular talc-phlogopite aggregate developing on a granular quartz-rich matrix (Plates 17 and 18). Totally limonitized, post-tectonic carbonate porphyroblasts overprint the talc-phlogopite-quartz groundmass.

Sections MWE 236 and MWE 245A illustrate the transition from "ultramafic" to guartzite exhalite facies. In MWE 236 the nematoblastic actinolite schist interfingers with a baryte-bearing quartzitic facies (Plate 19). The coarse baryte grains are intergrown with guartz and are in normal contact with the actinolite crystals. No evidence of disequilibrium between the two minerals is observed. In MWE 245A a foliated talc-actinolite-carbonate schist passes to a gossanous guartz-actinolite-baryte-rich facies. The transition occurs by means of rapid decrease in talc and concurrent increase in amphiboles and quartz. The quartz-actinolite-baryte facies is shows a grano-nematoblastic texture. The distinctly banded and coarse-grained actinolite bands are gossanous and are associated with abundant coarse-grained baryte aggregates (Plate 20).

The carbonate-rich rock, initially identified as a marble associated with the exhalative layer (see section 3, Part B), may in fact not be part of it. The <u>marble-like rock type</u> outcropping at the grid

reference 17/+3.5 (Fig. 21) is interpreted as the result of a hydrothermal pervasive carbonatic alteration replacing a barytebearing quartzite facies. From field observations, the replaced muscovite-bearing quartzite facies is probably part of the nearby outcropping magnetite quartzite. Calcite forms a granular aggregate, whose irregular texture is not comparable with that characterizing true marbles (Spry, 1969; Bard, 1980). There is no sign of deformation in the carbonates. Where the replacement is incipient, a sieve-like texture of the carbonate patches is noted. Amoeboidal baryte grains are associated with quartz in the groundmass. At advanced stages of replacement both baryte and quartz blebs are isolated in the carbonate groundmass.

3.2.3 Comparison between the ultramafic schists at Matchless West Extension and the Alpine-type Damaran ultramafic rocks

Thin sections cut from the ultramafic body outcropping near Otjihase Mine (see sect. 5.3 Part B) were also examined.

These rocks are represented by coarse-grained actinolite-chlorite schists (OTJ 684), talc-chlorite-actinolite schists (OTJ 685) and magnetite-bearing talc-chlorite-carbonate schists (OTJ 686A-B). The fabric varies from massive to coarsely foliated.

Mixed talc and chlorite form a fine-grained felt-like groundmass (Plate 22). This groundmass hosts isolated or clustered actinolite porphyroblasts and disseminated magnetite grains. Rhombic cavities are remainders of carbonate porphyroblasts. In the foliated actinolite schists in OTJ 684 euhedral actinolite porphyroblasts are included in a coarse, rutile-bearing fibro-lamellar chlorite matrix. Chlorite laths are grown both in parallel and in fan-like aggregates in the intergranular spaces. Comb-like texture of the chlorite aggregates is often seen between close actinolite crystals.

The mineralogy of these Damaran ultramafic rocks and that of the ultramafic schists of assumed exhalative origin are very similar in their main components (i.e., talc and actinolite). However, some

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differences can be observed, namely:

a) in the serpentinites quartz is basically absent, whereas it can be an important constituent in some of the exhalative schists examined (e.g., the quartz-rich groundmass in the talc-phlogopite schist in MWE 246A);

b) the unusual association of actinolite - quartz - baryte in the exhalative schists;

 c) the absence of magnetite, a common product of serpentinization of igneous ultramafics, in the exhalative schists;

 d) the layered distribution of the different facies of the ultramafic schists at Matchless West Extension (e.g., talc-rich, actinolite-rich) is probably a primary depositional feature;

e) the interfingering of the exhalative ultramafic schists with mineralized quartzite facies is observed even at the scale of the thin section.

The association of talc flakes with the quartz-rich massive gossan proximal to the alteration zone can be interpreted as an indication of the genetic link of talc, or premetamorphic Mg-rich precursors (sepiolite or palygorskite; McLeod and Stanton, 1984), with the oreforming hydrothermal system. Actinolite is observed as a primary mineral associated with mineralization in the present-day Atlantis II Deep in the Red Sea (Zierenberg and Shanks, 1983, 1988). The mineral is considered as an indirect input of volcanism contemporary with the mineralizing event.

3.3 The Ongombo sulphide body

A detailed account of the "ore zone" and enclosing schists from drillcore is given below; only limited work has been carried out on gossan material.

3.3.1 The amphibole-rich facies of the Ongombo Gossan

Attention has been focussed on the portion of the outcrop characterized by the discordant gossanous amphibole-rich patches. These features were not observed in the cores (see Sect. 2.3.3 Part B).

In the gossan exposure, the magnetite quartzite proximal to the amphibole-rich patches (OG 192A) is pockmarked by weathered sulphide aggregates and by discontinuous amphibole-rich gossanous bands. These features are very similar to those observed at Ongeama. Flaky talc aggregates are disseminated both in the amphibole-rich patches and in the quartz matrix.

The discordant amphibole-rich pockets (OG 192B, 196, 281) are preserved only as fine-grained limonite replicas of the fibrous amphibole aggregates. By analogy with what was observed at Ongeama, it is believed that the amphibole in this facies was probably anthophyllite. Large massive black and colloform exotic limonite patches possibly correspond to weathered sulphides. Coarse poikilitic, prismatic plagioclase crystals form radiating aggregates within the gossanous portion in OG 281. They contain the acicular traces of dissolved amphibole inclusions. These plagioclase aggregates may be equivalent to the zones of plagioclase enrichment observed within the ore zone in the core OGB 153A (see below).

3.3.2 The "ore zone" and the enclosing schists in drill cores

A series of 40 thin sections has been obtained from the ore-hosting sequence in the OGB 153A borehole (Eastern Shoot). The total length of core portion examined is 12,5 m. A schematic log is reported in Appendix E.

The aims of this detailed examination are :

a) to verify the existence of an alteration-related mineral association in the h/w schists in approaching the "ore zone";

b) to discriminate between ore-related alteration, if any, and contact metasomatism in the schists enclosing the metasills;

c) to distinguish the different styles of mineralization within the "ore zone".

The <u>h/w schist sequence</u> consists of alternating metapelitic and metapsammitic schist units and includes a metamorphosed gabbroic sill. The main mineral constituents of the metasediments are quartz, plagioclase (oligoclase), biotite and chlorite. Accessory epidote, apatite, tabular ilmenite and subordinate disseminated sulphides are constantly present. Abundant zircon imparts a mottled appearance to the host biotite. The occurrence of muscovite, staurolite and garnet is restricted to mica-rich layers. Plagioclase can be fairly abundant and occurs either finely intergrown with quartz or in coarser flattened poikiloblastic lenses associated with the mica layers. Sieve-textured amphibole, epidote and minor calcite distinguish a calc-silicate-bearing layer.

The distribution of chlorite varies from one layer to the other, but considerable chlorite enrichment is observed towards the margins of the metamorphosed sill intrusion. Part of the chlorite appears to be a product of retrograde metamorphism.

Chlorite, however, is not commonly found in the plagioclase-rich schists between the "ore zone" and the mafic sill. These schists contain abundant biotite and muscovite, associated with staurolite and garnet. Tourmaline can be relatively common in the muscovite aggregates rimming staurolite grains.

There is no outstanding magnetite dissemination in approaching the "ore zone". Magnetite does occur, with disseminated sulphides, in the chloritized schists and "calc-silicates" in the structural h/w of the mafic unit (OG 710A-B) and in the vicinity of a thin amphiboleenriched layer of probable volcanic origin (OG 706).

A comparison between the h/w schists and unaltered schists away from the "ore zone" and those in the f/w, highlights the few differences between these three groups of schists. The only variations (e.g. chloritization, magnetite dissemination), of possible hydrothermal origin, are confined to the aureoles of the intrusive mafic units.

An important feature that distinguishes the schists within the <u>"ore</u> <u>zone"</u> from those of the h/w and f/w, is the colour of biotite. This biotite displays a marked colour change from dark greenish (Plate 23) to reddish-brown in the sharp contact from either h/w and f/w to the "ore zone", thereby defining a selvedge zone approximately 1 cm-thick (Plate 24). This reddish-brown biotite also constitutes a major component along the entire vertical development of the "ore zone" (sections OG 718-751). The only exception is represented by a minor intercalation of unaltered quartz-biotite schist (OG 731), containing dark brown-greenish biotite. This intercalation may result from tectonic interleaving.

Quartz, biotite, plagioclase (oligoclase) and staurolite are the most abundant silicate minerals within the "ore zone". Chlorite is scarce, and it becomes significant only in the lowest part of the "ore zone". Accessory fine- to coarse-grained apatite is present in remarkable amounts. Tabular ilmenite is disseminated along the schistosity planes.

The sharp contact from the unaltered schists to the "ore zone" is mineralogically defined by the development of the coarse biotite-rich selvedge zone, mentioned earlier. This selvedge zone is characterized by compact biotite bands with disseminated coarse- to fine-grained staurolite, coarse apatite and poikiloblastic flattened sulphide grains (pyrite "augen") (Plate 24).

The "ore zone" is approximately 6 m-thick. Below the biotite selvedge zone, the "ore zone" can be ideally subdivided in two sections, each defined by a magnetite quartzite band. The first section, 1.5 m-thick, is characterized by coarse plagioclase-rich, staurolite-bearing biotite quartz schists. The schists contain densely disseminated sulphide-rich centimetre-sized quartz lenses. These lenses thicken downwards. Sulphides are only subordinately disseminated in the schists. Such arrangement is suggestive of a deformed stringer system passing to a zone of poorly developed semimassive mineralization. The first section ends with a 40 cm-thick magnetite quartzite layer, in which the mineralization is recrystallized in semimassive bands. Sulphides are partially remobilized along metamorphogenic quartz veins rimmed by coarse plagioclase and chlorite.

In the second portion of the "ore zone", up to 20 cm-thick chloritebearing quartzite layers contain well developed coarse semimassive sulphide mineralization. A second, poorly mineralized magnetite quartzite occurs at the structural f/w of the "ore zone" (OG 743, m 367.83-368.29). The semimassive-ore facies represents the main style of mineralization in this second section of the "ore zone". A remarkable feature of the second section is the increase in the plagioclase content within schist bands included between richly mineralized quartzite horizons (e.g., OG 732 and 735). In these schist layers deformed biotite aggregates are included in a coarse-grained poikiloblastic plagioclase groundmass (Plate 25). The plagioclase (oligoclase) is zoned, locally twinned and often sericitized. The abundance of plagioclase in these layers may be a primary feature. It may reflect the metamorphic recrystallization of original authigenic albite(?)-rich sedimentary layers. The plagioclase enrichment in the sediments might have been induced by the ore-related hydrothermal activity.

The transition from the lower magnetite quartzite horizon (OG 749) to the f/w schist package is marked by the occurrence of a moderately chloritized biotite selvedge zone carrying flattened sulphide lenses. An unusual pod-like, coarse-grained poikilitic plagioclase aggregate marks the lower contact of the "ore zone" (OG 752). It contains coarse to very coarse euhedral apatite crystals (up to 6 mm in diameter) rimmed by reddish biotite and muscovite laths. In the hand specimen the color of the apatite is pale green.

The passage to the dark brown biotite-bearing f/w schists is sharp.

3.3.3 The sulphide mineralization

The sulphide mineralization from core OGB 153A (Eastern Shoot) is characterized by the following styles, namely:

- a) coarse disseminated sulphides in biotite schist (OG 720B, 744, 750);
- b) semimassive ore in magnetite-devoid quartz-rich bands (OG 718, 722, 729, 733, 735, 737, 741);
- c) disseminated in magnetite quartzite (OG 724, 747, 748);
- d) partly remobilized ore along S₂-syntectonic quartz veins (OG 723, 726).

The ore assemblages characterizing these styles of mineralization are the following:

 in biotite schists, dominant pyrite with subordinate chalcopyrite and pyrrhotite as sulphide species; ubiquitous ilmenite and occasional magnetite as oxide species;

2) in the semimassive ore, pyrite, chalcopyrite and pyrrhotite are the main components, with subordinate sphalerite. Only chalcopyrite, pyrrhotite and sphalerite occur in OG 729. Magnetite is accessory in OG 718 and 733. Galena is in minimal amounts in OG 718 and 741;

3) in magnetite quartzite, magnetite is accompanied by variable amounts of chalcopyrite and pyrite, subordinate pyrrhotite and accessory galena; sulphides are concentrated in bands;

4) in the remobilized ore, chalcopyrite, pyrite, pyrrhotite and minor

sphalerite are the sole components.

A characteristic of the pyrite in this mineralization is a slight but perceivable anisotropy. Two main generations of pyrite can be distinguished: the first generation is represented by early porphyroblasts deformed into "augen" (Plate 24) and by recrystallized aggregates; the later generation includes undeformed cubic porphyroblasts.

Magnetite, chalcopyrite, pyrrhotite and galena display the usual optical characters. Marmatitic sphalerite occurs as typical tiny starshaped and drop-like exsolutions in chalcopyrite, whereas globular chalcopyrite blebs are exsolved in the rare coarser sphalerite patches. Star-shaped sphalerite exsolutions are typical of hightemperature chalcopyrite (Ramdohr, 1980). Their presence may be related to the high temperatures of recrystallization undergone by the sulphide mineralization during lower amphibolite facies metamorphism. Pyrrhotite is mostly associated with chalcopyrite, where it occurs as small amoeboidal blebs and anhedral fractured patches. Tabular ilmenite crystals display characteristic lamellar hematite exolutions. The crystals occur parallel to the S₂ foliation of the hosting schists and are locally clustered and recrystallized in the hinges of microfolds. Ilmenite locally contains elongated chalcopyrite inclusions incorporated during growth.

Pyrite and magnetite crystals commonly show inclusions of ore and gangue minerals: chalcopyrite, pyrrhotite and galena, and quartz, apatite and mica, respectively (Plate 26). Galena is only observed in inclusions, alone or associated with chalcopyrite (Plate 27).

Regional dynamo-thermal metamorphism, in the form of temperature increase and directed pressure, is responsible for an overall coarsening of the grain size, recrystallization and reorganization of the initial paragenesis through deformation. This is reflected in the general texture of the mineralized layers. These layers are, in fact, characterized by the development of coarse-grained idioblastic to flattened pyrite grains (Plate 26) and annealed aggregates (Plate 27) over a chalcopyrite- and pyrrhotite-rich recrystallized matrix (Plate

i.

30). A similar texture is seen where magnetite is abundant, although magnetite does not show evidence of plastic deformation as pyrite does (see below).

The chalcopyrite-pyrrhotite matrix around the pyrite augen is intensely strained; this is clearly visible in the pyrrhotite, which shows a noticeable wavy extinction.

Experimental studies on sulphide deformation with increasing temperature and confining pressure (Kelly and Clark, 1975; Cox et al., 1981; McClay and Ellis, 1983) show (Fig. 28):

a) a change from brittle to moderately ductile rheological behavior in pyrite in metamorphic grades higher than greenschist facies and at low strain rates;

b) the overall greater ductility of chalcopyrite and sphalerite than pyrite in the different geological conditions;

c) the rapid strength loss of pyrrhotite with increasing temperature.

The formation of pyrite "augen" is thereby referred to a pressure solution strain mechanism active during D_2 . The ductility contrast among sulphides results in the inclusion of harder pyrite grains in a matrix of softer ore minerals. The resulting texture does not reflect the original relationship among the components. Grain growth and annealing must have occurred between the D_2 and D_3 phases. A late, brittle deformation manifested in fractured pyrite porphyroblasts is referred to the D_3 deformation phase by Klemd et al. (1987). The open fractures in pyrite are filled by softer matrix material, i.e. pyrrhotite and chalcopyrite (Plate 28).

In the biotite schists, other possible effects of deformation on pyrite may also be seen in the recrystallization of fine-grained, thin pyrite lenses ("ribbon pyrite" in Klemd et al., 1987) arranged in *enechelon* patterns within the deformed micaceous laminae. The ribbon pyrite develops along the borders of subhedral pyrite grains, on the side of the grains parallel to the foliation. These structures are



Fig. 28 - Comparison of strengths of pyrite, chalcopyrite, pyrrhotite, sphalerite and galena at a constant 1 kb confining pressure (Cox et al., 1981).

confined to limited portions of the schists, and probably indicate the action of slow plastic deformation (creep) induced by shear stress. Coarsely recrystallized pyrite and magnetite generally show a strong tendency to idiomorphism and to development of crystal faces. Where both pyrite and magnetite occur in remarkable amounts, as in the magnetite quartzite in OG 747, the contemporaneous recrystallization of the two minerals leads to the formation of interference textures, with mutual inclusion and development of atoll-like, embayed aggregates (Plate 26).

4. DISCUSSION

Petrological studies of country rocks and of the alteration zones have provided evidence of important mineralogical changes, which, at least in some cases, are confirmed by field observations. The scheme in Fig. 29 reports the main mineralogical characters observed in the alteration zones.

In spite of the postulated common genetic mechanism for these sulphide bodies (see Part A) and of the relatively short distance between them, the mineralogy of the alteration zones is variable. For example, at Ongeama the alteration pipe consists of a chlorite core and a sericitic envelope, whereas at Matchless West Extension the pipe consists of a quartz-sericitic core and a poorly developed chloritic margin (see also Sect. 3 Part B). Nevertheless, the chemical environment inferred from the alteration mineral assemblages is essentially similar. This appears to be especially true for Ongeama and Matchless West Extension, where the alteration pipes are better defined both in shape and relationship with the mineralization.

The occurrence of chloritoid, staurolite, garnet, magnetite and chlorite at Ongeama, and the dense pyrite dissemination in the sericitic quartzite at Matchless West Extension, reflect conditions of high Fe/Mg ratio, high Al and low alkali (Na, Ca and K) along the axis of the alteration pipe. Although devoid of plagioclase, both alteration zones are highly enriched in quartz, due to extensive



Fig. 29 - Scheme of the mineralogical zonation within the alteration zones at Ongeama and Matchless West Extension. The zonation is referred to a generalized scheme of zoned alteration pipe. The minerals are listed according to decreasing abundance.



Fig. 30 - Theoretical stability fields for the systems Fe_30_4 -Ti0₂-FeTi0₃-Fe₂O₃ as function of oxygen pressure and between 400° and 600°C (Stanton, 1972).

At Matchless West Extension the marginal zone of phlogopite-bearing muscovite-chlorite schists may correspond to an overlap between the sericite zone and a marginal Mg-enriched zone. The occurrence of sporadic phlogopite may be also interpreted as marking a late and marginal domain of potassic alteration. According to Costa et al. (1983), this limited potassic alteration would be induced by late hydrolysis of mica in the outer regions of the mineralizing system, hence supply of K to the fluids, and breakdown of alteration-related chlorite.

The presence of anthophyllite and talc in the ore proximal to the assumed discharge vent at Ongeama and Matchless West Extension, respectively, is suggestive of major Mg enrichment external to the alteration pipe. Mg, however, is further enriched in the talc- and actinolite-bearing exhalites at Matchless West Extension. Mg enrichment in positions proximal to the discharge vent is reported for both ancient and present-day sub-seafloor deposits (e.g., Guaymas Basin) (Costa et al., 1983; Aggarwal and Nesbitt, 1984; Lonsdale et al., 1980). By analogy with what was observed at Ongeama and Matchless West Extension, the anthophyllite and talc enrichment of the gossanous central-western portion of the Central Shoot (Ongombo) could be considered as indicating proximity to a main discharge site. The absence of these minerals in the Eastern Shoot mineralization may be, on the other hand, interpreted as a "distal" feature.

Sulphide disseminations and lenses, which, on weathering, impart the gossanous aspect to the schists within the alteration pipes, may be interpreted as the result of deformation and metamorphism of a "stringer"-type mineralization. "Stringer" mineralization was formed within the fracture network, which acted as fluid channelways, along the alteration pipe.

In spite of the higher metamorphic grade, the mineralogy of the alteration zone at Gorob, as reported by Haussinger and Okrusch (1990) seems to reflect the same trends of major element variations as those

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inferred for Ongeama and Matchless West Extension.

At Ongombo, chloritization and magnetite and sulphide dissemination in the schists above the ore zone, although irregularly distributed, are spatially associated with the mafic sills. This suggests that the enrichment in chlorite, magnetite and sulphides is probably the result of contact metasomatism induced by the intrusion of basaltic sills into wet unconsolidated sediments, and not to ore deposition. It is, therefore, concluded that there are at least two separate events: one related to contact metasomatism and the other to the action of oreforming fluids. This separation is reflected also by the trace-element distribution in the ore-hosting sequence (see Sect. 2.3.2 Part D).

Klemd et al. (1987, 1989) attribute similar chlorite- and magnetiteenriched schists in the h/w of the ore zone at Matchless Mine to the effects of an ore fluid system. Their interpretation, however, is possibly incorrect, if the observations above are taken into account.

Studies of the drill core from Ongombo suggest that the change in colour of biotite, among other factors, is considered a fundamental indicator of the interaction of the ore-bearing fluids and the schists. The abrupt colour change of biotite both in the h/w and f/w of the "ore zone", and the deformed nature of the "smeared" pyrite augen disseminations within the marginal biotite "selvedge zones", are suggestive of tectonic deformation. The strain would be largely concentrated along the margins of the sulphide-bearing horizon.

Assumed remnants of sheared f/w stringer zones are observed also in ancient deformed mineral deposits similar to those associated with the Matchless Belt (e.g., Hoy et al., 1984).

The mineralogical and textural features of the Ongombo ore zone indicate a fairly well defined vertical lithological zonation. The reconstruction of this vertical zonation, after eliminating the effects of deformation, is as follows:

a) h/w "stringer"-feeder zone;

- b) siliceous ore zone containing semimassive sulphides in lenses and layers;
- c) f/w sulphide-bearing magnetite quartzite.

The inverted polarity of the sequence given above is confirmed by the inverted zonation observed in the mafic sills, in which ultramafic differentiates occur in the h/w.

The gangue mineralogy of the Ongombo "ore zone" is somewhat different from that of Ongeama and Matchless West Extension. At Ongombo the gangue is characterized by the almost total lack of chlorite, abundance of zircon-bearing biotite and staurolite, less pronounced silicification, and unusual abundances of plagioclase and of apatite. Phosphorus enrichment is also reported by Haussinger and Okrusch (1990) at Gorob. The presence of plagioclase, however, does not appear to be restricted to the Eastern Shoot, as the mineral also occurs in the anthophyllite-rich zones of the western Central Shoot (see Sect. 3.3.1 above). It is worth noting that in studies of iron formations in the Southern Marginal Zone of the Damara Orogen, Breitkopf and Maiden (1986) and Breitkopf (1988) have found high abundances of apatite, plagioclase and amphibole. These authors consider these minerals as evidence for the volcano-exhalative origin of the mineralization.

Unusual mineralogical associations, similar to those described in this study for the "ore zone" at Ongombo, were also noted by Gair and Slack (1982), in relation to the Appalachian Besshi-type sulphide deposits of the Gossan Lead District (Blue Ridge Belt, Virginia, USA). The two authors report, in fact, the occurrence of almost monomineralic biotite schists, with high content of fluoro-apatite and zircons, and of plagioclase-rich rocks alternating with bands of massive mineralization. The origin of these rocks is attributed by the authors to the action of ore-related metasomatizing fluids.

At Ongombo, the absolute prevalence of biotite and staurolite over chlorite might be simply a result of metamorphic conditions. The abundance of highly degradable minerals within the "ore zone" is possibly the reason of the lack of surface expression of lithofacies other than the resistant magnetite quartzite.

It is clear from the petrological work carried out that mineralogical changes are important in the attempt to define an alteration zone. These changes can be discerned in spite of the obliterating effects of regional metamorphism and deformation.

Magnetite disseminations (e.g., in the "I" anomaly area north of the Ongeama gossan and in the Ongombo cores), unless associated with other important mineralogical changes in the host rocks, are not to be considered "a priori" as pathfinders to the mineralization.

During metamorphism, oxides can undergo pronounced chemical and less notable physical modifications. The latter occur by simple coarsening and crystal face development (Stanton, 1972). Chemical modifications during metamorphism largely depend on f_{02} (Fig 30). Therefore, the occurrence of magnetite in barren metapsammitic horizons of the Kuiseb Formation can be interpreted as:

a) recrystallization of detrital magnetite grains;

b) metamorphism of moderately hematite-rich metapsammitic horizons.

Magnetite occurs as disseminations also in "calc-silicate" layers within the metapsammitic horizons. The "calc-silicate"-bearing metapsammites in the Kuiseb Formation appear to be correlated with the north-westerly calcareous Tinkas turbidites, developed from the Karibib carbonate platform (Porada and Wittig, 1983; see Sect. 1 Part A). In view of this possible connection between the metapsammites in the Kuiseb Formation and the Tinkas turbidites, the presence of magnetite disseminations in the former may be interpreted as the result of metamorphic breakdown of siderite (James, 1955), or may represent an indirect, detrital contribution of the mafic volcanism characterizing the the north-westerly Karibib platform (Miller, 1983b). The preservation of alteration zones, well defined in shape and mineralogy, in spite of deformation and metamorphism, confirms Stanton's (1982) opinion that stratiform ores and their envelopes are striking examples of compositional inhomogeneity that metamorphism has not been able to homogenize. This appears to be the result of the restriction of metamorphic diffusion to minute distances and of the consequent lack of attainment of metamorphic equilibrium even in very small domains (e.g., single grain).

The alteration zones studied contain contrasting greenschist- and amphibolite-facies mineral assemblages (e.g., chlorite-staurolite at Ongeama), or display mineral associations of apparent slightly higher grade than the enclosing schists (e.g., the chlorite-poor biotitestaurolite schists in the "ore zone" at Ongombo). In Stanton's view, these restricted anomalies can be attributed to primary precursor patterns, depending on the interplay between hydrothermal fluids, seawater and sediments along the channelways, rather than to localized variations in metamorphic grade.

PART D

GEOCHEMISTRY OF MINERALIZATION, ALTERATION ZONES AND COUNTRY ROCKS

1. INTRODUCTION

The 487 samples collected during the field work were submitted to chemical analysis.

A suite of 22 elements was selected for analysis: Cu, Pb, Zn, Co, Ni, Ag, Au, Mn, Cr, V, Ti, Ba, Mo, Zr, Nb, Y, Rb, Sr, Sn, W, As and Bi. Concentrations of the first nine elements, from Cu to Cr, were determined by means of atomic absorption spectrometry (AAS), and the balance by X-ray fluorescence (XRF) analysis. These analyses were carried out at Scientific Services Ltd. in Windhoek and Cape Town. Results and detection limits are given in Appendix E, Tables 1 to 18. Selected samples of altered and unaltered schists and ultramafic rocks were analyzed for major elements by XRF analysis. Fusion discs and powder pellets were analyzed by means of a Philips 1410 XRF spectrometer at Rhodes University (Table 9). A number of selected elements was used to construct element distribution profiles, contour maps and diagrams.

Anomalous major- and trace-element distribution characterizes present-day hydrothermal mineralizing systems in the oceanic environment and well-studied ancient equivalents. Therefore, one of the aims of this study is to verify whether similar distribution patterns exist or are still preserved in the metamorphosed and deformed deposits at Ongeama, Matchless West Extension and Ongombo.

2 GEOCHEMISTRY OF SULPHIDE BODIES AND ASSOCIATED ALTERATION ZONES

2.1 Geochemical trends at Ongeama: trace element distribution

The geochemical data from the weathered mineralization and associated alteration zone at Ongeama (table 2, Appendix E) were used to construct element distribution profiles. In the evaluation of each profile, the average content of a given element in barren schists has been taken into account, and shown as a reference line.

Appendices F and G include the geochemical profiles along strike for the minor and major gossan band, respectively. The different rock types are indicated by symbols. Each rock type is characterized by a distinct range of element concentration. However, within the same rock type, variations can occur in space and can represent, in some cases, a well defined zonation.

The two gossan bands are considered separately because of their differences in lithological complexity and geochemical character.

The minor gossan outcrop consists of three rock types: massive gossan, magnetite quartzite and subordinate non-magnetic ferruginous muscovite schists (see Part B for details).

In general, the massive gossan in this outcrop appears to be relatively more enriched in Cu (up to 0.73 %), subordinately in Mn and Ba and Ti, and depleted in Pb, Zn, Ag, V, Rb, Sr, Zr, Sn and As.

Magnetite quartzite is enriched in Pb (up to 0.19 %) and, to a lesser extent, in Zn (up to 471 ppm), Mn (1164 ppm) and Sn (110 ppm). Although the massive gossan locally contains high Au values (on average 455 ppb Au), the average Au content in the magnetite quartzite is higher (698 ppb Au). Magnetite quartzite is slightly enriched in Ag relative to the other two rock types.

The ferruginous schists show the highest Ba (up to 503 ppm), Ti (up to 0.84 %), V (up to 224 ppm) and Zr (up to 148 ppm) contents, and moderate enrichment in Zn, Pb and Sn. They are depleted in the other elements.

Lateral variations of the most significant elements (i.e. Cu, Zn, Pb, Ag, Au, Ba and Mn) are observed. Detailed profiles for all the elements considered are contained in Appendix G-a. A general increase towards the western end of the outcrop is shown by most of the elements considered (Fig. 31a-e).

In particular:

- The <u>Cu</u> distribution outlines a clear zonation with westward increase within the massive gossan. This trend is not shown in the adjacent magnetite quartzite. The Cu increase in the massive gossan



Fig. 31a-e - Ongeama Prospect, minor gossan outcrop: geochemical profiles for Cu, Zn, Pb, Au and Ba along strike.

outcrop correlates with its thickening. It may be either a primary feature or it may be related to deformation. However, a similar distribution for Ti, which is typically considered as an immobile element (Finlow-Bates and Stumpfl, 1981), would suggest that it is a primary feature.

- <u>Zn</u> displays a general increase westward, from massive gossan to magnetite quartzite, although within each facies the values are irregularly distributed.

- The <u>Pb</u> distribution shows a prominent, isolated peak (0.19 % Pb) next to the boundary between magnetite quartzite and massive gossan and proximal to the Cu-enriched portion of the outcrop. This isolated peak notwithstanding, a general westward increase is observed.

- <u>Au</u> exhibits a gross westward increase within the massive gossan. In the latter, Au enrichment appears to coincide with the thickening of this horizon. In the magnetite quartzite the average content is slightly higher (see above) but with no apparent regular distribution trend. In the massive gossan isolated peaks may correspond to late remobilization of mineralization in veins.

- <u>Ba</u> is clearly enriched in the centre of the massive gossan outcrop next to the boundary with magnetite quartzite. The Ba profiles for these two rock types are similar. <u>Sr</u> peaks (not shown in the figure) coincide with Ba highs.

It is worthy to note that, in general, the small massive gossan lens situated at the western end of the magnetite quartzite band displays the highest element concentrations (see profiles in Appendix F).

For the major gossan outcrop (profiles in Appendix G), the five rock types considered in this section are: siliceous gossan (MS), amphibole-rich semimassive gossan (AS), magnetite quartzite (MQ), nonmagnetic ferruginous schists (FS), magnetic muscovite schists (MM) and magnetic chlorite schists (CS) (see section 2.2.2 Part B for details). MS, AS and MQ ("ore zone") are enriched in Cu, Zn, Pb, Ag, Au, Mn and, subordinately, Ti and Sn relative to the "alteration" rocks (FS, MM and CS), which are enriched in Ba and, subordinately, in Sr, Rb, As, Zr, Ti, V, Ni and Cr.

In the major gossan outcrop, the most important feature is the increasing trend of some elements towards the centre of the <u>sulphide</u> <u>body</u>, i.e. towards the portion of the outcrop characterized by the AS facies (Fig. 32a-h).

The main points of note are:

- The <u>Cu</u> distribution clearly reflects this trend of progressive centreward increase, with the highest value in the AS facies, close to the intersection with the assumed alteration pipe.

- The <u>Au</u> distribution is characterized by numerous, isolated peaks (up to 4870 ppb). The position of these peaks shows an apparent increase towards the periphery, and depletion in the central Cu-rich portion. However, this trend may be more apparent than real. It is reasoned that, if the lateral peaks are ignored, as they may be related to local remobilization, then a normal centreward increase is observed.

- Zn is highest in the MS portion surrounding the AS core, whereas a relatively flat profile characterizes the marginal portion of the outcrop.

- \underline{Mn} shows a flat profile in the E, but peaks towards the centre: the major peaks correspond to the thickening of MS and to the occurrence of AS in that position. A similar pattern is shown in MQ. Overall higher Mn contents are recorded in the western portion of the MS outcrop.

- The <u>Pb</u> profile in the MS is relatively flat, with isolated peaks. A slight enrichment occurs in AS and persists towards W. The Pb concentration is highest in MQ. In this facies Pb shows a prominent enrichment next to the alteration pipe (up to 0.22 %).

- <u>Ti</u> shows a distribution pattern with an isolated peak in the AS facies with relative enrichment in MS towards the margins.

- the <u>Ba</u> and <u>V</u> distribution exhibits a limited enrichment within the AS facies.

- <u>As</u> is enriched in the central MS portion but it also shows an increase to the E both in MS and MQ facies. <u>Sn</u> (not shown in Fig. 32) exhibits a countinuous profile only in MQ.

In the profiles shown in Appendix G,elements which are preferentially concentrated in the <u>"alteration" rock types</u> (CS, MM, FS) (i.e. Ba, Sr, Zr, As, Rb, Ti, V, Ni and Cr) suggest a lateral zonation with increasing values from the periphery towards the axial zone. Zonation along the pipe is further illustrated in section 2.3 below).

The following trends are noteworthy:

- <u>Ba</u> is highly concentrated in the MM facies and progressively increases towards the chloritic pipe, in which the Ba distribution is more irregular (Fig. 32g). The affinity of <u>Sr</u> and <u>Ba</u> is reflected in their similar distribution trends.

- <u>Zr</u> is highest in the chloritic pipe and slightly depleted in MMS and FS towards the margins.

- <u>As</u> (Fig. 32h) and <u>Bi</u> are exclusively concentrated along the axial zone.

- the flat distribution profile for $\underline{\text{Ti}}$ (Fig. 32f): higher Ti contents occur in the eastern portion of the outcrop. In proximity to the pipe a decrease of the Ti content in MM and CS is coincident with the Ti peak in the AS facies.



Fig. 32a-h - Ongeama Prospect, major gossan outcrop: geochemical profiles for Cu, Au, Zn, Mn, Pb, Ti, Ba and As along strike.

e)

Fig. 32



Fig. 32 - See previous page.

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- <u>Cr</u> and <u>Ni</u> are slightly enriched in the MM and CS of the axial zone; Cr appears to increase in the CS closer to the intersection with the mineralized horizon.

A comparison between the schists within the alteration pipe and the unaltered country rocks highlights different distribution trends of the elements.

The altered schists are remarkably enriched in Cu, Au and Ba, and moderately in Sn, As, Bi, Cr and Co relative to the country rocks. The schists in the pipe are also strongly depleted in Rb, and, to a lesser extent, in Sr and Ni. The Zr, Ti, Zn, Pb, Ag concentrations in the country rocks compare fairly well with those in the alteration pipe.

Tectonic disturbances (e.g. faults, quartz veins) in the outcrop are outlined by the coincidence of either highs or lows in the trace element distribution. For example, in the eastern sector of the outcrop, probable late ore remobilization is suggested by coincident Au and Ag peaks in both MS and MQ facies. In the western sector (grid ref. 10060 E in Plan 1 Appendix A), the prominent fault, causing the drag-like deformation in the gossan, is indicated by enrichment of Zr in MS and Cr in MM, and by depletion for Ba in MM and Zn and Pb in MS.

In comparing the trace element distribution patterns between the two distinct gossan bands, several differences exist. It is noted that Zn, Au, Ag, Mn and Sr show opposing concentration in the MQ and MS facies of the two gossan bands. In the minor gossan outcrop these elements are enriched in the MQ, whereas in the major outcrop they are enriched in the MS facies.

2.2 Geochemical trends at Matchless West Extension: trace element distribution

The trace element distribution trends for the Matchless West Extension mineralization are represented as contour maps in Appendix H. Threshold values for the contour maps were obtained by means of construction of histograms. Elements were grouped as follows:

- a) Cu + Zn + Pb + Ba + Mn representing a volcano-exhalative association;
- b) Ni + Co representing an association in sulphide deposits related to mafic volcanism;
- c) Ti + V + Cr representing elements related to mafic magmatism;
- d) Au + Ag;
- e) Sn + W + Mo + Bi, elements of crustal affinity,
- f) Rb, Sr and Zr, lithophile elements.

In this section the six rock types considered are: gossanous pyriterich muscovite-quartz schist of the axial zone of the alteration pipe (QMP schist), marginal muscovite-chlorite schist (MC schist), massive gossan (MS), magnetite quartzite (MQ), talc-actinolite schists (TA schists) and marble-like rock (MB).

The mineralized body is depleted in Ti, Zr and Rb. The concentrations of Ti, Zr, Au, Ag, Sn, W, Mo and Bi in the altered schists are comparable with those in the barren country rocks. Cu, Zn, Pb and Ba are enriched, whereas Mn, V, Ni, Co, Rb and Sr are slightly depleted, in the alteration zone relative to barren country rocks (see Tables 3 and 6, Appendix E).

Within the ore system, Cu, Zn, Pb, Ba and Au best show a well defined increasing trend from the alteration pipe to the overlying mineralized body (Fig. 33a-f).

The following features are observed:

- The main <u>Cu</u> enrichment is recorded proximal to the alteration pipe, with MS containing up to 8.49% Cu. Notable "distal" Cu enrichment occurs in the TA schists (up to 3.56% Cu). In the MQ Cu does not exceed 0.7%.

- Zn and <u>Pb</u> are enriched in a more distal position compared with the Cu distribution pattern. However, a remarkable, "proximal" Zn increase is recorded in the eastern portion of the MS outcrop, where Zn attains concentrations of up to 0.30%. High Zn values occur in the TA schists (up to 0.87% Zn) and, to a lesser extent, in the enclosing MQ units. A Zn peak marks the MB outcrop. The Pb distribution pattern shows a more gradual enrichment compared with Zn towards the margins of the mineralized body. MS to the east and MQ above the alteration pipe show isolated Pb peaks (up to 0.37% Pb in MQ). In the western, distal portion of the outcrop the highest Pb contents are recorded in MQ and, subordinately, in TA. Within the upper MQ layer, the f/w, amphibole-richer, and the h/w, amphibole-poorer, portions of the layer, show a drop in the Pb content from the base to the top (from 1.54% to 315 ppm Pb).

- The degree of <u>Ba</u> enrichment in the Matchless West Extension gossan is noteworthy if compared with Ongeama. Lower Ba contents are recorded in the f/w QMP schists and in the eastern portion of the MS outcrop. The highest Ba enrichment is found within the central portion of the outcrop, in MS, in MQ (up to 5.4% and 4.42% Ba, respectively) and in ultramafic schists intercalated with quartzitic or phyllitic material (up to 7.24% Ba). Further Ba enrichment is observed in the distal MQ outcrop, with contents of up to 4.94% Ba in the lower MQ horizon. As expected, the Sr distribution reflects in part the Ba enrichment within the central and western sectors of the outcrop.

- A gradual <u>Au</u> increase occurs in the transition zone between alteration pipe and overlying massive gossan (up to 749 ppb Au). A minor, distal Au enrichment is irregularly shown in MQ and in associated gossanous TA schists.



Fig. 33a-f - Schematic contour maps illustrating the element distribution within the mineralized system at Matchless West Extension (reference sketch-map modified after Smalley, 1990).

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- <u>Mn</u> is significantly concentrated in the western distal sector of the outcrop and only enriched to a limited extent in a proximal position in MS. This element is irregularly distributed in MQ and TA schists (up to 1843 ppm Mn), but the highest peak is recorded in MB (2385 ppm Mn).

- \underline{Sn} , \underline{Mo} , \underline{Bi} and \underline{W} show well defined trends . Mo is mainly enriched proximal to the alteration pipe (up to 219 ppm Mo in MS), whereas Sn, W and Bi are concentrated in intermediate and distal positions. Peaks in the Bi concentration are observed in the actinolite schists and in MQ.

An account of the trace-element zonation recorded in the alteration pipe at Matchless West Extension is given in Section 2.3 below.

2.3 Major element distribution in the alteration pipes at Ongeama and Matchless West Extension

Tables 10, 11 and 12 in Appendix E compare the major-element contents of selected samples from the alteration pipes at Ongeama and Matchless West Extension with unaltered country rocks.

Attention is drawn to the following features:

- an outstanding Fe enrichment (on average 24% Fe₂O₃, but locally exceeding 60%) characterizes the schists in the alteration zones relative to unaltered schists. The Fe content appears to increase towards the margins of the pipes;

- Ca, Na and K are remarkably depleted, and locally below detection, in both alteration pipes;

- Al₂O₃ is relatively depleted in the alteration zones, in favour of an Fe_2O_3 and/or SiO₂ enrichment. However, the two alteration pipes display a contrasting distribution of these oxides. In the pipe at Ongeama the axial zone appears, in fact, to be more aluminous than the margins, whereas at Matchless West Extension the axis of the pipe is almost totally silicified (in excess of 90% SiO₂);

- at Ongeama, Mg is mostly immobile along the axis of the pipe, and considerably depleted at the margins. The element displays an opposite distribution at Matchless West Extension.

2.4 Geochemical zoning within the alteration pipes at Ongeama and Matchless West Extension

The alteration pipes associated with the Ongeama and Matchless West Extension sulphide deposits were subdivided into a number of sectors along their axial and lateral extensions (Fig 34 and 35).

The average content for each element was calculated in each sector (Tables 13 and 14; Appendix E). These averages were plotted against distance (expressed as progressive numbers of the sectors) in diagrams. The purpose was to verify the existence of possible trace element zonation within the alteration zones. The trends observed at Ongeama are illustrated in Fig. 36 and those of Matchless West Extension in Fig. 37.

At <u>Ongeama</u> (Fig. 36a-i) only one half of the muscovite-rich sheath (which is developed along the eastern portion of the ore body and the least tectonically disturbed) was considered. It is assumed that lateral variations in the element distribution are symmetric relative to the axial zone.

Most of the elements show regular variations within the alteration zone. The results can be summarized as follows:

- The distribution of Zn, Ba, Sr, Cr and Sn show an axial zonation , with values increasing from sector VII to sector IV. This zonation is remarkable for Ba and Zn and less pronounced for Sn and Sr. Outside the axial zone the distribution of these elements varies irregularly.

- <u>Cu</u>, <u>Co</u>, <u>Au</u> and <u>Rb</u> display a lateral zonation with values



Fig. 34 - Ongeama Prospect: subdivision of the outcrop of the alteration zone into sectors.



Fig. 35 - Matchless West Extension deposit: subdivision of the outcrop of the alteration zone into sectors (sketch-map modified after Smalley, 1990).

ONGEAMA





Fig. 36a-i - Ongeama Prospect: trace element zoning within the alteration pipe. Roman numbers along the X axis (distance) correspond to the subdivision in sectors of the alteration zone, as shown in Fig. 34. Element concentration for each sector is obtained by averaging the values of the samples included in the sector.







Fig. 36a-i - See previous page.

ONGEAMA





Fig. 36a-i - See previous pages.

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---- Cu --- Ba

Fig. 37a-e - Matchless West Extension deposit: trace element zoning within the alteration pipe. Roman numbers along the X axis (distance) correspond to the subdivision in sectors of the alteration zone, as reported in Fig. 35. Element concentration for each sector is obtained by averaging the values of the samples included in the sector.

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Fig. 37a-e - See previous page.



Fig. 38 - Ongeama Prospect: scheme of the position of the samples in the alteration zone. The surface geology of the central sector of the major gostan outcrop is schematically shown.

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Fig. 39a-b - Ongeama Prospect: zoning of Zn and Ba within the alteration pipe. Concentration values of all the samples along the pipe are plotted. Along the X axis, distance is considered from a 0 metre reference point, which corresponds to the intersection between the alteration pipe and the sulphide body.

increasing from the axial zone towards the margins, in the muscoviterich envelope. Along the axial zone the concentration of these elements is fairly constant, except for Rb, which displays variations within the same range of concentration.

- <u>Zr</u>, <u>As</u>, <u>Bi</u> and <u>Ni</u> exhibit a lateral zonation with concentrations increasing from the margins to the core. This trend is particularly well displayed by the Zr and As profiles. In the latter a remarkable difference in the As content distinguishes the chloritic pipe from the muscovite-bearing envelope.

Two profiles for the Zn and Ba distribution were obtained by plotting values of each sample (see Fig. 38 for sample location) instead of averages (Table 15, Appendix E). This procedure was used in order to test the validity of the method. The results (Fig. 39a-b) match the trends previously assessed for these elements by using averages.

At <u>Matchless West Extension</u> only Cu, Zn, Ba, Au, Zn and As define a clear zonation within the alteration pipe (Fig. 37a-e). In particular, Cu, Ba, Au, As and Zn define an axial zoning with upward and lateral increase, whereas Zr contents increase with depth along the pipe. The Ba distribution shows the most regular profile, whereas the Zn and Cu contents record an abrupt increase in the marginal part of the alteration zone, i.e. towards the west.

2.5 Geochemical trends at Ongombo

2.5.1 Trace element distribution in the gossan exposure

The profiles in Appendix I represent the trace element distribution along strike in the gossan exposure at Ongombo. The irregularity and the complexity of these profiles are due to uneven sample density.

The most clearly discernible trends are as follows (Fig. 40):

- important Cu, Zn, Mn, Ni and Co enrichment is confined to the

Central Shoot and to its western extension. This seems to be especially the case for the localized distribution of Ni and Co. In this sector of the outcrop, Co has concentrations exceeding 500 ppm. Bischoff et al. (1983) report similar high Co contents in the Galapagos Rift;

- Ba shows remarkable enrichments on both ends of the outcrop;

- \underline{V} and \underline{Ti} display a slight, overall increase west of the fault along the White Nossob River (F_a);

- the two major faults are indicated by the position of peaks for <u>Ba</u> and <u>Cu</u>; peaks of <u>Mn</u>, <u>Ni</u> and <u>Co</u> outline the F_b fault.

In particular, it is noted that:

- The <u>Cu</u> content in the Ongombo Gossan is lower than that recorded in the magnetite quartzite of the other two deposits examined. In fact, the Cu distribution at Ongombo is characterized by peaks greater than 1000 ppm above a background of 200-400 ppm. The main Cu enrichment occurs in proximity of the major western fault (fb) westwards. If that sector the gossan outcrop is affected by conspicuous veining and is characterized by the presence of the amphibole-rich patches. One of the highest values (0.32% Cu) coincides with a sample representative of one of these amphibole-rich pockets. The western extension of the Central Shoot appears to be the Curichest portion of the outcrop, with an average Cu content of approximately 1092 ppm Cu. The East Shoot has an average of 329 ppm Cu.

- In the irregular <u>Ba</u> profile, two main zones of enrichment are distinguished:

a) to the east, high Ba contents (up to 5221 ppm) characterize the outcrop close to the intersection of the Eastern Shoot EMP trend with surface;

b) at the western end of the mineralized outcrop Ba displays a



Fig. 40a-h - Ongombo Prospect: geochemical profiles for Cu, Zn, Mn, Ni, Co, Ba, Ti and V along the strike of the gossan outcrop. (F_a , F_b , and F_c correspond to faults)



Fig. 40a-h - See previous page.

series of peaks (from 4127 ppm up to 1.98% Ba). These localized high Ba anomalies appear to be related to faulting (e.g. the 4127 ppm peak of the F_C fault). The two main faults bordering the Central Shoot sector are outlined by Ba peaks. However, it is also possible that a primary Ba enrichment may have occurred in the margins of the mineralized body.

- The <u>Pb</u> profile (not shown in Fig. 40) shows a variable distribution with contents not exceeding 158 ppm. Only at the western end of the outcrop is there Pb enrichment.

- <u>Au</u> displays a relatively smooth profile (not shown in Fig. 40). Apart from small-scale variations within the range of the general average (163 ppb Au), peaks occur in correspondence with, and close to, the Eastern Shoot-surface intersection.

- \underline{Sn} , \underline{Zr} , \underline{W} , \underline{Bi} , \underline{Sr} and \underline{Mo} show a "broken" profile (not shown in Fig. 40). The first five elements are enriched in the western end of the outcrop. Sr shows some outstanding peaks (up to 364 ppm Sr), probably reflecting the Ba enrichment.

2.5.2 Trace element distribution in drillcores

Vertical profiles of the trace-element distribution in the drillcores OGB 151A, OGB 153A and OGB 154A from Ongombo are represented in Appendices J and K (see also Tables 5 and 6, Appendix E).

As reported in Part B and Part C, the metapelitic schists in the h/w of the ore zone are intercalated with mafic units of probable intrusive origin. Zones of chloritization and of sulphide- and/or magnetite dissemination appear to be spatially related to these mafic units. In the f/w, the unaltered schists locally carry fine disseminations of sulphides and/or magnetite, as well as remobilized ore in quartz veins.

From the examination of the geochemical profiles along the

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drillcores, the first point of note is that, on the whole, there is only a slight difference in the element distribution between the schists in the h/w and in the f/w of the ore zone. The peaks occurring in the h/w profiles coincide either with important lithological changes, i.e. from metapelitic schist to mafic unit, or with local sulphide and magnetite disseminations.

No significant trend of either enrichment or depletion in the h/w sequence towards the ore zone is discernible (see profiles in Appendix J and Fig. 41).

The following remarks are made:

- <u>Cu</u>, <u>Zn</u> and <u>Pb</u> peaks commonly coincide with zones of sulphide disseminations in the h/w schists. Cu and/or Zn peaks, of variable extent, characterize mafic and ultramafic units, although in correspondence with thin chlorite-amphibole schists (in OG 154) only a relative Zn and Pb enrichment is observed.

- <u>Au</u> is in isolated, small peaks, occasionally coinciding with Cu, with quartz veins and sulphide disseminations.

- <u>Ni</u> shows conspicuous enrichments in the mafic units and especially in their h/w ultramafic margin. Ni and Co peaks locally match, although Co is locally enriched in magnetite- and sulphide-bearing schist layers.

- The \underline{V} and \underline{Ti} distribution is mainly controlled by the distribution of mafic rocks.

- The <u>Cr</u> values in the barren f/w schists in the Ongombo drill cores are remarkably higher than those in the barren schists at Ongeama (compare Tables 5 and 7, Appendix E). Cr peaks correspond to maficultramafic units. Within these units differentiation is shown by the gradual variation in the Cr content. An inverted zonation for Cr is observed, e.g., in the mafic layers in the drill core OGB 153A. Within these mafic units, the highest Cr values coincide with the occurrence of the actinolite-chlorite-talc-rich h/w margins. The observed Cr distribution reflects, and further confirms, the overturning of the stratigraphic sequence, previously inferred on petrological grounds.

- The <u>Mn</u> distribution shows extreme variations. Prominent peaks occur in magnetite-bearing schist horizons, whereas low contents characterize some of the mafic units. The thickest mafic units show slight Mn enrichments towards the f/w (e.g. OGB 154A).

- <u>Ba</u> exhibits an irregular distribution, with values below 750 ppm in the h/w. A trend of Ba depletion in the h/w schists relative to the f/w schists is also observed. This trend is opposite to the Mn trend. In the h/w, major Ba lows correspond to the mafic-ultramafic units. An abrupt increase in the Ba content is observed in OGB 153A in schists directly overlying the ore zone, although magnetite- and sulphidebearing schists proximal to the ore zone do not generally show Ba enrichment. Ba peaks in the f/w only locally coincide with fine sulphide disseminations at the margins of late quartz veins (OGB 151A).

- The <u>Rb</u> and <u>Sr</u> distribution profiles are sympathetic, although Rb forms remarkable lows over mafic units. The Sr and Ba distributions are not entirely sympathetic. Because of its affinity for K, Rb depletion is indicative of K-feldspar-poor lithologies such as amphibolites. Beside Ba, Sr has affinity for Ca (Wedepohl, 1969). In amphibolites Sr abundance may be related to the Ca content in plagioclase and/or in calcic amphiboles (e.g. hornblende). In OGB 151A, Rb is depleted in the schists included between two mafic units above the ore zone. This depletion trend is followed to a lesser extent also by Sr, and is concomitant with some of the irregularities in the distribution of other elements (e.g. Ni, Pb, Zn, Mn, Ba).

- The irregular \underline{Zr} profile displays depletion in coincidence with mafic units. Zr peaks match Rb and Ba peaks. Zr and Y profiles are more irregular in the h/w than in the f/w schists.

- Nb, Mo, Sn, Bi and As show "broken" profiles with some isolated





at Ongombo (borehole No. OGB 153A).

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Fig. 42 - Geochemical profiles for Cu, Au, Pb, Zn, Co, Ni, Cr, Mn and Ba in the ore zone at Ongombo (borehole No. OGB 153A).

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values. Bi, Mo and Sn peaks are locally observed in magnetite- and sulphide-bearing schist horizons.

With regard to the ore zone, the following observations can be made (Fig. 42 and Appendix K):

- as expected ,the ore zone is characterized by a remarkable \underline{Cu} enrichment relative to the enclosing schists. Major peaks (11.75% Cu) are associated with layers of semimassive sulphide mineralization and with the upper magnetite quartzite horizon;

- <u>Au</u> (up to 3.78 ppm Au) and <u>Ag</u> peaks commonly match Cu highs in semimassive mineralization ; Ag is slightly enriched in MQ.

- the Zn profile sympathetically follows the Cu distribution, whereas <u>Pb</u> shows a flat profile with a few minor peaks;

- the <u>Ba</u> concentration in the ore zone is distinctly higher than that in the enclosing schists. Within the ore zone, Ba increases towards the bottom of the ore zone. Ba peaks correspond to schist horizons containing disseminated and stringer-type mineralization;

- <u>Cr</u> shows a variable distribution: peaks with over 100 ppm characterize either magnetite quartzite or disseminated mineralization in schists and in biotite selvedges;

- Zr, <u>Rb</u> and <u>Sr</u> (not represented in Fig. 42) are remarkably depleted within the ore zone relative to the wall rocks.

2.6 Alteration index as a tool for the evaluation of hydrothermal alteration

In order to evaluate the geochemical changes occurring in the orehosting metapelites during premetamorphic hydrothermal alteration, Haussinger and Okrusch (1990) propose the calculation of an "alteration index". This index is defined as the ratio between the sum of enriched elements and the sum of the enriched plus depleted elements in a rock.

The alteration index, calculated for both major and trace elements, is plotted against an index representing the elements immobile during metamorphism and alteration (Ti and Zr; Finlow-Bates and Stumpfl, 1981). Altered and unaltered lithologies define two different fields in the diagram, i.e. an upper field for altered schists and a lower field for unaltered schists. Fig. 43, for example, shows the result of the application of Haussinger and Okrusch's alteration index to the Gorob sulphide deposit. These authors recommend this method as a minescale exploration tool in order to locate a buried ore body in drill cores.

In the present study, the alteration indices for major and for trace elements were used for the purpose of comparing the chloritized h/w schists in the Ongombo drill cores and the magnetite-bearing schists from the "I anomaly" at Ongeama, with the assumed altered schists at Ongeama and Matchless West Extension (Tables 16a-g, Appendix E).

Haussinger and Okrusch's alteration index for major elements, Imajor, is defined as:

 $Fe_2O_3tot + MgO + MnO$ $I_{major} = ------,$ $(Fe_2O_3tot + MgO + MnO) + (CaO + K_2O + Na_2O)$

the alteration index for trace elements, Itrace, as

and TZ as



Fig. 43 - Alteration index for trace elements against TZ index for the alteration zone at Gorob (modified after Haussinger and Okrusch, 1990). The dashed field includes smples of unmineralized schists from the Kuiseb Formation.



- alteration zone (Ongeama)
- alteration zone (Matchless West Extension)
- + unmineralized country schist
- magnetic schists ("I anomaly" area)

Fig. 44a

Fig. 44a-c - Discrimination between unmineralized schists and hydrothermally altered schists associated with mineralization (alteration zone) by means of Haussinger and Okrusch's (1990) alteration index for trace and major elements (see text for details). a) Itrace vs. TZ index diagram for schists from alteration pipes and unmineralized country schists. b) Itrace vs. TZ index for schists from alteration pipes and schists from the ore-hosting sequence at Ongombo. c) Imajor vs. TZ index for schists from alteration pipes and unmineralized country schists.



Fig. 44a-c - See previous page.

TiO₂ TZ = ----- x 100 Zr

Results are illustrated in Fig. 44a-c. It is further observed that:

- samples from the assumed alteration zones of Ongeama and Matchless West Extension plot in a field characterized by both I_{major} and I_{trace} values greater than 0.9 (Fig. 44a-c);

- magnetic schists from the "I anomaly" plot within the field defined by I_{major} and I_{trace} values less than 0.7 and 0.9, respectively, and occupied by the Ongeama barren schists (Fig. 44a and 44c).

- samples from the chloritized h/w schists in the Ongombo drill cores plot in the lower I_{trace}-TZ field, in which both the f/w schists and the barren schists from Ongeama are included (Fig. 44b). Some samples from these h/w schists plot in a position intermediate between the two main fields. Some of these samples represent schist units containing magnetite and sulphide disseminations either related to mafic units or isolated (see the corresponding profiles in Fig. 41). In the immediate h/w of the ore zone , only some drill core samples (from OGB 154A) show I_{trace} index values comparable with those in altered schists (see Table 16g, Appendix E). In those core samples Ba enrichment is recorded.

3 ULTRAMAFIC EXHALITES AND IGNEOUS ULTRAMAFIC ROCKS: A COMPARISON

Tables 3, 5, 17 and 18 (Appendix E) contain trace- and major-element abundances of the ultramafic schists from Matchless West Extension, the ultramafic bodies outcropping near Otjihase and, in part, of the ultramafic differentiates in the mafic units at Ongombo (see sections 3 and 4.3 Part B and section 3.32 Part C for details).



+ ultramafic schists (Matchless West Extension)

igneous ultramafic rocks (Elisenheim, Otjihase, Ongombo;

Fig. 45 - Comparison between ultramafic schists associated with the Matchless West Extension sulphide deposit, and igneous ultramafic rocks by means of the "Ex" and "Um" indexes (see text for details).

A comparison between the major-element geochemistry of the exhalative ultramafic schists, from Matchless West Extension, and the Damaran ultramafic bodies (data in Table 18, Appendix E, and in Finnemore, 1974, and Barnes, 1983), highlights the fact that the major-element contents are quite similar in the two rock categories.

Outstanding differences are shown, on the other hand, in the traceelement geochemistry of these two ultramafic rock types. Therefore, two groups of trace elements were selected (Table 19, Appendix E) in order to discriminate between the assumed exhalative ultramafic schists and ultramafic rocks of igneous derivation.

The first group of elements includes Cr, Ni and Co. These elements were chosen because of their typical association with mantle-derived material (Henderson, 1982). The second group of elements includes Ba, Cu, Zn and Pb, which are considered as discriminators for an exhalative environment. For each sample the indexes "Um" and "Ex" are defined by the sum of the (Cr + Ni + Co) and of the (Ba + Cu + Zn + Pb) contents, respectively.

In Fig. 45, two distinct distributions are obtained. The ultramafic exhalites plot within a field defined by high Ex and low Um values. This field is separated from the high Um-low Ex field including the igneous ultramafic rocks.

The geochemical separation between these two rock groups is considered significant for discriminating the exhalative, and not igneous, nature of the ultramafic schists at Matchless West Extension.

DISCUSSION

Geochemical results discussed in this chapter document the existence of well-defined trends of element distribution within the mineralized outcrops at Ongeama, Matchless West Extension and Ongombo. It is pointed out that the presence of geochemical zoning at Ongeama was not recognized by previous lithogeochemical sampling (Andrew, 1984). The presence of element zoning along the alteration pipe at Ongeama and Matchless West Extension is meaningful because:

a) together with field observations and petrological study, it is proof of the hydrothermal origin of these pipe-like features;

b) the presence of an alteration pipe aids in the interpretation of the element zoning within the mineralization in terms of proximal and distal facies.

In the three mineralized bodies studied, large-scale zoning exhibits some common features.

- Cu primarily, and, to a variable extent, Zn, Au and Mn, are typically enriched close to the alteration pipes in both the Ongeama and Matchless West Extension gossans. The similar Cu + Zn + Mn distribution, together with the coincident, and localized, Ni and Co enrichment in the western extension of the Central Shoot suggest proximity to the source of the ore-bearing fluids. The association of these metals with the amphibole-rich ore facies at both Ongeama and Ongombo, may be more than just a coincidence.

- In the three orebodies studied, Pb, Ag, Ba, Sn and, to some extent, Zn and Au, represent the elements concentrated in a "distal" position (e.g. exhalites).

Fig. 46 schematically shows these trends.

However, it is noted that there are some minor differences, namely:

- at Matchless West Extension and Ongombo, "distal" enrichment of W and Bi is observed (see Appendices H and I). However, at Ongombo, the occurrence of Bi, W, together with Sn and Pb highs at the western end of the gossan suggests the affiliation of these metals to late tourmalinite veins, which are common and well developed in that part of the outcrop.



Fig. 46 - Schematic representation of the trace element distribution within the mineralization in relation to the vertical and horizontal distance from the vent. Elements in brackets are enriched to a limited extent in the corresponding zone.

- only the Matchless West Extension deposit exhibits an unusual Cu enrichment in ultramafic exhalites and in a distal position relative to the alteration pipe (Fig. 33 and Appendix H). This feature may be interpreted in three different ways, namely:

a) in consideration of the moderate to high mobility of Cu in an oxidizing environment, such as is found in semi-arid climates (Rose et al., 1979), one possible explanation may be the deposition of secondary Cu-sulphides and/or oxides from migrating groundwater;

b) high Cu content (up to 0.7%) is recorded in present-day exhalative talc deposits in the Guaymas Basin (Lonsdale et al., 1980). Modern Mg-rich deposits are typically formed proximal to the hydrothermal vents (Lonsdale et al., 1980, and references therein; McLeod and Stanton, 1984). This would imply that the Mg and Cu enrichment in the westerly "distal" ultramafic schists at Matchless West Extension may, in fact, correspond to an additional small vent subsidiary to the main vent in the east (Smalley, 1990, pers. comm.);

c) the third possibility takes into account the actual position of the Cu-rich ultramafic beds within the mineralized horizon at Matchless West Extension (see Fig. 21). This may be the result of tectonic remobilization of ductile, talc-rich material, proximal to the vent, along planar discontinuities (e.g. compositional layering in the magnetite quartzite) during deformation.

Residual enrichment of weathering-resistant baryte, unaffected by the acidic environment of sulphide oxidation, accounts for the anomalous Ba contents within the mineralized bodies considered. Baryte is a common residual mineral in the Matchless Belt gossans (Andrew, 1984). Although concentrated in exhalites, Ba also displays a strong positive correlation with the alteration zones. In fact, the hydrothermally altered schists are characterized by significant Ba contents relative to the unaltered country rocks. In the major gossan outcrop at Ongeama, Ba is more enriched in the h/w altered schists than in the ore zone. In the Ongombo drillcores, Ba peaks characterize the disseminated sulphide-bearing red biotite schists, and progressive increase in the Ba content is also detected in the schists immediately above the ore zone in borehole OGB 154A (see Fig. 41 and 42).

In the hydrothermally altered host rocks of Cyprus-type mineralization, Ba is believed to be incorporated into hydrothermal sericite and illite in substitution for K (Herzig, 1986, 1988). Babearing micas are included in the gangue mineralogy of metamorphosed sediment-hosted mafic volcanic-related sulphide deposits of the Broken Hill-type (Plimer, 1986). The presence of Ba in the mineralized system suggests a strong similarity of the ore-forming fluids to fluids responsible for modern Cu-Zn deposits in oceanic environments (Lonsdale and Becker, 1985; Peter and Scott, 1988; Koski et al., 1988; Kappel and Franklin, 1989). Ba is probably derived either from leaching of mafic rocks and consequent breakdown of a Ba-bearing phase (plagioclase with intermediate anorthite content; Herzig, 1986), or from seawater sulphate (Lydon, 1988).

There is, in general, a fair degree of concurrence between the results of this study and those from the previous investigations (Hoffmann, 1976; Andrew, 1984). Agreement is found on the position of the main anomalous zones within the mineralized outcrops and on the main trends of metal distribution. However, a few minor discrepancies are also observed. They are:

- the overlap between the Cu and Ag distribution observed by Hoffmann (1976) and Andrew (1984) is not recorded in the geochemical profiles of the gossans. The sympathetic relationship between these two metals is, nevertheless, noticeable in the unweathered mineralization (Ongombo drillcores), where Ag could be lattice-bound to chalcopyrite. A similar primary association is observed at Otjihase (Goldberg, 1976);

- no significant Co distribution was detected at Ongeama during the present study.

According to Andrew (1980, 1984) the composition of the sulphide ore and the progressive geochemical changes during supergene alteration have a fundamental influence on the ultimate composition of the corresponding gossan. The ratio of pyrite and/or pyrrhotite versus matrix sulphides (e.g., chalcopyrite, sphalerite) in massive ore is proportional to the acid production during oxidation, to leaching and consequent dispersion of metals in groundwater. Insufficient acid buffering dependent on a chemically inert gangue, such as quartzite, further promotes metal dispersion. However, where sulphides are less abundant, e.g. in disseminated to semi-massive mineralization, at Ongeama and Ongombo, and in other deposits in the Matchless Belt, acid leaching can be less pronounced and higher concentrations of trace metals are retained in the corresponding gossan. This may be a reason for the preservation of well defined zoning in the deposits examined.

Hydrous Mn oxides compete with Fe-hydroxides for trace metal ions amenable to adsorption (Rose et al., 1979). Significant association of Mn with Zn and Cu in both "proximal" and "distal" positions may be, in fact, of secondary origin, as this association does not occur in unweathered ore. The significant Cu-Zn association, probably due to the exsolution of sphalerite in chalcopyrite and viceversa (see Section 3.3.3 Part C), may be enhanced during gossan formation by coprecipitation of the two metals with Fe- and Mn-hydroxides.

In spite of metamorphism and deformation, the geochemical zoning revealed in this study is comparable with that observed in ancient and present-day seafloor volcanogenic polymetallic sulphide deposits. In these deposits, a high-temperature Cu(-Zn)-rich core typically grades into lower-temperature Zn-Pb-Ba-rich margins (Bischoff et al., 1983; Hoy et al., 1984; Peter and Scott, 1988; Embley et al., 1988; Barrett et al., 1988; Lydon, 1988). In currently forming sulphide deposits, the development of metal zoning appears to be a gradual process. Zoning is dependent on redox potential and temperature variations, and, partly, on selective remobilization of metal ions during the interaction of high-temperature solutions with massive sulphides previously deposited (Janecky and Shanks, 1988).

In the deposits considered in this study, the pronounced lateral zonation of elements and the high metal values in either the central or thicker portion of the mineralized bodies are probably due to an original depositional configuration. Although the patterns of zoning may also be the result of deformation, the coincidence of the zonation patterns observed with the development of unusual rock types, previously referred to as alteration zone, cannot be explained by simple tectonic deformation.

The occurrence of Au enrichment in either "proximal", hightemperature ore facies (massive sulphide ore) or in "distal", lowertemperature ore facies (exhalite), as seen in the major and minor gossan outcrop at Ongeama respectively, is believed to be a function of the different chemistry of the ore-bearing fluids (Hannington and Scott, 1988). Au is deposited from a bisulphide complex in pyritic Zn-Pb-Ba-Ag-SiO₂ assemblages at lower temperatures (less than 300°C), at high oxygen and sulphur activity and near-neutral pH. At temperatures. higher than 300°, the metal is deposited from chloride complexes in sulphur-poorer, Cu-rich assemblages, characteristic of low-pH, highsalinity fluids. Therefore, the Au increase in the magnetite guartzite, the concomitant overall Ag enrichment in the minor gossan outcrop relative to the major gossan (see Section 2.1 above for details), suggests that the former could be the surface expression of a well defined mineralizing system. This mineralizing system may be distinct from that represented by the major gossan outcrop, which is characterized by a different fluid chemistry.

The differences in the geochemistry of the schists within the assumed alteration zones, compared with the country rocks, and the existence of lateral and axial zoning of elements within the discordant alteration pipes, can be considered as definitive evidence of a hydrothermal origin of the latter. The elements involved in the zoning are Cu, Zn, Ba, Au, As, Zr, Ni, Co, Cr, Rb, Sr, Sn and Bi (see Sect. 2.4 above and Fig. 47). The alteration pipes at both Ongeama and Matchless West Extension are characterized by Rb and Sr depletion relative to the country-rock schists. Because of the affinities of Rb with K and Sr with Ca and Ba (Wedepohl, 1969), Rb and Sr variations may monitor variations in alkalis in rocks. In the alteration pipes examined, the Rb and Sr depletion significantly coincides with the



Fig. 47 - Scheme of the geochemical zonation within the alteration pipe at Ongeama (a) and Matchless West Extension (b). Arrows indicate the direction of increasing abundance of the corresponding elements. - 132 -

alakali depletion and the absence of plagioclase and, at Ongeama, with the occurrence of chloritoid. The latter is an index mineral for a low-alkali chemical environment (Deer et al., 1982). Alkali depletion along hydrothermal alteration pipes is due to hydrolysis of rockforming minerals by means of interaction between high-temperature acidic solutions and wall rocks. This is a common feature in volcanogenic sulphide deposits. The lesser Sr depletion, relative to Rb, recorded at Ongeama and the type of Sr distribution observed at Matchless West Extension suggest a limited association of Sr also with Ba.

The alteration index, proposed by Haussinger and Okrusch (1990) as a discriminator between altered and unaltered turbidite schists, was tested in this study on the alteration zones at Ongeama and Matchless West Extension. It is concluded that the use of this index is effective. With this assumption, the diagrams obtained for the h/w chloritized schists in the Ongombo drillcores (Fig. 44b) further indicate that the chloritization, and the subordinate magnetite + sulphide disseminations, are not caused by ore-related hydrothermal alteration. Given the spatial association of chloritization with the mafic intercalations in the h/w schist sequence, it is contended that these modifications originate from contact metasomatic processes induced by sheet-like mafic intrusions in H₂O-rich deep-sea sediments. The values of the alteration index suggest that also the magnetite-bearing schists from the "I anomaly" (Fig. 44a and 44c) do not correspond to any ore-related hydrothermal alteration.

Apart from the alteration index, the use of base metals and Ba, together with Cr and Ni, is also effective in discriminating between different ultramafic rock types and in revealing the exhalative nature of the ultramafic schists at Matchless West Extension. CONCLUSIONS

1 GENERAL DISCUSSION

The aims of this study were to:

 a) evaluate the effects of metamorphism and the significance of mineralogical changes in both mineralization and wall rocks;

b) interpret the mineralogical changes and related geochemical patterns in the mineralization and wall rocks;

c) attempt and relate these patterns to an original geochemical signature using examples from similar present-day sulphide deposits.

Sulphide mineralization in the Matchless Amphibolite Belt is generally attributed to a Besshi-type genetic model according to Fox's (1984) definition (e.g. Breitkopf and Maiden, 1988). In the belt, mineralization consists of sediment-hosted cupriferous iron sulphide deposits, associated with mafic rocks, and containing primarily pyrrhotite and/or pyrite, chalcopyrite and sphalerite. This mineralization is thought to have formed within a narrow, oceanic basin, the Matchless Trough, in which abundant terrigenous sediments accumulated. The present-day Guaymas Basin (Gulf of California) may be a modern example of this tectonic setting (Lonsdale and Becker, 1985) (see Fig. 8 for location).

The main features observed in the sulphide deposits at Ongeama, Matchless West Extension and, only in part, Ongombo, can be summarized as follows:

a) the presence of a discordant, pipe-like feature in the schists enclosing the mineralized horizon. This discordant feature is distinct from the country rocks, and is characterized by mineralogical and metal zoning. This feature is interpreted as an originally typical, inverted cone-shaped footwall alteration pipe; - the presence of massive to semi-massive sulphide mineralization rich in silica, but characterized by Mg-rich minerals (talc, anthophyllite) and high Fe, Cu and Zn contents in its original high-temperature portion adjacent to the assumed alteration pipe (syn-discharge deposit; Bonatti, 1983);

- the presence of a Pb-Zn-Ba-enriched ferruginous siliceous cap consisting of metamorphosed chemical oxide sediments, locally interlayered with sulphide lenses (post-discharge deposit; Bonatti, 1983).

The results of this study indicate that the sulphide deposits at Ongeama, Matchless West Extension and Ongombo originated as sulphide mounds. A good example is afforded by the small sulphide body at Matchless West Extension, in which a sulphide mound configuration is suggested by:

- 1) the rapid lateral zonation in rock types;
- 2) the development of the massive sulphide portion above the vent;
- 3) the peripheral location of oxidized chemical sediments.

The formation of this deposit was probably due to a very short-lived hydrothermal system (see below).

At Ongeama and Ongombo the evidence for sulphide mounds is less compelling. The elongated nature of the mineralization in these two sites, and, especially at Ongeama, the development of an elongated massive sulphide lens in the basal portion of the ore zone, overlain by an almost continous exhalite horizon, may indicate that these two deposits formed from density-stratified brine-pools. However, this hypothesis may not be feasible. There is some geochemical evidence that the eastern sector of the original Matchless Trough, where Ongeama and Ongombo are situated, was in fact characterized by a fast spreading rate, and that conditions of full ocean opening were approached (Breitkopf and Maiden, 1988). Apart from considerations of fluid density, a fast spreading rate would not allow the formation of long-lived topographic depressions within which brines can pond over a lengthy period.

At Ongeama, the occurrence of specific ore rock types (i.e. the porous cupriferous anthophyllite-bearing semimassive facies) which only occur proximal to the vent, is suggestive of a mound-like zone. A more sustained, and probably episodic hydrothermal activity, compared to that at Matchless West Extension, would have been responsible for the lateral accretion of the sulphide body by means of:

a) lateral accumulation of massive sulphide detritus resulting from the disaggregation of inactive chimneys;

b) leaking of ore-bearing fluids through the relatively porous structure of the accreting sulphide mound.

The overall banded nature of the massive-sulphide ore and the common occurrence of phyllitic lenses in it suggest an original roughly bedded, clastic detrital ore deposition at the sides of the mound (sulphide talus).

Similar considerations may be applicable to the Ongombo deposit, if the anthophyllite and corresponding metal enrichment in the western extension of the Central Shoot are regarded as indicative of proximity to a discharge vent.

The extremely elongate lens-like shape of the Ongeama and Ongombo deposits is interpreted to be the result of tectonic stretching. This, coupled with metamorphic recrystallization, may be responsible for the textural homogenization of the mineralization.

Based on the geological, mineralogical and geochemical characteristics discussed in this study, a general model that may account for the formation of these sulphide deposits, is represented in Fig. 48.

The model shown in Fig. 48 considers the interaction between fracture zones and the basaltic sill-sediment complex (Einsele et al., 1980; Einsele, 1985). According to this model, the effect of sills intruding soft sediments results in a strong decrease in porosity, contact metamorphism ("baking" of the enclosing sediments), decomposition of organic matter and expulsion of metal-enriched, hot pore-fluids. Three distinct, but superimposed, hydrothermal circulation systems can be considered (Einsele et al., 1980; Lonsdale and Becker, 1985):

 a) short-lived expulsion of pore water, as a result of the intrusion of shallow mafic sills;

 b) a more sustained and important deep-seated circulation, probably due to a cooling magma chamber: this circulation discharges at high temperature through fractures in the overlying sills;

c) slowly recharging hydrothermal convection within the sediments overlying the shallows sills, possibly driven by heating from stillhot intrusions.

The main recharge to the system would occur along major riftbounding faults. The discharge is necessarily focussed along faults or fracture zones which intersect the otherwise impermeable cap of mafic sills, and along which also the sedimentary cover is more permeable. Lonsdale and Becker (1985) noticed that in present-day, heavily sedimented oceanic basins, mineralization overlies zones of high permeability (faults) in the igneous rocks, and is not formed because the fluids rise along faults in the superficial sediments.

One possible source for ore-bearing solutions is seawater modified and enriched with metals by reaction with basaltic rocks at water/rock ratios between 16 and 22 (Reed, 1983). The other possible source for metals is magmatic. This is shown by the enrichment of volcanogenic



mound (e.g., Ongeama) (b).

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helium (³He) in the bottom waters next to discharge sites in the Guaymas Basin (Lupton, 1979). Sulphur-isotope analyses of the Otjihase mineralization (chalcopyrite and pyrite) give δ^{34} S values ranging from +7.9%. to +9.4%. (Pirajno, pers. comm., 1990). On the basis of a sulphur-isotope research programme being conducted by F. Pirajno (Rhodes University) and J. Kinnaird and A. Fallick (St. Andrews University, Scotland) on different types of sulphide-bearing mineralization in Namibia, it appears that the above values for Otjihase may be interpreted as mixing between magmatic sulphur (δ^{34} S from -2% to +5%.) and seawater-derived sulphur (δ^{34} S approximately +20%.).

The Cu-Zn-rich sulphide accumulations were formed at the sedimentwater interface by short-lived to sustained discharge of hot (150-350°C) metal-bearing fluids. Evidence of episodic hydrothermal activity may be seen in the alternation of sulphide mineralization with magnetite guartzite observed at Ongombo (see Part B and C). Boiling probably did not occur because of the high hydrostatic pressure at the seafloor (Plimer, 1981). Although it is established that metamorphic desulphurization can turn pyrite into pyrrhotite, the metamorphic grade here is rather low for this to have happened (McDonald, 1967). The interaction between ore-bearing fluids and reducing organic matter may account for an original pyrrhotite-rich nature of the mineralization. Organic matter, as hydrothermal petroleum condensates associated with mineralization, occurs at the discharge sites in the Guaymas Basin and in the Escanaba Trough (Simoneit and Lonsdale, 1982; Von Damm et al., 1985; Koski et al., 1988). Graphitic schists relatively proximal to the mineralization at Ongeama (Hoffmann, 1976) and at Otjihase (Thomson, 1989) may have had the same origin.

A rapid and short-lived discharge is inferred for the sulphide deposits at Matchless West Extension. This is suggested by the size of the deposit, and the strong talc enrichment. According to Lonsdale et al. (1980), rapid, unidirectional and focussed fluid motion along the hydrothermal conduit can explain the preservation of high Mg - 140 -

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concentrations in the discharged fluid. In recirculating hydrothermal systems Mg can be partly extracted at depth by rock alteration reactions. Costa et al. (1983) explained similar talc enrichments at Mattagami as being caused by turbulent mixing of hot saline fluids with Mg-rich seawater.

Along the fluid conduit, where steep thermal gradients are encountered, ion exchange reactions between metal-bearing solutions and wall rocks would cause a variable degree of hydrothermal alteration with intensity decreasing with distance from the conduit. Silicification, addition of metals (Fe, Cu, Zn, Pb, Ba, etc.) and depletion in alkalis due to hydrolysis of rock-forming minerals, are the main characteristics of the alteration zone.

The exhalative activity probably developed in conjunction with an active magmatic phase, as observed in present-day oceanic environments (Einsele et al., 1980; Kappel and Franklin, 1989).

A metamorphosed equivalent of sill-sediment complex associated with the mineralization is well preserved in the eastern Windhoek sector of the Matchless Belt. Ongeama and Ongombo, for example, display a sequence of mafic and metaturbidite rocks in their structural hangingwall (i.e. stratigraphic footwall).

Evidence revealed in this study indicates that the sequence at Matchless West Extension, and at Matchless Mine is the right way up. this indicates that a thick volcano-sedimentary pile is present only in the hangingwall of the mineralization (see Fig. 12). Although exhalative activity might have preceded the magmatic activity (Smalley, 1990, unpubl.), thrusting along the Colvania Horizon might have been responsible for the slicing-off of an original sill-sediment complex in the footwall of the sulphide deposits. This appears to be confirmed by the width of the Matchless Amphibolite Belt in the western Windhoek sector (where Matchless West Extension and Matchless Mine are situated), which is less than that in the Eastern Windhoek Sector (where Ongeama, Ongombo and Otjihase occur).

2 IMPLICATIONS FOR EXPLORATION

During metamorphism, the mineralization and associated hydrothermal alteration behaved as a closed system, i.e. their components experienced very limited chemical diffusion. This favoured the preservation of the main original features and of the zoned element distribution, as outlined in this study. Therefore, important implications for exploration are that some of the characteristics of hydrothermal alteration zones, even if overprinted by metamorphism and disrupted by tectonism, may still be recognized.

In the case of exploration for blind volcanogenic mineralization in metamorphic terranes, the detection of hydrothermal alteration is fundamental in indicating the proximity to the mineralization during follow-up operations. The alteration zone is characterized by both mineralogical and geochemical variations with respect to the country rocks. Although mineralogical changes can be easily detectable during detailed mapping, this study has shown that, in some cases, the original pre-metamorphic hydrothermal alteration can be recognized by means of subtle changes (e.g. the change in colour of biotite under the microscope, as at Ongombo). The geochemical variations detected are comparable to the general geochemical patterns observed in modern and ancient volcanogenic mineralized systems.

It is emphasized that a combination of mineralogical and geochemical factors is required to distinguish successfully between ore-related alteration and alteration-like features induced by other processes, not related to ore-formation.

The careful evaluation of selected geochemical data and the use of alteration indices are useful and effective exploration tools. These help in determinating the potential for mineralization.

In metamorphosed terranes, where deformation has modified most of the original sedimentary and volcanic structures, and where reliable markers are rare, the determination of the stratigraphic position of the alteration zone relative to the mineralization is of foremost importance for the interpretation of the structure and the way up of the strata in the explored area.

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APPENDIX A

Ongeama Prospect

Geological map of the gossan outcrop

Scala 1:1000

(in the back pocket)

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APPENDIX B

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Ongombo Prospect Logs of the ore-bearing sequence in drill cores

Boreholes no. OGB 151A, OGB 152A, OGB 153A, OGB 154A

(in the back pocket)

APPENDIX C

LIST OF THE SAMPLES

Abbreviations used in the table:

QBC - quartz-biotite-chlorite schist gossan. - gossanous T1 - massive quartz-biotite schist qtz = quartz T2 - foliated quartz-blotite-muscovite-chlorite schist chlor. - chlorite T3 - contorted biotite-muscovite-chlorite-quartz schist amph. - amphibole magn. - magnetite QMP - quartz-muscovite-pyrite schist actin. - actinolite H/W - hangingwall (relative to the ore zone) sulph. - sulphide(s) F/W = footwall (relative to the ore zone) ES - Eastern Shoot (Ongombo) carb. - carbonate CS - Central Shoot (Ongombo) epid. - epidote

ONGEAMA

zanto	le no.	description	grid reference	06
				OG
00	2	qtz-rich massive gossan	11060E/10240N	OG
OG	3	qtz-rich massive gossan	11050E/10247N	OG
OG	4	qtz-rich massive gossan	11040E/10239.5N	OG
OG	5	qtz-rich massive gossan	11040E/10252.5N	DG
OG	6	qtz-rich massive gossan	11020E/10241.7N	OG
OG	7	ferruginous schist	11020E/10255N	OG
OG	8	qtz-rich massive gossan	11000E/10245.5N	DG
OG	9	qtz-rich massive gossan	10980E/10240.5N	OG
OG	10	qtz-rich massive gossan	10960E/10250N	OG
DG	11	qtz-rich massive gossan	10940E/10250N	OG
OG	12	qtz-rich massive gossan	10920E/10255.3N	06
OG	13	magnetite quartzite	10900E/10240N	OG
OG	14	magnetite quartzite	10900E/10250N	OG
OG	15	qtz-rich massive gossan	10900E/10258.2N	DG
OG	16	magnetite quartzite	10880E/10246N	OG
OG	17	magnetite quartzite	10880E/10260.3N	OG
OG	18	magnetite quartzite	10850E/10263M	OG
OG	19	magnetite quartzite	10840E/10271N	OG
OG	20	magnetite quartzite	10820E/10268.5N	OG
06	21	magnetite quartzite	10800E/10272.3N	OG
0G	22	magnetite quartzite	10780E/10272.9N	DG
OG	23	magnetite quartzite	10750E/10276N	OG
OG	24	qtz-rich massive gossan	10760E/10273N	OG
OG	25	ferruginous schist gossan	10720E/10253.3N	OG
OG	26	qtz-rich massive gossan	10560E/10257.4N	OG
OG	27	qtz-rich massive gossan	10540E/10244N	OG
OG	28	qtz-rich massive gossan	10520E/10244N	OG
OG	29	qtz-rich massive gossan	10500E/10252N	OG
OG	30	ferruginous schist gossan	10500E/10250N	OG
OG	31	qtz-rich massive gossan	10480E/10235.8N	OG
OG	32	ferruginous schist gossan	10480E/10237N	ÓG
OG	33	qtz-rich massive gossan	10460E/10230M	OG
OG	34	magnetite quartzite	10460E/10232N	OG
OG	35	qtz-rich massive gossan	10440E/10225M	OG
0G	36	magnetite quartzite	10440E/10219N	06
OG	37	ferruginous schist gossan	10420E/10224M	OG
OG	38	gtz-rich massive gossan	10420E/10220N	0G 1
OG	39	magnetite quartzite	10420E/10214N	0G 1
OG	40	ferruginous schist gossan	10400E/10222N	0G 1
OG	41	gtz-rich massive gossan	10400E/10219N	0G 1
00	42	magnetite quartzite	10400E/10212N	06 1

OG	43	magnetic muscovite schist	10380E/10216N	OG 105	qtz-rich m
OG	44	magnetite quartzite	10380E/10206N	OG 106	magnetite
OG	45	glz-rich massive gossan	10380E/10212N	DG 107	magnetic m
OG	45	magnetic muscovite schist	10360E/10213N	05 108	qtz-rich m
OG	47	gtz-rich massive gossan	10360E/10206N	OG 109	magnetic m
OG	48	magnetite quartzite	10360E/10204N	OG 110	qtz-rich m
OG	49	magnetic muscovite schist	10340E/10209N	OG 111	magnetic m
OG	50	gtz-rich massive gossan	10340E/10205N	OG 112	gtz-rich m
OG	51	magnetite quartzite	10340E/10200N	OG 113	magnetic m
OG	52	magnetic muscovite schist	10320E/10203N	OG 114	atz-rich m
OG	53	gtz-rich massive gossan	10320E/10196.5N	0G 115	atz-rich m
0G	54	magnetite quartzite	10320E/10190N	DG 115	magnetic m
OG	55	magnetic muscovite schist	10300E/10203N	OG 117	atz-rich m
OG	56	otz-rich massive gossan	10300E/10194N	OG 118	atz-rich m
OG	57	magnetite quartzite	10300E/10187.5N	0G 119	unminerali
OG	58	magnetic muscovite schist	10280E/10201N	06 120	mafic meta
OG	59	gtz-rich massive oossan	10280E/10193N	OG 121	mafic meta
DG	60	magnetite quartzite	10280E/10188N	05 122	mafic meta
05	51	ntz-rich massive nossan	10277E/10197N	05 123	mafic meta
nc	62	magnetic muscowite schist	102505/102018	06 124	mafic meta
00	63	semi_massive mossan	10260E/10197N	06 125	mafic meta
00	64	formuninous schist	10260F/10193N	06 126	mafic meta
00	65	of z_rich massive dossan	10250F/10199N	OG 127	mafic meta
05	66	mannetite quartzite	10260E/10182N	OG 128	mafic meta
DG	67	magnetic muscovite schist	10240E/10189N	OG 129	unminerali.
OG	68	sem1-massive possan	10240E/10182N	OG 130	unminerali
OG	69	magnetite guartzite	10240E/10177N	ÓG 131	unminerali
OG	70	otz-rich massive oossan	10240E/10175N	OG 132	unminerali
06	71	unmineralized OBC schist (F/W)	10240E/1016BN	OG 133	unminerali
06	72	semi-massive opssan	10220E/10180N	OG 134	unminerali
DG	73	magnetite quartzite	10220E/10172N	06 135	unminerali
06	74	otz-rich massive possan	10200E/10176.5N	06 136	unminerall
DG	75	mannet ic muscovite schist	10200E/10179N	0G 137	magnetic 0
06	76	magnetic muscovite schist	10180F/10191N	0G 138	magnetic c
DG	77	mannet ic chlorite schists	10180E/10178N	06 139	magnetic 0
05	78	magnetic chlorite-miscovite schist	10180E/10165N	06 140	magnetic 0
OG	79	semi-massive nossan	10180E/10165N	OG 141	magnetic 0
06	80	magnetic chlorite schists	10160E/10190N	0G 142	magnetic 0
06	81	magnetic chlorite schists	10160E/101BON	06 143	magnetic 0
06	82	atz-rich massive nossan	10160E/1015BN		and the state of
00	83	mannetite nuartzite	10160E/10165N		
00	84	magnetic chlorite schlete	101405/101908	ONCOMBO COSSAN	
00	86	magnetic miscoulte schist	101405/101808	01100100 00000111	
00	00	ats wish massive morean	101405/101728	Concernation of the second	PARCES.
00	00	qtz-rich massive gossan	101205/101058	sample no	description
00	89	magnetic chlorite schists	101206/101908	sampre no.	desci iprio
00	80	formulaus schiet	101206/101838	05 144	mannetite
00	09	att nich marcina antran	101206/10181#	06 145	magnetite
00	90	que-rich massive gossan	101202/101014	00 145	magnetite
00	91	magnetic ciriorite scitists	101005/102028	00 147	magnetite
Ou oc	92	magnetic muscovite schist	101005/101938	DC 1AR	magnetite
UL	93	magnetic muscovite schist	101002/101934	00 140	magnetite
00	94	qtz-rich hassive gossan	101005/101918	00 149	magnetite
06	95	qtz-rich massive gossan	101002/101/04	06 150	magnetite
00	90	magnetic chiorite schists	100805/1019/#	00 151	magnetite
0G	5/	quz-rich massive gossan	100805/101908	00 152	magnetite
UG	50	magnetic muscovite schist	10000E/10192N	05 153	Regnetite
OG	99	magnetic muscovite schist	100002/101/08	06 154	magnetite
OG	100	qtz-rich massive gossan	10080E/10172N	06 155	magnetite
OG	101	magnetic muscovite schist	10060E/10192N	06 156	magnetite (
00	102	qtz-rich massive gossan	10060E/10175N	OG 157	magnetite d
00	7.13	the second s	The second se	00 100	
OG	103	magnetite quartzite	10060E/10170N	OG 158	magnetite d

qtz-rich massive gossan	10040E/10194N
magnetite quartzite	10040E/10191N
magnetic muscovite schist	10040E/10184N
qtz-rich massive gossan	10040E/10180N
magnetic muscovite schist	10020E/10201N
qtz-rich massive gossan	10020E/10196N
magnetic muscovite schist	10000E/10202N
qtz-rich massive gossan	10000E/10198N
magnetic muscovite schist	9980E/10212N
qtz-rich massive gossan	9975E/10210N
qtz-rich massive gossan	9968E/10213N
magnetic muscovite schist	9940E/10217N
qtz-rich massive gossan	9940E/10213N
qtz-rich massive gossan	9950E/10191N
unmineralized QBC schist (H/W)	10920E/10270N
mafic meta-lavas - upper layer	B400E/11160N
mafic meta-lavas - upper layer	B090E/11100N
mafic meta-lavas - central layer	8000E/10840N
mafic meta-lavas - central layer	8180E/10860N
mafic meta-lavas - central layer	8300E/10860N
mafic meta-lavas - lower layer	8100E/10310N
mafic meta-lavas - lower layer	8160E/10290N
mafic meta-lavas - lower layer	8260E/10310N
mafic meta-lavas - upper layer	8700E/11280N
unmineralized QBC schist (F/W)	10700E/10164N
unmineralized QBC schist (F/W)	10690E/10168N
unmineralized QBC schist (F/W)	10400E/10120N
unmineralized QBC schist (F/W)	10380E/10100N
unmineralized QBC schist (H/W)	10360E/10300N
unmineralized QBC schist (H/W)	10320E/10300N
unmineralized QBC schist (H/W)	10290E/10255N
unmineralized QBC schist (H/W)	10400E/10330N
magnetic QB schist ("I" anomaly)	9900E/11125N
magnetic chlorite-rich schist	9905E/11058H
magnetic QB schist ("I" anomaly)	9909E/11030N
magnetic QB schist ("I" anomaly)	9800E/11930N
magnetic QB schist ("I" anomaly)	10100E/11100N
magnetic Q8 schist ("I" anomaly)	10100E/11100N

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0G 144	magnetite guartzite	11900E/10220N
OG 145	magnetite quartzite	11820E/10220N
OG 146	magnetite quartzite	11740E/10225N
OG 147	magnetite quartzite	11650E/10200N
OG 148	magnetite quartzite	11525E/10175N
OG 149	magnetite quartzite	11475E/10175N
OG 150	magnetite quartzite	11425E/10170M
OG 151	magnetite quartzite	11100E/10150N
OG 152	magnetite quartzite	11050E/10150N
OG 153	magnetite quartzite	11000E/10135N
OG 154	magnetite quartzite	10950E/10135N
OG 155	magnetite quartzite	11150E/10150N
OG 156	magnetite quartzite (ES)	10850E/10150N
OG 157	magnetite quartzite (ES)	10830E/10150N
OG 158	magnetite quartzite (ES)	10810E/10150W
OG 159	magnetite quartzite (ES)	10790E/10150N

grid reference

				4				
				2				
OG 160	magnetite guartzite (ES)	10770E/10150N	OG 291	amphrich magnetite quartzite	8550E/9950N	OG 346	OGB 153 A chlor. amph. schist	355.57-356.30 H/W
OG 161	magnetite quartzite (ES)	10750E/10160N	OG 292	amphrich magnetite quartzite	8500E/9925N	OG 347	OGB 153 A T1 T2	356.30-356.45 H/W
OG 162	magnetite quartzite (ES)	10730E/10160N	OG 293	magnetite quartzite	8450E/9900N	OG 348	OGB 153 A T2	356.45-357.00 H/W
OG 163	magnetite quartzite (ES)	10710E/10160M	OG 294	magnetite quartzite	8400E/9900N	OG 349	OGB 153 A T1 T2	357.00-358.17 H/W
OG 164	magnetite quartzite (ES)	10690E/10160N	OG 295	magnetite quartzite	8350E/9900N	OG 350	OGB 153 A T2	358.17-358.375 H/W
OG 165	magnetite quartzite (ES)	10670E/10160N	OG 296	amphrich magnetite quartzite	8300E/9900N	OG 351	OGB 153 A chlor. amph. schist	358.375-359.18 H/W
OG 165	magnetite quartzite (ES)	10650E/10160N	DG 297	magnetite quartzite	8250E/9925N	OG 352	0G8 153 A T2	359.18-359.36 H/W
OG 167	magnetite quartzite (ES)	10530E/10170N	OG 298	amph rich magnetite quartzite	8200E/9950N	OG 353	OGB 153 A T2 (magn.)	359.36-359.53 H/W
OG 168	magnetite quartzite (ES)	10610E/10170N	OG 299	magnetite quartzite	8150E/9970N	OG 354	0GB 153 A T3	359.53-359.70 H/W
OG 169	magnetite quartzite	10525E/10175N	OG 300	amphrich magnetite quartzite	8100E/9970N	OG 355	OGB 153 A T1 T2 (magn.)	359.70-360.00 H/W
OG 170	magnetite quartzite	10400E/10180N	OG 301	magnetite quartzite	8050E/9980N	DG 355	OGB 153 A T1 T2	360.00-360.25 H/W
OG 171	magnetite quartzite	10250E/10170N	OG 302	magnetite quartzite	8000E/10000N	OG 357	OGB 153 A T2 (magn.)	360.25-360.685 H/W
OG 172	magnetite quartzite	10200E/10170N	OG 303	magnetite quartzite	7950E/10000N	OG 358	OGB 153 A T3	360,685-361.01 H.W
OG 173	magnetite quartzite	9960E/10040N	OG 304	magnetite quartzite	7920E/10000N	OG 359	OGB 153 A chlor. amph. schist	361.01-361.18 H/W
OG 174	magnetite quartzite	9910E/10045N	OG 305	magnetite quartzite	7850E/10000N	OG 360	OGB 153 A quartz vein	361.18-361.31 H/W
OG 175	magnetite quartzite	9860E/10060N	OG 306	magnetite quartzite	7800E/10000N	OG 361	OGB 153 A amphibolite	361.31-362.34 H/W
OG 176	magnetite quartzite	9820E/10070N	OG 307	magnetite quartzite	7750E/9970N	OG 362	OGB 153 A T1 T2	362.34-362.44 H/W
OG 177	magnetite quartzite	9770E/10100N	OG 308	magnetite quartzite	7700E/9950N	OG 363	OGB 153 A T2 T3	362.44-362.88 H/W
OG 178	magnetite quartzite (CS)	9640E/10100N	OG 309	amphrich magnetite quartzite	7650E/9940N	OG 364	. OGB 153 A 73	362.88-353.00 H/W
OG 179	magnetite quartzite (CS)	9620E/10095N	OG 310	amphrich magnetite quartzite	7600E/9920N	0G 365	0GB 153 A 11 12	369.27-370.26 F/W
OG 180	magnetite quartzite (CS)	9500E/10090N	OG 311	magnetite quartzite	7550E/9900N	06 365	0G8 153 A 11	3/0.20-3/0.74 F/W
OG 181	magnetite quartzite (CS)	9580E/10085N	OG 312	magnetite quartzite	7500E/9830N	00 307	000 153 A 12 13	371 30 372 36 5/4
0G 182	magnetite quartzite (CS)	9560E/10080M	0G 3)3	magnetite quartzite	7450E/9810N	00 300	000 153 6 13	372 36 372 57 5/4
06 183	magnetite quartzite (CS)	9540E/100/5N	06 314	magnetite quartzite	7400E/9800W	06 309	000 153 A 11 12	372.30-372.32 F/M
0G 184	magnetite quartzite (LS)	9520E/10075N	06 315	magnetite quartzite	73505/98000	06 370	000 153 A 11 12 000 153 A 13	373.00.374.00 F/W
06 185	magnetite quartzite (CS)	95002/100708	0G 316	magnetite quartzite	72502/95000	06 371	000 153 A 13	374 00-374 00 F/H
06 185	magnetite quartzite (CS)	94502/100/04	06 317	magnetite quartzite	72002/97800	00 372	000 153 A TI	374 00-375 00 E/W
06 187	magnetite quartzite (LS)	94002/100058	06 318	magnetite quartzite	71502/97600	00 375	000 153 4 11	375 00-375 79 F/W
06 188	magnetite quartzite (CS)	94402/100058	00 319	magnetite quartzite	71002/97701	00 375	008 153 4 13	375 70-377 50 E/W
06 189	magnetite quartzite (LS)	94202/100504	06 320	magnetite quartzite	70302/97408	00 375	000 153 4 11 12	377 60 377 82 E/W
DG 190	magnetite quartzite (CS)	0390C (10055N	06 321	magnetite quartzite	7000E/9720H	05 370	OCH 153 A T3	377 82-378 24 F/W
06 191	magnetite quartzite (CS)	9360E/10055N				00 377	0CB 153 A T1	378 24 370 DO E/W
00 192	magnetite quartzite (CS)	93002/100558		LCORES		00 370	OGB 153 A T2 T3	379 00-379 44 F/W
06 193	amphrich magn. quartzite (CS)	93402/100504	UNGONBU - UKII	LLUKES		06 380	OGB 153 A T1 72	379 44-379 60 F/W
00 194	magnetite quartrite (CS)	9300E /10045N				06 381	0CB 153 A 13	379.50-379.98 F/W
DG 195	amph _rich magn_nuartz(te (CS)	9280E/10045N	eamolo no	description	denth (metres)	OG 382	OGB 153 A T1	379,98-380,20 F/W
0G 197	magnetite quartzite (CS)	9260E/10045N	sample no.	description	uppen (meer ca)	OG 383	0GB 153 A T2	380.20-380.62 F/W
0G 198	magnetite quartzite (CS)	9240E/10050N	06 322	068 153 A T2 T3	343.20-344.40 H/W	0.000	The state of the state	22020.0 242024 4 4 4
OG 199	magnetite quartzite (CS)	9220E/10050N	0G 323	OGB 153 A TI	344.40-344.79 H/W	OG 384	0G8 151 A T2 T3	335.69-336.22 H/W
OG 200	magnetite quartzite (CS)	9200E/10050N	OG 324	OGB 153 A T2	344.79-345.89 H/W	OG 385	OGB 151 A T1 T2	336.22-336.48 H/W
0G 270	magnetite quartzite (CS)	9180E/10050N	0G 325	OGB 153 A T3	345.89-346.68 H/W	OG 386	OGB 151 A T2	335.48-340.39 H/W
OG 271	magnetite quartzite (CS)	9160E/10050N	OG 326	OGB 153 A T1	346.68-346.77 H/W	OG 387	OGB 151 A T1 T2	340.39-340.90 H/W
OG 272	amphrich magn. quartzite (CS)	9140E/10050N	OG 327	OGB 153 A T2 T3	346.77-346.87 H/W	OG 388	OGB 151 A T1	340.90-341.14 H/W
OG 273	amphrich magn. quartzite (CS)	9120E/10050N	OG 328	OGB 153 A T1 T2	345.87-347.35 H/W	OG 389	OGB 151 A chlor. amph. schist	341.14-341.61 H/W
OG 274	magnetite quartzite (CS)	9100E/10050N	OG 329	068 153 A T3	347.35-347.60 H/W	OG 390	OGB 151 A T1	341.61-342.82 N/W
OG 275	magnetite quartzite (CS)	9080E/10045N	OG 330	OGB 153 A T1 T2	347.60-348.35 H/W	OG 391	OGB 151 A T1 T2	342.82-345.67 H/W
OG 276	magnetite quartzite (CS)	9060E/10045N	OG 331	0G8 153 A T2 T3	348.35-349.14 H/W	OG 392	0G8 151 A T1	345.67-347.16 H/W
OG 277	magnetite quartzite (CS)	9040E/10045N	OG 332	OGB 153 A T1 T2	349.14-349.60 H/W	OG 393	OGB 151 A T1 (magn.)	347.16-347.20 H/W
DG 278	magnetite quartzite (CS)	9020E/10040N	OG 333	0G8 153 A T2	349.60/349.93 H/W	OG 394	OGB 151 A T2	347.20-347.25 H/W
OG 279	amphrich magn. quartzite (CS)	9000E/10040N	OG 334	OGB 153 A T3	349.93-350.27 H/W	OG 395	OGB 151 A chlor. amph. schist	347.25-347.66 K/W
06 280	magnetite quartzite (CS)	8980E/10040N	OG 335	OG8 153 A T1 T2	350.27-350.38 H/W	OG 395	OGB 151 A T2	347.66-348.23 H/W
OG 281	amphrich magn. quartzite (CS)	8960E/10040N	OG 336	OGB 153 A T3	350.38-350.56 H/W	OG 397	OGB 151 A chlor, amph. schist	348.23-348.35 H/W
OG 282	magnetite quartzite (CS)	8940E/10040N	OG 337	OGB 153 A T2	350.56-350.92 H/W	OG 398	OGB 151 A T1 T2	348.35-349.55 H/W
OG 283	magnetite quartzite (CS)	8920E/10045N	OG 338	OGB 153 A T1 T2	350.92-351.47 H/W	OG 399	OGB 151 A T1	349.55-350.00 H/W
OG 284	amphrich magn. quartzite (CS)	8900E/10050N	OG 339	OGB 153 A T3	351.47-352.33 H/W	OG 400	OGB 151 A T1 T2	350.00-351.41 H/W
OG 285	amphrich magnetite quartzite	8850E/10020N	OG 340	OGB 153 A T2	352.33-353.65 H/W	OG 601	OGB 151 A T2 T3	351.41-351.74 H/W
06 286	amphrich magnetite quartzite	8800E/9980N	OG 341	OGB 153 A T1	353.65-353.74 H/W	OG 602	OGB 151 A T3	351.74-351.83 H/W
OG 287	amphrich magnetite quartzite	8750E/99B0N	OG 342	OGB 153 A chlor. amph. schist	353.74-354.02 H/W	OG 603	OGB 151 A T1 T2	351.83-352.27 H/W
OG 288	magnetite quartzite	8700E/9980N	OG 343	OGB 153 A T1 T2	354.02-354.55 N/W	OG 504	OGB 151 A T3	352.27-352.46 H/W
OG 289	amphrich magnetite quartzite	8650E/99BON	OG 344	OGB 153 A chlor. amph. schist	354.65-354.71 H/W	OG 605	OGB 151 A T1 T2	352.46-353.09 H/W
OG 290	amphrich magnetite quartzite	8600E/9970N	OG 345	OGB 153 A feldspatic amph.	354.71-355.57 H/W	DG 605	OGB 151 A T3	353.09-353.56 H/W

OG 607	OGB 151 A T1 T2	353.56-354.00 H/W
OG 608	OGB 151 A chloramph. schist	354.00-355.31 H/W
OG 609	0G8 151 A T1	355.31-355.38 H/W
OG 610	OGB 151 A T2 T3	355.38-355.93 H/W
OG 611	OGB 151 A T2 T3	362.66-363.05 F/W
OG 612	OGB 151 A T3	363.05-363.89 F/W
OG 613	OGB 151 A T2	363.89-364.12 F/W
OG 614	OGB 151 A T1	364.12-364.87 F/W
OG 615	OGB 151 A T2 T3	364.87-365.05 F/W
OG 616	OGB 151 A T1 T2	365.05-365.40 F/W
OG 617	0GB 151 A T3	365.40-365.82 F/W
06 618	OGB 151 A T1	365.82-365.87 F/W
06 619	OGB 151 A T3	365.87-367.18 F/W
06 620	0GB 15) A TI	367.18-367.29 F/W
00 020	000 151 4 13	367 20-367 63 F/W
00 622	000 151 4 13	367 63-368 00 E/W
00 022		368 00.369 80 F/W
00 023	OGB 151 A sta usin t sulah	368 63 368 73 F/W
06 624		368.03-300.73 F/W
06 625	068 151 A 12	333 15 372 24 5/4
06 626	068 151 A 11 12	3/1.13-3/2.24 F/W
06 627	UGB 151 A 12 13	372.24-372.94 7/8
06 628	OGD ISLA II	372.94-3/3.44 7/1
06 629	068 151 A 11 12	AD2 83 403 70 H/U
00 630	DGD 154 A 12	492.03-493.70 1/1
06 631	DGB 154 A 12 13	403 83 404 54 H/W
06 632	OGB 154 A cirtor. ampi. scirist	455.05-454.54 H/H
06 632	OGB 154 A amphibolite	494. 34-493.47 N/H
06 633	DGB 154 A CHIOF, anph. Schise	405 71 A05 70 U/U
06 634	068 154 A 12	495.71-490.70 07 07
06 635	068 154 A 12 (magn.)	490.70-497.07 n/w
0G 636	0GB 154 A FA	497.07-497.12 h/w
OG 638	OGB 154 A 12 13	497.12-497.08 H/W
OG 639	OGB 154 A T1	497.68-498.50 H/W
OG 640	OGB 154 A chloramph. schist	498.50-498.62 H/W
OG 541	OGB 154 A T2	498.62-499.38 H/W
OG 642	OGB 154 A T2 (m)	499.38-499.56 H/W
OG 643	OGB 154 A amphibolite	499.56-501.15 H/W
OG 644	OGB 154 A amphibolite	501.15-502.64 H/W
OG 645	OGB 154 A chloramph. schist * T1	502.64-502.73 H/W
OG 646	OGB 154 A T1	502.73-503.23 H/W
OG 647	OGB 154 A T2 T3	503.23-503.47 H/W
OG 648	OGB 154 A T1 T2	503.47-503.83 H/W
OG 649	OGB 154 A chlor. amph. schist	503.83-503.90 H/W
OG 650	OGB 154 A T1 T2	503.90-504.27 H/W
OG 551	OGB 154 A T2 T3	504.27-504.53 H/W
OG 652	OGB 154 A T1 T2	504.53-505.36 H/W
OG 653	OGB 154 A T2 T3	505.36-505.79 H/W
OG 654	OGB 154 A T3	505.79-506.48 H/W
OG 655	OGB 154 A T2 T3	506.48-507.40 H/W
OG 656	OGB 154 A T1 T2	507.40-507.68 H/W
OG 657	OGB 154 A T3 -	507.68-507.74 H/W
OG 658	OGB 154 A T1 T2	507.74-507.89 H/W
OG 659	OGB 154 A T2 T3	507.89-508.82 H/W
OG 660	0GB 154 A T2	510.21-511.00 F/W
06 661	OGB 154 A T2 T3	511.00-511.71 F/W
06 662	0G8 154 A T1 T2	511.71-512.04 F/W
06 663	0G8 154 A T2 T3	512.04-512.19 F/W
00 554	0C8 154 A T	517,19-513 40 F/W
00 565	0CG 154 A T2	513.40-514 49 F/W
00 005	000 104 0 12 12	514 49-515 48 5/4
00 000	000 104 M 12 10	515 48-516 ON E/U
06 007	000 104 A 11	515 00 517 10 C/W
06 668	UGB 154 A 12	510.00-517.10 F/W

OG 669	068 154 A	73	517.18-517.41 F/W
OG 670	OGB 154 A	TI	517.41-517.66 F/W
OG 671	OGB 154 A	T1 T2	517.66-517.91 F/W
OG 572	OGB 154 A	T2 T3	517.91-518.95 F/W
OG 673	OGB 154 A	T1 T2	518.95-519.29 F/W
OG 674	OGB 154 A	T3	519.29-521.15 F/W

MATCHLESS WEST EXTENSION DEPOSIT

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sample no.	le no. description :		grid reference	
MWE 201	QMP schist	246/-5.5		
MWE 202	QMP schist	246/-19		
MWE 203	qtz-rich gossan	243/10		
MWE 204	muscovite chlorite schist	240/-18		
MHE 205	QMP schist	240/-13		
MWE 205	QMP schist	235/-6		
HWE 207	talc-rich gossan	235/10		
MWE 208	magnetite quartzite	233/9.5		
MHE 209	QMP schist	230/-11		
MWE 210	QMP schist	221/-3.5		
MWE 211	muscovite chlorite schist	224/1		
MWE 212	qtz-rich gossan	218/10		
MHE 213	qtz-rich gossan	208/6		
MHE 214	QMP schist	208/-5		
MWE 215	muscovite chlorite schist	210/-8		
MWE 216	QMP schist	200/2.5		
HWE 217	QMP schist	200/-3.5		
MHE 218	gtz-rich massive gossan	192/6		
MWE 219	OMP schist	190/5		
MWE 220	OMP schist	191/-1		
MWE 221	OMP schist	180/1		
MWE 222	otz-rich massive gossan	179/10		
MWF 223	magnetite quartzite	179/12.5		
MWF 224	OMP schist	171/6		
HUF 225	magnetite quartzite	170/8.5		
MWE 226	OMP schist	158/1		
MWF 227	talc-actinolite schist	158/6		
MWF 228	magnetite quartzite	160/8		
MWF 229	OMP schist	147/1	-	
MUE 230	magnetite quartzite	150/8		
MWE 231	OMP+talc-actin, schist	140/6.5		
MWF 232	OMP schist	138/3.5		
MUE 233	gossan talc_actinolite schist	130/3.5		
MUE 234	OMP schist	131/1		
HUF 235	OMP schist	123.3		
MUE 236	OMP+actinolite schist	123/4		
MUE 237	OMP schist	110/2.5		
MUE 238	OMP schist	103/1.5		
MUE 230	chlor -actin -tale schist	103/3.5		
MUE 200	magnetite quartzite	98/4		
MUE 240	amph wich magnetite quartzite	05/3 5		
MUC 242	magnetite quartzite	90/3 5		
MUE 242	chlorite actinglite schirt	80/2.5		
MIE 243	magnetite quantaite	80/0		
TINE 244	magnetite quartzite	93/3 5		
MHE 245	tale-actincarp. schist	70 5/3.5		
MHE 246	tale-actincarb. Schist	10.5/3.5		
MWE 247	magnetite quartzite	81/5		
MWE 248	gossan talc-actin. schist	69/3		

HE.	249	amphibrich magnetite quartzite	72/6	
ME	250	magnetite quartzite	72/7	
WE	251	magnetite quartzite	61/3	
WE	252	gossan, talc-chlor, schist	61/4	
WE	253	magnetite quartzite	58/8	
HE	254	magnetite quartzite	51/6	91
WE	255	gossan. talc-actincarb. schist	50/7.5	
1HE	256	magnetite quartzite	53/11.5	
INE	257	magnetite quartzite	44/6.5	
WE.	258	gossan. talc-actinchlor. schist	42/10	
WE	259	marble-like rock	17/13.5	
HE	260	magnetite quartzite	17/15	
AB	261	Friedenau amphibolite	see attached may	p
AB	262	Friedenau amphibolite	see attached may	p
AB	263	Friedenau amphibolite	see attached may	p
AB	264	Matchless Mine Horizon amphibolite	see attached may	p
AB	265	Matchless Hine Horizon amphibolite	see attached may	p
BAN	266	Matchless Hine Horizon amphibolite	see attached may	р
MAB	267	Friedenau amphibolite	see attached may	p
MAB	268	Friedenau amphibolite	see attached may	P
MAB	269	Friedenau amphibolite	see attached may	р

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MAFIC AND ULTRAMAFIC ROCKS FROM THE OTJIHASE AREA

description	location
talc-actin. schist (coarse-gr.)	Otjihase
talc-actin. schist (fine-gr.)	Otjihase
metagabbro	Otjihase
talc-carb. schist	Elisenheim
talc-carbchlor. schist	Elisenheim 68
chlor. actin. schist	Elisenheim 68
talc schist	Elisenheim 68
chlorite schist	Otjihase
epidrich chlorite schist	Otjihase
actinolite schist	Von Francois Ost 6
chlortalc schist	Von Francois Ost 6
talc-olivine(7)-magn. schist	Von Francois Ost 6
	description talc-actin. schist (coarse-gr.) talc-actin. schist (fine-gr.) metagabbro talc-carb. schist talc-carb. schist talc-carb. schist chlor. actin. schist talc schist epidrich chlorite schist actinolite schist talc-olivine(7)-magn. schist

OG 669 OGB 154 A 13 517.18-517.41 F/W OGB 154 A TI 517.41-517.66 F/W OG 670 517.66-517.91 F/W OG 671 OGB 154 A T1 T2 OGB 154 A T2 T3 517.91-518.95 F/W OG 672 518.95-519.29 F/W OG 673 OGB 154 A T1 T2 519.29-521.15 F/W OG 674 OGB 154 A T3

MATCHLESS WEST EXTENSION DEPOSIT

			UG 018	UGB 151 A 11
sample no.	description	grid reference	OG 619	OGB 151 A T3
			OG 620	OGB 151 A T1
MWE 201	QMP schist	246/-5.5	OG 621	OGB 151 A T3
MHE 202	QMP schist	246/-19	DG 622	OGB 151 A T1 T2
MWE 203	qtz-rich gossan	243/10	0G 623	OGB 151 A T3
MHE 204	muscovite chlorite schist	240/-18	OG 624	OGB 151 A qtz vein + sulp
HWE 205	QMP schist	240/-13	OG 625	OGB 151 A T2
MWE 206	QMP schist	235/-6	OG 626	OGB 151 A T1 T2
MWE 207	talc-rich gossan	235/10	OG 627	OGB 151 A T2 T3
MWE 208	magnetite quartzite	233/9.5	DG 628	OGB 151 A T1
MWE 209	QMP schist	230/-11	0G 629	OGB 151 A T1 T2
MHE 210	QMP schist	221/-3.5	OG 630	OGB 154 A T2
MHE 211	muscovite chlorite schist	224/1	OG 631	OGB 154 A T2 T3
MHE 212	qtz-rich gossan	218/10	0G 632	OGB 154 A chlor. amph. sc
MWE 213	qtz-rich gossan	208/6	OG 632	OGB 154 A amphibolite
MHE 214	QMP schist	208/-5	OG 633	OGB 154 A chlor. amph. sc
MHE 215	muscovite chlorite schist	210/-8	0G 634	OGB 154 A T2
MHE 216	QMP schist	200/2.5	OG 635	OGB 154 A T2 (magn.)
MWE 217	QMP schist	200/-3.5	DG 636	0GB 154 A FA
MHE 218	gtz-rich massive gossan	192/6	06 638	0GB 154 A T2 T3
MHE 219	OMP schist	190/5	06 639	DG8 154 A T1
MWE 220	OMP schist	191/-1	DG 640	OGB 154 A chlor -amph. sc
MWE 221	OMP schist	180/1	06 641	0GB 154 A T2
MWE 222	gtz-rich massive gossan	179/10	05 542	OCE 154 A 12 (m)
HHE 223	magnetite quartzite	179/12.5	06 643	OGB 154 A amphibolite
MHE 224	OMP schist	171/6	06 544	DGB 154 A amphibolite
MHE 225	magnetite quartzite	170/8.5	00 645	OCB 154 A chlor -amb sc
MWE 226	OMP schist	158/1	00 645	
MHE 227	talc-actinolite schist	158/6	00 647	OCB 154 A T2 T3
HWE 228	magnetite quartzite	160/8	00 649	000 154 A TE TS
HWE 229	OMP schist	147/1	00 640	OCB 154 A chlor amph so
HWE 230	magnetite quartzite	150/8	00 049	OCB 154 A T1 T2
HWE 231	OMP+talc-actin, schist	140/6.5	00 050 00 651	OCB 154 A T2 T3
HWE 232	OMP schist	138/3.5	00 652	OGB 154 A T1 T2
NHE 233	nossan talc-actinolite schist	130/3.5	00 653	OCB 154 A T2 T3
HUF 234	OMP schist	131/1	00 654	000 154 A 12 15
MUE 235	OMP schist	123 3	00 655	000 154 A TO TO
MUE 236	OMP+actionlite schist	123/4	00 055	000 154 A T1 T2
MUE 237	OMP schist	110/2.5	00 050	OCB 154 A TI TZ
MUE 238	OND schist	103/1 5	00 03/	008 154 A 13
HUE 230	chlor _actin _tale schist	103/3 5	06 030	000 154 A 11 12
MUE 240	magnetite quartzite	98/4	06 059	DGB 154 A 12 15
MUE 241	amph rich magnetite quartzite	95/3 5	06 660	068 154 A 12
MUE 242	magnetite quartzite	90/3 5	06 001	068 154 A 12 13
MUE 243	chlorite actinolite schist	89/2 5	06 662	UGB 154 A 11 12
MUE 244	magnatite quartaite	80/0	UG 663	U68 154 A 12 13
LAUE 244	tale actin carb schict	83/3 5	OG 664	OGB 154 A 11
MUE 240	tale actin carb, schist	78.5/3.5	OG 665	068 154 A T2
THE 240	ta te-actini-carp, schist	81/5	OG 666	OGB 154 A T2 T3
PINE 247	magnetite quartzite	61/3	OG 667	OGB 154 A T1
MWE 248	gossan talc-actin. schist	L/60	0G 668	DGB 154 A T2

OG 607	OGB 151 A T1 T2	353.55-354.00 H/W
DG 608	OGB 151 A chloramph. schist	354.00-355.31 H/W
OG 609	0GB 151 A T1	355.31-355.38 H/W
OG 610	DGB 151 A T2 T3	355.38-355.93 H/W
OG 611	OGB 151 A T2 T3	362.66-363.05 F/W
OG 612	OGB 151 A T3	363.05-363.89 F/W
06 613	068 151 A T2	363.89-364.12 F/W
00 614		364 12 364 B7 E/W
00 615		364 87 366 AF E/U
00 015	DGB 151 A TZ TJ	304.07-305.03 F/H
06 010	068 151 A 11 12	305.05-305.40 F/W
0G 61/	0GB 151 A 13	305.40-305.82 F/W
OG 618	OGB 151 A T1	365.82-365.87 F/W
OG 619	OGB 151 A T3	365.87-367.18 F/W
OG 620	OGB 151 A T1	367.18-367.29 F/W
OG 621	OGB 151 A T3	367.29-367.63 F/W
OG 622	OGB 151 A T1 T2	367.63-368.00 F/W
OG 623	OGB 151 A T3	368.00-368.89 F/W
OG 624	OGB 151 A qtz vein + sulph.	368.63-368.73 F/W
OG 625	OGB 151 A T2	368.83-371.15 F/W
0G 626	0GB 151 A T1 T2	371.15-372.24 F/W
DG 627	OGB 151 A T2 T3	372.24-372.94 F/W
06 628	068 151 A T1	372.94-373.44 F/W
06 629	0GB 151 A T1 T2	373.44-375.32 F/H
00 630	OCB 154 A T2	492 83-493 70 H/W
00 631	OCR 154 A T2 T3	403 70.403 83 H/H
NC 632	OCR 154 A chlor amph schiet	403 83 404 54 H/H
00 032	OCD 154 A ciribit, anpit, scirist	404 54 405 47 1/1
06 032	DGB 154 A amphibolite	494.54-495.47 8/8
0G 633	UGB 154 A Chior, amph. schist	495.47-495.71 H/W
0G 634	OGB 154 A 12	495./1-496./0 H/W
OG 635	OGB 154 A T2 (magn.)	495.70-497.07 H/W
OG 636	OGB 154 A FA	497.07-497.12 H/W
DG 638	OGB 154 A T2 T3	497.12-497.68 H/W
OG 639	OGB 154 A T1	497.68-498.50 H/W
DG 640	OGB 154 A chloramph. schist	498.50-498.62 H/W
OG 641	OGB 154 A T2	498.62-499.38 H/W
OG 642	OGB 154 A T2 (m)	499.38-499.56 H/W
OG 643	OGB 154 A amphibolite	499.56-501.15 H/W
OG 544	OGB 154 A amphibolite	501.15-502.64 H/W
0G 645	OGB 154 A chloramoh. schist + T1	502.64-502.73 H/W
06 646	DGB 154 A TI	502 73-503 23 H/W
DG 647	OCR 154 A T2 T3	503 23-503 47 H/W
00 649	000 154 A 12 13	503.23-303.47 H/H
00 040	OCD 154 A shier such schiet	503.47-503.05 N/H
00 049	OGD 154 A CHIOL MINT. SCHIST	503.03-503.90 H/H
06 650	068 154 A 11 12	503.90-504.27 H/W
05 051	Uud 154 A 12 13	504.27-504.53 H/W
UG 652	UGB 154 A 11 12	504.53-505.36 H/W
OG 653	OGB 154 A T2 T3	505.36-505.79 H/W
OG 654	OGB 154 A T3	505.79-506.48 H/W
OG 655	OG8 154 A T2 T3	506.48-507.40 H/W
OG 656	OGB 154 A T1 T2	507.40-507.68 H/W
OG 657	OGB 154 A T3	507.68-507.74 H/W
OG 658	OGB 154 A T1 T2	507.74-507.89 H/W
DG 659	OGB 154 A T2 T3	507.89-508.82 H/W
OG 660	OGB 154 A T2	510.21-511.00 F/W
06 661	0GB 154 A T2 T3	511.00-511.71 F/W
06 662	OGB 154 A T1 T2	511.71-512.04 E/W
DC 563	OCR 154 A T2 T3	512 04-512 19 F/N
00 003	000 104 A 12 10	512 10 512 AD FA
00 004	Uud 154 A (1	512.19-513.40 F/W
04 005	Uub 154 A 12	513.40-514.49 F/W
OG 666	OGB 154 A T2 T3	514.49-515.48 F/W
OG 667	OGB 154 A T1	515.48-516.00 F/W

516.00-517.18 F/W

amphibrich magnetite quartzite	72/6
magnetite quartzite	72/7
magnetite quartzite	61/3
gossan. talc-chlor. schist	61/4
magnetite quartzite	58/8
magnetite quartzite	51/6
gossan. talc-actincarb. schist	50/7.5
magnetite quartzite	53/11.5
magnetite quartzite	44/6.5
gossan. talc-actinchlor. schist	42/10
marble-like rock	17/13.5
magnetite quartzite	17/15
Friedenau amphibolite	see attached map
Friedenau amphibolite	see attached map
Friedenau amphibolite	see attached map
Matchless Mine Horizon amphibolite	see attached map
Matchless Mine Horizon amphibolite	see attached map
Matchless Mine Horizon amphibolite	see attached map
Friedenau amphibolite	see attached map
Friedenau amphibolite	see attached map
Friedenau amphibolite	see attached map
	amphibrich magnetite quartzite magnetite quartzite gossan. talc-chlor. schist magnetite quartzite gossan. talc-chlor. schist magnetite quartzite gossan. talc-actincarb. schist magnetite quartzite gossan. talc-actinchlor. schist magnetite quartzite friedenau amphibolite friedenau amphibolite Friedenau amphibolite Matchless Mine Horizon amphibolite Friedenau amphibolite Friedenau amphibolite Friedenau amphibolite Friedenau amphibolite Friedenau amphibolite Friedenau amphibolite

MAFIC AND ULTRAMAFIC ROCKS FROM THE OTJIHASE AREA

Υ.

sample no.	description	location
0TJ 675	talc-actin. schist (coarse-gr.)	Otjihase
OTJ 676	talc-actin. schist (fine-gr.)	Otjihase
OTJ 677	metagabbro	Otjihase
OTJ 678	talc-carb, schist	Elisenheim
013 679	talc-carbchlor. schist	Elisenheim 68
01J 680	chlor. actin. schist	Elisenhelm 68
0TJ 681	talc schist	Elisenheim 68
01J 682	chlorite schist	Otjihase
0TJ 683	epidrich chlorite schist	Otjihase
OTJ 684	actinolite schist	Von Francois Ost 6
0TJ 685	chlortalc schist	Von Francois Ost 6
0TJ 686	talc-olivine(?)-magn. schist	Von Francois Ost 6



Location of the meta-lava samples. See legend in Fig. 12.

APPENDIX D

LIST OF THE THIN AND POLISHED SECTIONS

Sample no.	Location	Grid reference or depth (in drill holes)	Type of sectio	on Remarks	Sample no.	Location	Grid reference or depth (in dril) ho
06 2	Origeama gossan	11060E/10240N	P/S	qtz-rich massive gossan	DG 710A	Ongombo - DH 0GB 153A	360.25-360.685
06 15	Ongeama gossan	10900E/10258.2N	P/S	gtz-rich massive gossan	06 7108	Orgonbo - DH 0GB 153A	360.25-360.685
06 17	Ongeama gossan	10880E/10260.3N	1/5	magnetite quartzite	06 711	Ongombo - Dil 0GB 153A	360.685-361.01
0G 19A	Ongeana gossan	10840E/10271N	T/S,P/S	magnetite quarizite	06 712	Ongombo - DH 0GB 153A	361.01-361.18
06 198	Ongeana gossan	10840E/10271N	1/5,P/S	magnetite quartzite	06 713	Ongombo - DH 0GB 153A	361,31-362.34
06 25	Ongeama gossan	10720E/10253.3N	1/5	ferruginous schist	66 714	Ongorrbo - DH 0GB 153A	362.34-362.44
06 38	Ongeama gossan	10420E/10220N	P/S	qtz-rich massive gossan	51/ 50	Ongombo - Dit 0GB 153A	362,44-362,88
06 47	Ongeama gossan	10360E/10206N	1/5, P/S	qtz-rich massive gossan	91/ 90	Ongombo - DH OGB 153A	362.68-363.00
DC 03	Ungeama gossan	10260E/1019/M	2/4.6/1	amphrich sem!-mass. gossan	91/ 90	ALCL GUT UN - DOMOROO	00.000-00.000
50 20	Onoreana possan	10180E /1017BN	1/5	magnetic chloritoid-rich chloratz schist	06 220	Oncombo - DH 0GB 153A	363.56-364.46
82 50	Ongeams gossan	10180E/10166N	1/5. P/S	magnetic chlormuscovite schist	12/ 50	Ongombo - DH 0CB 153A	363.56-364.46
06 135	Oncease grant	10290E/10255N	1/5	magnetic chloritoid-rich chlordtz schist	06 722	Ongombo - DH 0GB 153A	364,46-364,82
06 144	Ongombo gossan	11900E/10220N	P/S	sulphrich magnetite quartzite	06 723	Ongombo - DH 0GB 153A	364,46-364.82
DG 192A	Ongombo gossan	9360E/10055N	1/5	magnetite quartzite close to 06 1928	06 725	Ongombo - DH 0GB 153A	364.46-364.82
06 1928	Ongontoo gossan	9360E/10055N	7/5, P/5	amphrich gossan	06 726	Ongombo - DH 0GB 153A	364,46-364,82
00 196	Drigombo gossan	9280E/10045N	S/d'S/1	amphrich magn. quartzite	0G 727A	Ongoorbo - DH 0GB 153A	364.46-364.82
MIE 204	MME deposit	240/-18	1/5	muscovite-chlorite schist	06 /2/8	Ongombo - DH 0GB 153A	364.82-364.95
NAE 207	HHE deposit	235/10	1/5, P/5	taic-bearing gossan	06 720	Ongomod - DH ULB 153A	\$1.505-66.905
MAE 209	Mult deposit	C.E/CC3	5/1	SulpaFicn maga. quartille DMP schist	06 730	Ongombo - DH DGH 153A	365.14-365.37
Hole 217	MKE deposit	200/-3.5	1/5	OMP schist	06 731	Oncombo - DH 0GB 153A	365.14-365.37
MAE 236	MWE deposit	123/4	1/5	actintalc-baryte-qtz schist	06 732	Ongombo - DH 0GB 153A	365.37-365.56
MHE 239	MME deposit	103/3.5	1/5, P/5	chloractin. schist	06 733	Ongombo - DH OGB 153A	365.37-365.56
1HE 241A	MWE deposit	95/3.5	1/5, P/S	amphrich magnetite quartzite	96 734	Ongomba - DH OGB 153A	365.37-365.56
ME 2418	MWE deposit	95/3.5	1/5	amphrich magnetite quartzite	567 30	Ongombo - DH 0GB 153A	365.56-366.05
HHE245A	MKE deposit	83/3.5	1/5	talc-actinbaryte quartz	06 /30	Ungombo - DH UGB 153A	365.56-366.05 Sec 15 355 AT
100 240 240	THE DEPOSIT	03/3-5 76 5/3 5	5/1	Actinolite schist	067 20	ACCT 000 RU - 0000000	20,000-00,000
MAE 253	MME deposit	58/8	1/5. P/S	magnetite quartzite	06 740	Ondombo - DH 0GB 153A	366.05-366.35
MAE 259	MME deposit	17/13.5	1/5	marble-like rock	06 741	Ongombo - DH 0GB 153A	366.05-366.35
MME 261	HMH		1/5	Friedenau amphibolite	06 743	Ongombo - DH 0GB 153A	366.35-366.54
MHE 264	HHH		1/5	WHH amphibolite	. 06 744	Ongombo - DH 0GB 153A	366.54-367.00
00 281	Ongombo gossan	B960E/10040N	1/S.P/S	amphrich magn. quartzite	06 745	Ongombo - DH 0GB 153A	366.54-367.00
06 636	Ongombo - DH 0GB 154A	497.07-497.12	1/5	12 (magn.) schist (F/H of mafic unit)	06 745	Ongombo - DH 0GB 153A	367.00-367.83
00 682	Unguation - Uni Uuti 1344	11. #5#-1.05#	2112	HI SCHIST (F/H and H/H OT MATIC UNIT)	00 748	Andrew - DR NCR 1534	02 835 58 735
06 684	Von Francois Ost 60		1/5	matic metalava actinolite schist	06 749	Discrete - DH 0GB 153A	367.83-368.29
06 685	Von Francois Ost 60		1/5	chlorite-talc-carb. schist	06 750	Ondombo - DH 0GB 153A	368.48-368.87
G 686 (A+B)	Von Francois Ost 60		2 1/5	talc-carbmagn. schist	06 751	Ongombo - DH 0GB 153A	368.48-368.87
06 688	Ongontio Prospect	377.56-377.86	1/5	T1 schist (F/W of mafic unit)	06 752	Dingombo - Dil 0GB 153A	368.87-369.25
06 690	Dingonitio - DH DGB 153A	267.41-267.84	S/1	TI (magn.) schist (H/H of mafic unit)	06 754	Ongombo - DH 0GB 153A	368.87-369.25
06 691	Ongombo - DH 0GB 135A	77.30-77.40	1/5	magnetic amphibolite	06 755	Ongombo - DH 0GB 153A	325.83-325.95
269 503	Ongombo - DH 068 131C	101.35-103.48	1/5	hbl-rich T1 (magn.) schist	06 /56	Ongombo - DH 0GB 153A	326.07-326.17
060 000 DC 600	VIST BOO HO - OCUDEND	CU. THE	SIT	II SCHISE (N/M DI MAITC UNIC) Ti schict (N/N of mafte unit)	10/ 50	VICT ONN UN - NONNAMIN	04.201-02.201
06 695	Ongombo - DH 0GB 151A	347.55-347.60	1/5	chlorite-amphibole schist			
06 696	Ongombo - DH 0GB 151A	357.90	1/5	T2 schist (F/W of mafic unit)		Abhreviations use	d in the table.
269 50	Ongombo - DH 0GB 151A	348.00	1/5	T1 (magn.) schist (F/H of mafic unit)			
06 698	Ongombo - DH DGB 151A	348.48	1/5	II schist (H/M of mafic unit)		MME - Matchless M	est Extension
669 50	Ongortho - DH OGB 151A	348.60	1/5	T1 (magn.) schist (F/H of mafic unit)		HMH - Matchless M	ine Horizon amphibolite
00/ 00	Area - DH UGB 1534	00.666	1/5	metagabbro		DH - drillhole	
06 704	Acct an nu - unuguin Acct an 068 153A	11.000-100	5/1	11-12 schist T2 schist		T/S - thin section	
06 705	Orgombo - DH 0GB 153A	359.18-359.36	1/5	T1-T2 schist (close to thin mafic unit)		T1 - macelyan set	stron -tr biotite schiet
06 706	Ongombo - DH 0GB 153A	359.36-359.53	5/1	T2 (magn.) schist		T2 - foliated que	artz-biotite-miscovite-ch
06 707	Orgombo - DH 0GB 153A	359.53-359.70	1/5	T3 schist		T3 - contorted blo	of ite-muscovite-chlorite-
06 708	Ongombo - DH 0GB 153A	359.70-360.00	1/5, P/5	T1-T2 (magn.) schist		remob remobility	red
602 200	Ongombo - DH UGB 153A	360.00-360.25	1/5	TI-T2 (magn.) schist		diss disseminat	ted
						semi-mass, * semi-	-mass fve

ple no.	Location	Grid reference I or depth (in drill holes)	ype of sect	on Remarks
7104	Ongombo - DH OGB 153A	360.25-360.685	7/S,P/S	T1-T2 (magn.) schist
1108	Orgonbo - DH 0GB 153A	360.25-360.685	1/5	qtz-chloramph. schist (close to mafic uni
6 711	Ongombo - DH 0GB 153A	360.685-361.01	5/1	T3 schist (H/M of malic unit)
6 712	Ongombo - DH OGB 153A	361.01-361.18	1/5	chlorite-amphtalc schist
6 713	Ongombo - DH 0GB 153A	361.31-362.34	1/5	metagabbro
6 714	Ongombo - DH 0GB 153A	362.34-362.44	1/5	TI-T2 schist (F/W of mafic unit)
5 715	Ongombo - Dit 0GB 153A	362,44-362,88	5/1	T2-T3 schist (staurrich)
6 716	Ongombo - DH 0GB 153A	362.68-363.00	1/5	13 schist (staurrich)
9179	Ongombo - DH UGB 153A	363.48-363.56	1/5, P/S	biotite selvedge zone
611 9	Ongombo - DH 0GB 153A	363.55-364.45	5/1	biot. schist, diss. sulph.
6 720	Ongombo - DH 0GB 153A	363.56-364.46	S/4'S/1	sulphrich lenses in biot. schist
17/ 5	Ungonbo - UH UGB 153A	303.55-364.46	5/1	biot. schist
222 9	Ongombo - DH 0GB 153A	364.46-364.82	P/S	sulphrich magnetite quartzite
6 723	0ngomba - DH 0GB 153A	364,46-364.82	5/d	remob. sulph. (qtz vein)
52/ 5	Ongombo - DH UGB 153A	364.46-364.82	5/1	semi-mass. sulph. bands
07/ 5	Ungonto - UH UGB 153A	354,45-354,82	P/S	semi-mass. suiph, bands
8171	VEGT BOO HI - CONDUN	20,400-04,400	5/1	DiotFich schist (smeared suiph.)
0171	VICE I DO HI - OCUDIU	66.902-304.90C	5/1	DIOL STAUT TICH SCHISE (STREATED SULDA.)
0 120	VEST GIN UN - DOMORIU	P1.000-00.000	5/1	semi-mass. suipn. pards
6 730	Onorotion - DH DGH 153A	16.365.14-365	SIL	bint staur subli- usids
6 731	Onoombo - DH 0GB 153A	365.14-365.37	1/5	blotstaur, schist
G 732	Ondombo - DH 0GB 153A	365.37-365.56	1/5	biotite-blao. schist
5 733	Ongombo - DH 0GB 153A	365.37-365.56	5/4	seni-mass, sulph, hands
5 734	Ongomba - DH OGB 153A	365.37-365.56	1/5	semi-mass. sulph. bands
G 735	Ongombo - DH 0GB 153A	365.56-366.05	7/S.P/S	plagbiot. schist (diss. sulph.)
G 737	Ongombo - DH 0GB 153A	365.56-366.05	P/S	seni-mass. sulph. bands
6 738	Ongombo - DH 068 153A	365.56-366.05	1/5	qtz-chlor. schist (diss. sulph.)
6E1 3	Ongombo - DH 0GB 153A	366.05-366.35	1/5	biot. selvedge zone (diss. sulph)
6 740	Ongombo - DH 0GB 153A	366.05-366.35	1/S	qtz-chlor. schist
5 741	Ongombo - DH 0GB 153A	366.05-366.35	P/S	semi-mass. sulph. bands
6 743	Ongombo - DH 0GB 153A	366.35-366.54	1/5	chlorblotmusc. schist
5 744	Ongombo - DH 0GB 153A	366.54-367.00	P/S	blot. schist (diss. sulph.)
2 745	Ongombo - DH OGB 153A	366.54-367.00	1/5	biot. schist (diss. sulph.)
2 746	Ongombo - DH 0GB 153A	367.00-367.83	1/5	blotchlor. schist (diss. sulph.)
141	Ongombo - DH 0GB 153A	367.83-368.29	P/S	semi-mass. sulph, bands
5 748	Ongombo - DH 0GB 153A	367.83-368.29	P/S	sulphrich magn. quartzite
2 749	Ongombo - DH 0GB 153A	367.83-368.29	1/5	magnetite quartzite
2 750	Ongombo - DH 0GB 153A	368.48-368.87	P/S	biotite selvedge zone
151	Orgombo - DH 0GB 153A	368.48-368.87	1/5	biotite selvedge zone
2 752	Digombo - Dil 0GB 153A	368.87-369.25	1/S	biotchlor schist (apatite)
154	Ongombo - DH 0GB 153A	368.87-369.25	2/1	TI (magn.) schist
992	Annowick Put not team	CF.C2C-CO.C2C	2/1	12 schist
151 5	Oncombo - DH 0GB 151A	182.26-182.48	5/1	calc-silicate laver
	Abhreviat fons use	d in the tablet		
	MVE - Matchless Me	est Extension		sulph sulphide
	HMH - Hatchless M	ine Horizon amphibolite		qtz - quartz
	DH - drillhole			biot biotite
	I/S - thin section			magn magnetite
	P/5 - polished sec	ction		actin actinolite
	T7 - follotellot	riz-biotice scilist		hbi - hornblende
	12 - 101 laten yes	artz-biotite-miscovite-chiorite	SCRIST	staur, - staurolite
	11 - CUITOL EL	of ite-muscovite-chiorite-quartz	schist	plag plagioclase

APPENDIX E

Tables of the geochemical data of

mineralized and unmineralized rock types

*	Table	1	-	Lower detection limits for trace element analysis.
*	Table	2	-	Ongeama Prospect: trace-element geochemistry of the gossan exposure.
*	Table	3	-	Matchless West Extension deposit: trace-element geochemistry of the gossan exposure.
*	Table	4	•	Ongombo Prospect: trace-element geochemistry of the gossan exposure.
*	Table	5	•	Ongombo Prospect: trace-element geochemistry of the ore-bearing sequence in drill cores.
.*	Table	6		Ongombo Prospect: trace-element geochemistry of the ore-zone rock types.
*	Table	7		Trace-element geochemistry of the unmineralized schists.
*	Table	8	3	Ongeama Prospect: trace-element geochemistry of the magnetic schists from the "I anomaly" area.
*	Table	9	1	X-Ray fluorescence analytical conditions and rocks standards used.
*	Table	10	-	Ongeama Prospect: major element geochemistry of selected samples from the alteration pipe.
*	Table	11		Matchless West Extension deposit: major element geochemistry of selected samples from the alteration pipe.
*	Table	12		Major element geochemistry of unmineralized metapelitic schists (Kuiseb Formation).
*	Table	13	4	Subdivision in sectors of the alteration pipe at Ongeama: average of the trace element content for each sector.
*	Table	14	1	Subdivision in sectors of the alteration pipe at Matchless West Extension: average of the trace element content for each sector.
*	Table	15		Ongeama Prospect: barium and zinc content in the schists of the alteration pipe.
*	Table	16	a	-g - Alteration index for trace elements (I trace) and major element (I major) TiO ₂ /Zr index (TZ) calculated for unmineralized and ore-related rock types
*	Table	17	-	Trace element geochemistry from the Alpine-type Damaran ultramafic rocks from the Otjihase Mine area and Elisenheim.
*	Table	18	-	Major element geochemistry of ultramafic rock types.
*	Table	19		Concentrations of selected elements in different ultramafic rock types from Matchless West Extension, Ongombo and Otjihase, and values of the Ex and Um

indices.

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element	detection limit (ppm)	element	detection limit (ppm)
Си	5	Cr	2
Pb	5	Мо	10
Zn	5	Zr	10
Со	5	Nb	10
Ni	5	Y	10
Ag	0.5	Rb	10
Au	40 ppb	Sr	10
Mn	50	Bi	10
Ba	25	W	10
v	25		
Ti02	25		

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Table 1 - Lower detection limits for trace element analyses.

Abbreviations used in the table:

Table 2 - (Note: ND)	Ongeama means "	Prospect: t below detect	trace elem tion limit	ent geoch *).	enistry o	f the gos	san expos	ure.			MS - MQ - AS - S	massive s magnetite amphibole ulphide g	ulphide go quartzite -rich semi jossan	ossan e i-massive	FS - f HM - m CS - m	erruginou agnetic m agnetic c	s schist uscovite : hlorite se	schist chist					
sample no.	, and	Cu	РЬ	Zn	Ni	Co	Ag	Au	Mn	Ва	Cr	v	T102	As	H03	Mo	Sn	Zr	Rb	Sr	Bi	ND	Y
rock type			ppm	ppm	bbw	bba	ppm	ppb	ppm	ppa	ppm	ppm	ppre	P.P.	ppa	p.p.u.	phu.	ppn	Ppm	ppa	ppm	ppn	P.Fen
OG 2	MS	0.33	88	388	36	34	21	144	482	274	35	40	1086	29	NO	ND	ND	ND	ND	ND	ND	ND	ND
DG 3	MS	0.28	87	176	34	32	18	168	401	296	37	108	2302	ND 57	ND	ND	ND	B	ND	16	ND	ND	ND
OG 4	KS	0.24	01	23/	29	31	22	139	482	286	44	62	1080	ND	ND	ND	28	ND	ND	ND	ND	ND	ND
06 5	MS	0.25	114	153	28	26	21	257	321	300	47	RR	1996	14	ND	ND	ND	20	ND	26	ND	ND	ND
06 7	FS	0.22	96	130	29	30	12	136	522	503	20	205	7700	17	ND	ND	24	115	ND	16	ND	ND	ND
OG 8	HS	0.31	104	355	34	34	26	430	652	338	40	84	1426	26	ND	ND	ND	8	ND	16	ND	ND	ND
0G 9	NS	0.31	86	116	27	30	14	304	522	320	38	58	2602	50	ND	ND	8	26	ND	2	ND	ND	ND
0G 10	MS	0.40	82	239	34	30	20	1510	522	230	46	70	2434	48	ND	ND	42	8	ND	6	ND	ND	ND
OG 11	HS	0.60	121	161	33	32	12	244	562	322	28	54	2008	ND	ND	ND	16	20	ND	6	ND	ND	ND
0G 12	HS	0.73	112	237	35	40	21	737	873	516	43	106	1632	19	ND	DA	ND	26	ND	14	ND	ND	ND
0G 13	HQ	0.29	303	132	30	26	27	323	502	306	43	ND	956	ND	ND	ND	62	10	ND	0	AB.	NU	ND
OG 14	NQ	0.45	475	182	32	34	28	1154	582	452	55	58	2950	71	ND	ND	ND	20	ND	24	40 ND	ND	ND
06 15	HO HO	0.49	442	112	33	33	13	938	502	232	40	102	1830	22	ND	ND	34	34	ND	12	ND	ND	ND
06 17	HO	0.41	1900	427	31	33	14	365	1164	312	30	52	852	13	ND	ND	86	ND	ND	8	HD	ND	ND
0G 18	HO	0.29	364	258	31	34	30	530	381	250	48	58	1484	36	ND	ND	78	26	ND	12	24	ND	ND
OG 19	NQ	0.41	357	436	32	35	29	892	552	428	40	58	1000	41	40	ND	58	22	2	32	12	ND	ND
OG 20	HQ	0.48	391	300	34	29	35	519	682	480	45	36	624	14	ND	ND	22	10	ND	20	ND	ND	ND
OG 21	HQ	0.41	794	289	33	35	19	1244	783	520	28	58	1565	13	ND	ND	76	20	ND	22	ND	ND	ND
0G 22	HQ	0.47	593	471	35	38	19	732	883	504	28	52	346	16	ND	ND	16	6	ND	16	ND	ND	ND
OG 23	HQ	0.17	487	163	27	25	45	485	271	228	65	38	288	ND	50	ND	110	30	8	20	10	NO.	ND
06 24	MS	0.45	011	367	35	35	14	1651	1104	1350	43	118	1185	59 ND	ND	NO	20	148	30	24	ND	ND ND	ND
06 25	HS NC	0.34	116	205	37	38	3	517	452	147	58	51	2431	ND	ND	ND	24	20	ND	4	ND	ND	ND
0G 27	MS	0.34	55	182	34	38	13	868	431	364	45	88	2714	ND	ND	ND	32	24	ND	12	ND	ND	ND
0G 28	MS	0.33	46	173	36	32	19	466	522	378	53	126	3858	ND	ND	ND	ND	12	ND	8	NO	ND	ND
OG 29	MS	0.35	273	161	32	34	13	168	401	262	53	80	4536	ND	ND	NO	ND	36	ND	38	ND	ND	ND
OG 30	FS	0.28	105	111	27	34	18	130	452	1740	68	100	7400	ND	NO	ND	ND	94	ND	30	ND	ND	ND
OG 31	MS	0.3	149	205	35	40	13	647	492	392	23	76	3510	ND	ND	ND	ND	ND	ND	2	ND	ND	ND
0G 32	FS	0.37	143	270	30	23	5	351	542	828	60	116	6100	ND	ND	ND	ND	38	ND	14	ND	ND	ND
OG 33	MS	0.29	294	209	26	23	7	1770	552	458	30	42	972	50	NO	ND	22	ND	ND	6	ND	DN IO	NU
06 34	HQ NC	0.12	4/3	15/	33	20	4/	310	401	194	60	123	45/	.67	ND	ND	94 ND	ND	ND	2	24 ND	ND	ND
00 35	MO	0.39	513	125	28	29	14	3887	773	100	20 58	177	501	70	ND	ND	72	10	ND	2	28	ND	ND
06 37	FS	0.32	117	287	30	20	5	321	1545	387	38	91	5464	15	ND	ND	18	26	ND	ND	ND	ND	ND
0G 38	HS	0.3	283	299	33	23	8	951	662	308	43	50	1224	54	ND	ND	ND	2	ND	4	ND	ND	ND
OG 39	MQ	0.07	141	168	23	9	70	232	221	128	60	116	438	12	ND	ND	122	12	ND	14	12	ND	ND
OG 40	FS	0.15	B7	112	31	23	16	199	291	636	73	116	6900	ND	ND	ND	ND	108	NO	14	NO	ND	ND
OG 41	MS	0.37	334	324	35	26	9	598	2849	748	35	42	922	54	ND	ND	55	ND	ND	ND	ND	ND	ND
OG 42	HQ	0.23	642	365	36	54	2	262	2017	490	13	28	550	23	ND	ND	58	10	ND	4	8	ND	ND
OG 43	HH	0.13	110	140	30	29	0.5	162	439	2057	15	84	6600	22	82	ND	ND	136	16	28	20	ND	ND
OG 44	NQ	0.25	916	219	28	32	5	290	1298	310	23	ND	244	20	ND	ND	50	10	ND	2	10	NU	ND ND
06 45	MM	0.38	304	115	27	23	0.5	162	180	1308	60	172	6693	ND	ND	ND	ND	130	28	42	NO	ND	ND
06 47	MS	0.42	235	277	33	37	3	247	1069	357	23	52	882	80	ND	ND	4	ND	ND	18	ND	ND	ND
0G 48	HQ	0.28	509	245	29	31	2	483	859	210	65	ND	236	48	ND	ND	44	18	ND	2	ND	ND	ND
OG 49	NH	0.2	111	144	29	31	0.5	167	290	988	52	108	7800	ND	ND	ND	32	115	22	32	ND	ND	ND
OG 50	MS	0.43	161	215	23	37	2	2116	629	242	37	ND	260	68	ND	ND	ND	ND	ND	NO	ND	ND	ND
OG 51	MQ	0.15	540	239	20	28	1	337	360	172	27	ND	236	20	ND	ND	74	10	ND	6	42	ND	ND
OG 52	MH	0,22	133	268	26	40	1	411	399	834	55	88	6800	ND	ND	ND	ND	60	10	8	ND	30	ND
OG 53	KS	0.49	242	177	17	25	1	591	799	284	30	34	1976	17	ND	ND	26	MD	ND	14	ND	ND	ND
0G 54	MQ	0.16	/51	257	23	27	0.5	893	539	130	52	ND	104	23	54	NO	44	16	ND	2	22	NO	ND
00 55	MC NC	0.1/	251	249	2/	34	0.5	234	200	332	40	36	666	40	34	NO	40	100	30	40	2	ND ND	ND
00 50	NO NO	0.5	443	104	20	3/	0.5	700	400	126	40	36	600	49	ND	MD	10	4	ND ND	M	MD	NO.	ND.
06 57	MM	0.23	84	194	27	20	0.5	362	300	1676	62	107	6200	ND	62	ND	ND	130	30	46	20	NO	ND
06 50	MS	0.67	219	534	36	32	6	306	1967	450	25	68	666	30	ND	ND	ND	MD	ND	ND	ND	30	ND
0G 60	HO	0.21	1600	154	32	30	2	290	1218	272	60	32	148	ND	ND	ND	64	30	6	14	32	NO	ND
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뮾	udd	Carl	2 9	2 2	Q	R	9 9	2 9	2 9	2 9	2 5	2 9	2 9	9	1 9	Q	Q	Q	R	2	R	9 1	2 5	2 9	2	Q	2	Q	Q	Q	9	R	2	8 9	2 5	2 2	2	Q	Q	2 9	2 2	2 9	2	9	Ð	9	Q	2	21	2 5	5 5	2 9	1 5	2
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T102	mdd	2.9	290	5800	7400	2292	378	4800	1136	99	542	394		10025	100Es	6100	6300	109	5900	5900	2972	634	6000	5700	2762	0100	0065	3500	5300	6300	5400	1499	2846	5900	2474	5500	3226	6200	1788	870	0069	1978	6640	2498	00029	2110	6100	3800	5000	3250	5000	6700	4900	4000
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5	udd		30	8	1 5	12	15	52	22	40	35	22	8	8 3	6	1 2	55	11	52	20	20	42	20	22	8	2 2	R 9	32	1 21	1 2	12	20	11	L	32	10	02	42	27	37	25	52	1 1	4	15	12	17	12	22	11	25	12	15	1
Ba	udd		148	0130	240	238	674	875	112	286	230	584	R	6/4	nn2t	414	5662	161	205	2338	509	118	388	0000	646	040	202	516	105	276	010	473	506	204	666	676	374	320	434	346	695	390	200	818	474	452	680	222	668	464	526	458	220	192
Ŧ			228	626	1 019	168	927	200 2	176 1	6/6	601	611	051	318	AQC DLL	250	300	9/9	340	270 3	494	401	170	361	265	177	301	236	162	115	672 2	584	-566	162	083	802	105	1153	912	312	512	243	202	583	1/2	203	251 1	199	381	566	594	151	192	108
W	dq		94 1	09 00	26 36	83 1	1 06	88	47 3	16	89 1	32	88 :	1 50	15	12	- 6E	04 2	18	26	44 1	16	F	8	56	2 8	88 20	2 68	E	59	39	24 1	08 1	49	24 1	59 5	83	26	62	82	25	1 EZ	1 10	1 91	EI	1 66	E	IE	11	72 1	1 02	MI	8	44
9	A H		3	4			5 1	5 2	5 2	5	4						1 4	8	5	5 1	9	5 2	5 1	2	5		 -	. 9			5	7 4	4	2	2	5		1	7 3	2 2	9					1	1 1	1 2	1 2	2 16	4 48	2	2	2
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ž	udd		30	23	10	CE.	24	21	21	21	28	52	11	10	1		3.5	5	31	31	25	1	5	26	2		22	26	1 1	52	26	20	23	33	M i	9.9	30	29	19	22	52	24	2 2		24	9C	23	22	22	21	23	12	5	5
12	udd		154	247	848	826	195	188	284	358	196	208	19	202	011	48	19	2000	53	54	349	271	52	18	344	2 2	511	344	E	122	18	420	113	32	108	57	128	85	154	181	32	344	ott	FUL	75	182	68	212	111	157	199	422	284	380
Pb	mdd		184	211	101	109	2200	11	210	687	307	415	450	102	16	3 5	16	412	28	28	440	272	24	78	202	8 5	111	423	22	45	115	247	328	20	200	111	501	5	162	326	69	164	100	140	16	80	33	40	п	32	149	32	64	39
ŋ	*	5	0.60	0.16	0.50	02.0	0.33	0.38	1.30	0.24	0.79	1.58	60'0	65.0	11.0	010	0.11	1.03	0.03	0.07	0.54	0.19	0.07	0.21	0.35	0.0	0.25 AF	69.0	0.07	EE.0	0.32	0.55	0.60	0.07	0.45	0.26	0.66	0.24	0.45	0.25	0.28	0.36	12.0	17.0	0.33	0.55	0.16	0.41	1.0	0.67	15.0	16.0	0.51	0.59
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ample	ock ty		9 90	8	3 3		9	9 90	9 90	00 6	50	30	8	8	30		8	200	90	8	8	90	3 90	8	3	3 30	2 20	3 3	3 20	3 30	8	5 90	5 90	5 30	30	20 20	01 10	01 90	06 10	00 10	06 10	01 00	00 10	10 10	01 00	00 11	11 00	06 11	06 11	06 11	11 50	11 90	11 90	00 11

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ample no. ock type	and	Cu ¥	Pb ppm	Zn ppm	N1 ppm	Co ppin	Ag	Au ppb	Mn ppm	Ba ppm	Cr ppm	V	T102 ppm	As ppm	WO3 ppm	Mo ppm	Sn ppm	Zr	Rb ppm	Sr ppm	B1 ppm	ppm	Pf
DG 61	MS	0.50	784	154	30	38	3	694	1228	148	30	54	290	61	ND	ND	46	6	ND	0	28	ND	
0G 62	HE	0.16	211	247	29	25	2	460	439	3130	50	68	6800	ND	ND	ND	56	160	28	66	38	ND	
DG 63	AS	0.50	101	848	37	37	0.5	1300	3485	856	17	ND	304	36	ND	ND	ND	ND	ND	MU A	ND	ND	
OG 64	FS	0.23	87	432	30	32	0.5	226	619	1240	52	144	7400	ND	ND	ND	ND	ND	10	16	ND	ND	1
OG 65	MS	0.30	60	826	32	30	0.5	283	1168	238	27	120	2292	ND	ND	ND	12	24	ND	28	30	ND	1
06 66	NQ	0.35	2200	195	24	30	0.5	290	1927	2876	5/	125	178	ND	84	ND	ND	86	8	50	8	ND	1
00 67	AS	1.30	210	284	21	30	0.5	200	3176	1112	22	34	1136	28	ND	ND	ND	ND	ND	ND	ND	ND	1
06 69	HO	0.24	687	358	27	33	0.5	391	879	286	40	30	66	ND	ND	ND	42	8	ND	ND	6	ND	
0G 70	MS	0.79	307	195	28	35	4	589	1109	230	35	ND	542	56	ND	ND	ND	ND	ND	ND	ND	ND	
OG 72	AS	1.58	415	208	25	35	2	932	1119	584	22	124	394	35	ND	ND	4	ND	ND	4	CM	ND	
OG 73	HQ	0.09	450	51	17	13	1	188	140	78	60	28	ND	ND	ND	ND	76	24	ND	ND	ND	ND	
OG 74	AS	0.55	260	282	18	21	1	314	1318	674	30	44	3200	58	ND	ND	ND	14	ND	ND	ND	ND	
0G 75	111	0.11	31	145	21	23	0.5	237	389	4200	55	108	4800	49	58	ND	16	134	26	34	26	ND	
OG 76	HH	0.23	23	110	22	27	0.5	239	379	2324	42	122	5300	11	74	ND	14	152	16	34	20	ND	
OG 77	CS	0.10	21	48	37	28	0.5	121	250	414	62	132	6100	285	ND	ND	28	154	ND	20	30	NU	
OG 78	CS-MM	0.11	31	61	33	25	0.5	139	300	2662	55	130	5300	225	ND	ND	20	102	14	30	ND.	ND	
OG 79	AS	1.03	412	2000	27	35	8	704	2676	797	17	24	709	97	ND	ND	18	141	ND	22	12	NO	
0G 80	LS CE	0.08	28	53	3/	18	0.5	118	340	2379	52	150	5900	145	ND	ND	64	158	10	24	50	ND	
06 81	LS NS	0.07	20	340	20	25	0.5	205	1494	500	20	120	2702	48	ND	ND	ND	40	ND	ND	ND	ND	
06 83	MO	0.19	272	271	15	10	0.5	216	401	118	42	ND	634	16	ND	ND	32	22	2	ND	46	ND	
0G 84	CS	0.07	24	52	33	19	0.5	131	170	388	50	114	6000	262	ND	ND	40	170	6	14	30	ND	
DG 85	MM	0.21	78	81	26	24	0.5	190	361	3000	27	84	5700	133	ND	ND	ND	80	4	36	ND	ND	
0G 85	MS	0.35	202	344	23	18	5	226	942	646	30	48	2762	77	ND	ND	ND	4	ND	12	ND	ND	
OG 87	CS	0.05	38	55	35	20	0.5	175	221	1648	15	104	6700	60	ND	ND	36	177	16	34	22	ND	
DG 88	195	0.25	50	79	28	33	0.5	188	301	1262	30	134	5400	384	ND	ND	34	102	24	42	30	ND	
OG 89	MM	0.34	111	116	28	33	1	195	1454	2452	39	88	5200	23	ND	ND	ND	96	D	42	ND	ND ND	
DG 90	HS	0.69	423	344	26	20	6	689	2236	516	32	90	3500	ND	64	ND	22	132	ND	0	78	KO KO	
OG 91	CS	0.07	22	33	30	14	0.5	131	291	105	15	82	5300	76	/5	ND	22	152	16	42	ND	ND	
06 92	HM	0.33	45	122	25	18	0.5	165	511	1276	12	96	6300	44	30	NO	78	104	14	32	ND	ND	
06 93	110	0.32	115	81	20	21	0.5	139	0/2	20/0	12	19	3400	14	86	NO	ND	ND	ND	2	ND	ND	
00 94	MC	0.55	328	113	20	27		308	1035-	506	17	172	2846	10	40	ND	8	4	ND	ND	ND	ND	
06 95	CS.	0.07	20	32	33	20	0.5	149	291	204	7	82	5900	212	ND	ND	14	160	10	20	32	38	
OG 97	MS	0.45	200	108	30	27	7	524	1083	656	32	114	2474	ND	ND	ND	ND	28	ND	26	ND	ND	
OG 98	181	0.26	111	57	37	34	0.5	165	802	1676	10	104	5600	12	ND	ND	ND	105	6	34	18	ND	
QG 99	124	0.33	109	118	32	25	0.5	177	652	586	30	130	5000	ND	138	ND	ND	NO	ND	ND	ND	KD	
OG 100	MS	0.66	125	128	26	28	2	283	591	374	20	124	3226	ND	58	ND	10	244	ND	8	MD	ND	
OG 101	MM	0.24	64	85	29	37	0.5	126	531	1320	42	132	6200	ND	78	ND	ND	106	18	28	ND	ND	
OG 102	MS	0.45	162	154	19	16	7	362	912	434	27	85	1788	ND	ND	ND	ND	16	ND	ND	ND	ND	
OG 103	MQ	0.25	326	181	22	20	2	278	912	346	37	60	870	ND	ND	HD	26	32	KD	12	MD NO	NU	
OG 104	MM	0.28	69	92	25	26	0.5	152	612	695	25	157	6900	ND	88	ND	ND	104	ND	K4	ND	ND	
0G 105	MS	0.30	164	344	24	2/	0.5	3/3	1243	390	25	42	19/8	ND ND	AD	ND	ND	12	ND	ND	HO	ND	
06 105	HQ.	0.24	61	100	20	34	0.5	130	1303	1614	12	197	400	NO	62	ND	16	130	18	42	28	ND	
06 107	MC.	0.21	140	102	32	30	2.5	215	1293	518	7	104	2408	ND	ND	ND	ND	42	ND	34	ND	ND	
06 100	HH I	0.33	41	75	24	31	0.5	113	371	974	15	140	6300	ND	ND	ND	ND	98	12	10	MD	HD	
06 110	MS	0.55	80	182	30	36	3	193	1203	452	12	54	2110	ND	ND	ND	ND	16	ND	8	ND	ND	
DG 111	MPI	0.16	33	68	23	25	1	131	251	1680	17	132	6100	ND	ND	ND	6	146	32	26	ND	ND	
DG 112	MS	0.41	40	212	22	21	1	231	561	222	12	66	3800	ND	35	ND	ND	26	ND	18	ND	ND	
OG 113	-	0.1	11	111	22	19	1	211	381	668	22	72	5000	ND	94	ND	54	156	18	16	8	ND	
OG 114	MS	0.67	32	157	21	18	2	1672	1995	464	17	64	3250	ND	72	ND	16	16	ND	18	ND	ND	
OG 115	MS	0.51	149	199	23	26	4	4870	1594	526	25	58	5000	36	ND	ND	8	. 30	4	22	ND	ND	
OG 116	HH	0.37	32	422	27	23	2	334	451	458	12	118	6700	ND	ND	ND	ND	80	64	22	NO	ND	
OG 117	MS	0.51	49	284	30	26	2	563	251	220	15	188	4900	ND	NO	ND	ND	42	ND	14	ND	ND	
	MC	0.59	39	380	34	37	2	144	301	192	7	164	4000	14	ND	NO	NO	52	ND	ND	ND	ND	

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Table 4 - Ongombo Prospect: trace element geochemistry of the gossan exposure. (Note: ND means "below detection limit").

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Sample no.	Cu ppm	Pb ppm	Zn ppm	Ni ppm	Co ppm	Ag ppm	Au ppb	Ma ppm	Ba ppm	Cr ppm	V ppm	T 102	As ppm	WO3 ppm	Mo ppm	Sn ppm	Zr ppm	Rb ppm	Sr ppm	B1 ppm	Nb ppm	Y ppm
0G 144	0.21%	73	26	40	19	3	88	293	160	NO	126	983	ND	30	92	ND	NO	ND	ND	ND	ND	ND
0G 145	566	71	50	21	10	3	149	115	636	22	58	387	ND	ND	52	ND	ND	ND	ND	ND	ND	ND
OG 145	0.46%	57	46	27	13	2	182	272	247	12	80	213	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND
OG 147	307	52	43	21	13	ND	137	293	1980	10	62	293	ND	ND	ND	ND	20	ND	15	ND	ND	ND
OG 148	283	48	61	23	15	ND	B1	335	215	7	57	236	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND
OG 149	412	36	104	30	22	ND	88	356	101	2	86	256	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND
OG 150	182	36	34	36	18	ND	81	283	115	24	85	378	ND	ND	ND	KD	ND	ND	ND	ND	ND	ND
OG 151	236	22	58	31	11	2	119	115	3805	24	45	215	ND	ND	ND	ND	27	ND	39	ND	ND	ND
OG 152	90	38	27	24	10	ND	76	105	5221	2	71	179	ND	ND	ND	ND	36	ND	40	ND	ND	ND
OG 153	117	25	29	14	10	ND	88	105	662	41	58	146	ND	58	ND	ND	ND	ND	17	ND	ND	ND
OG 154	0.16%	46	39	23	28	2	116	220	4636	29	102	465	ND	ND	ND	ND	35	ND	53	ND	ND	ND
OG 155	0.17%	30	75	22	19	3	432	105	3822	24	69	295	ND	ND	34	20	31	ND	40	ND	ND	ND
OG 156	127	36	20	12	10	ND	73	388	495	19	50	195	ND	ND	ND	ND	16	ND	ND	ND	ND	ND
OG 157	234	47	22	12	8	ND	66	168	88	31	55	325	ND	ND	ND	ND	NO	ND	ND	ND	ND	ND
OG 158	135	38	21	13	10	ND	56	335	513	22	36	122	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND
OG 159	114	23	20	10	6	ND	88	388	112	31	40	75	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND
OG 160	399	42	23	16	12	2	349	189	4138	24	55	225	ND	ND	ND	ND	28	ND	42	ND	ND	ND
OG 161	405	45	28	18	16	NO	405	314	2080	22	44	182	ND	ND	ND	ND	NO	ND	28	ND	ND	ND
OG 162	333	50	31	15	13	ND	147	168	218	12	62	255	NO	ND	ND	ND	ND	ND	ND	NO	ND	ND
DG 163	518	36	23	21	16	ND	165	115	140	2	48	441	ND	ND	ND	ND	ND	ND	10	ND	ND	ND
OG 164	579	36	37	16	21	ND	159	136	124	2	35	358	NU	ND	ND	ND	NU	ND	ND	NU	NO	ND
OG 165	203	36	18	20	18	ND	144	241	91	NO	94	201	NU	80	ND	ND	ND	ND	NU	ND	NU	ND
OG 166	367	33	2/	10	18	ND	167	189	1912	15	80	183	ND	ND	ND	ND	10	ND	10	ND	ND	ND
06 167	231	55	22	21	1/	2	233	189	591	22	40	4/9	NU	NO	ND	ND	ND	ND	NO	ND	ND	ND
06 168	630	8/	33	21	14	ND	1/5	105	1121	17	01	841	NO	ND	ND	NU	18	ND	ND	ND	ND NO	NU)
06 109	120	20	25	14	11	ND	14/	105	2030	17	110	293	NU	ND	NU	ND	ND	KD	ND	ND	ND	NU
06 170	374	30	50	24	19	NO	301	109	224	19	119	528	NU	ND	39	ND	ND	ND	NU	ND	ND	ND
06 171	271	33	42	24	14	NO	194	300	75	12	00	305	NU	ND	NO	NU	ND	NU	ND	ND	ND NO	ND
06 172	204	24	4/	29	10	NO	129	130	13	14	25	293	ND	ND	ND	ND	ND	ND	NO	NU	ND	ND
06 175	235	22	10	17	10	ND NO	13/	130	647	24	71	180	NU	ND	NU	ND	NU	ND	ND ND	ND	ND	NU
00 174	0.24.07	122	20	10	42	1	215	1/0	392	20	109	332	21	ND	NU	ND III	ND	ND	ND	NU	ND	ND
00 175	0.109/	45	20	21	28		105	230	1142	10	113	1000	ND	ND	67	NU	19	ND	HD HD	ND	NO	ND
06 170	200	25	15	15	15		152	04	4505	20	45	124	ND	ND	47	ND	26	NO	RA	ND	ND	ND
06 178	326	37	15	11	22		180	220	530	22	37	192	ND	NO	AND	ND	33	ND	04	ND	ND	ND NO
06 179	699	44	18	18	21	î	223	220	1020	12	63	277	ND	ND	ND	ND	M	ND	ND	ND	ND	NO
OG 180	652	50	19	19	16	1	245	126	2160	2	63	395	ND	ND	ND	ND	23	ND	15	ND	ND	ND
05 181	627	45	20	21	17	î	189	241	2540	7	55	244	ND	ND	ND	ND	18	ND	ND	ND	NO	ND
0G 182	427	41	16	21	17	1	169	230	1206	19	73	721	ND	ND	ND	NO	16	ND	ND	ND	ND	ND
OG 183	545	44	19	19	19	1	215	210	1289	22	101	558	ND	ND	ND	ND	26	ND	ND	ND	ND	ND
OG 184	472	57	17	19	17	ND	172	105	1377	15	110	495	ND	ND	ND	ND	ND	ND	19	ND	ND	ND
OG 185	0.21%	57	23	20	25	ND	220	168	446	12	157	939	ND	ND	ND	ND	NO	ND	ND	ND	ND	ND
OG 186	725	43	25	23	24	ND	152	450	305	15	353	588	ND	ND	ND	KD	ND	ND	ND	ND	ND	ND
OG 187	0.12%	44	18	24	22	ND	182	189	358	5	314	647	ND	ND	ND	17	ND	ND	31	ND	ND	ND
OG 188	517	35	20	21	38	ND	172	189	299	10	146	597	ND	44	ND	ND						
OG 189	327	29	16	20	15	ND	167	147	302	15	119	609	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND
OG 190	233	38	16	19	12	ND	179	178	696	15	133	400	NO	ND	ND	ND	ND	ND	ND	ND	ND	ND
OG 191	227	21	28	14	33	ND	121	136	194	20	55	287	NO	44	ND	ND						
OG 192	310	41	18	20	21	ND	215	168	544	12	145	469	ND	ND	ND	ND	ND	ND	HD	ND	ND	ND
OG 193	384	30	22	28	43	ND	159	178	283	5	113	364	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND
OG 194	304	33	19	26	33	ND	174	272	236	2	119	676	ND	ND	35	ND	ND	ND	ND	ND	NO	ND
OG 195	454	33	29	22	43	ND	182	210	181	2	166	1195	ND	ND	ND	ND	NO	ND	ND	ND	ND	MD
OG 196	209	29	24	26	31	ND	134	230	167	2	82	889	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND
OG 197	264	23	29	14	34	ND	152	189	199	15	135	365	ND	ND	ND	ND	ND	ND	NO	ND	ND	ND
OG 198	395	30	30	26	38	NO	136	356	236	10	147	254	ND	ND	ND	ND	ND	ND	25	ND	ND	ND
OG 199	184	29	19	21	23	ND	136	178	311	20	72	722	ND	ND	ND	ND	ND	ND	17	ND	ND	ND
OG 200	258	34	24	26	27	ND	182	168	119	15	81	476	ND	ND	ND	ND	ND	ND	NO	ND	ND	ND
OG 270	220	40	21	24	31	1	146	125	533	42	105	1002	ND	48	ND	ND						

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sample no.		04	107				n qu	mu	mod	Dom	udd	*	mdd	udd	udd	udd	mdd	udd	udd	mdd	bba	udd
		L	L																			
116 20	120	U2	06	36	9	UN	116	178	180	25	107	642	UN	Q	CM	ND	g	ND	CAL IN	ND.	g	CIN
NC 375	735	3	a k	3	100	2	134	614	225	22	354	511	9	9	Q	9	9	9	15	9	9	9
06 273	6/4	36	15	9	113	2	139	210	166	32	182	1037	9	Q	Q	QN	2	Q	Q	QH	Q	NO
06 274	401	25	52	M	44	2	141	168	513	30	178	2816	2	9	R	QN	24	9	Q	Q	Q	Q
06 275	611	M	8	4	110	2	141	142	221	32	101	262	Q	Q	Q	DN	Q	QN	Q	QN	Q	QN
06 276	222	53	21	21	20	Q	16	151	100	15	24	315	R	QN	QN	QN	9	R	Q	Q	Đ	QN
06 277	884	43	101	50	175	Q	167	212	209	30	124	616	9	Q	42	QN	9	¥	Q	QN	9	QN
06 278	0.26%	89	148	16	110	Q	157	492	610	2	140	334	9	Q	36	Q	9	Q	Q	QN	모	QN
06 279	178	69	35	49	52	9	182	199	746	90	100	382	2	2	9	2	2	2	9	9	9	Q
06 280	850	46	M	52	6 i	2	134	220	470	2	3	543	2 9	2 9	2 9	2 9	2 1	2 9	2 9	2 9	2 9	2 9
06 281	848	8	102	23	44	2	141	168	8	2	8	250	2 1	2 9		2 :	2 9	2 9	2 9	2 1	2 1	QN S
06 282	0.32%		14	3	64	2 9	182	199	140	9	103	000	2 5	2 9	24	8		2 9	2 5		2 5	2 9
04 283	000 U	10	104	a c	10	2 9	201	001	126	35	8	364	2 9	2 9	2 9		2 9	2 9	2 5		2 9	2 9
DC 285	0/ 02:0	*	5	-	5	9	DEI	220	145	5	3	169	9		9	Q	2	9	2	9	2	Q
06 286	564	1 29	112	2	52	2	151	356	787	35	127	6//	9	R	2	QN	R	2	2	9	2	Q
06 287	404	31	215	45	198	9	167	1006	232	42	52	253	9	OH	QN	ON	Q	9	æ	ON	Q	QN
06 288	376	×	4	24	25	Q	146	147	137	32	71	258	Q	QN	9	19	R	QN	Q	QN	Q	ON
06 289	285	27	16	æ	112	9	124	283	3282	35	200	315	Q	R	QN	QN	25	R	57	QN	R	QN
06 290	0.32 %a	4	315	109	502	Q	162	765	201	27	98	326	Q	QN	QN	QN	R	Q	g	QN	Q	Q
162 50	680	36	44	33	52	9	159	335	297	27	286	486	Q	Q	Q	QN	Q	R	2	QN	Q	Q
06 292	169	29	44	33	40	Ð	146	189	236	24	194	525	Q	및	9	28	Q	Q	Q	ND	R	Q
06 293	0.19 %	31	28	30	32	9	152	105	294	53	145	176	Q	R	2	Q	11	R	21	QN	Q	QN
06 294	0.2	23	90	25	42	R	139	105	343	22	228	289	2	Q	35	Q	Q	R	Q	Q	g	Q
06 295	141	32	2	32	47	9	121	64	326	24	201	374	2	R	9	27	2	R	22	9	R	9
06 296	521	30	47	51	28	9	139	147	264	26	170	586	2	9	2	QN	2	R	9	2	2	2
00 297	0.25%	6	523	4	011	2 9	101	346	052	22	191	411	2 9	2 9	2 9	9	2 9	2 5	2 9	2 9	2 9	
06 298	100/	7	2 2	5 ;	8	2 9	107	501	1/0	22	151	250	2 9	2 9	2 9	2 9	2 9	2 9	2 9	2 9	2 9	2 9
06 200	5/5	4 0	9 7	17	51 5	2 9	Por les	071	200	1.	107	100	2 9	2 9	2 5	2 9	2 6		2 5	2 9		
102 301	0/ 07-0	5	2 %	24	00	2 9	TUR	190	243	20	100	CLL	2 9	1 9				1 9		2 5	2 9	
06 302	307	24	20	10	12	9	212	136	368	36	150	356	2	2	4	2	2	9	R	9	2	2
E0E 303	406	19	25	15	14	QN	162	84	341	34	11	402	Q	QN	Q	ND	ON	QN	Ð	QN	9	9
06 304	348	36	21	14	5	Q	232	168	3102	34	29	310	QN	QN	Q	18	22	QN	38	Q	Q	Q
06 305	117	16	56	16	5	Ø	167	73	8741	38	165	175	Q	100	Q	18	69	Q	192	Q	2	ą
00 306	192	19	61	18	11	Q	154	115	1402	41	162	182	2	R	Ð	15	N	Q	2	Q	Q.	¥
00 307	110	2	24	61	9	2	146	a	4127	F 1	872	516	2 9	2 9	2	9 9	Z I	2 9	8 9	2 9	2 1	
906 308	330	82	22	4 1	5	2 9	130	8/1	65/1	7 8	517	1245	2 9	2 9		2 9	2 9	2 9	2 9	2 9	2 5	
ALL DO	250	5 2	5 62	10	10	2 2	221	283	ILEI	T PE	301	155	9	19	9	9	2 2	2 9	2 2	9	2	2 9
00 311	122	E	5	12	13	9	225	199	513	8	125	139	Q	QN	9	R	N	2	2	GN	2	Q
06 312	109	12	16	15	14	9	136	199	257	36	264	243	Q	R	Q	Q	Q	QN	9	N	9	QN
06 313	135	12	14	18	10	Q	106	210	697	26	88	345	Q	QN	Q	QN	CN	ON	9	N	R	QN
06 314	212	18	22	21	26	ę	119	367	308	50	160	25	QN	QN	Q	QN	Q	QN	Ð	QN	QN	QN
06 315	119	R.	26	21	10	Ø	131	147	17200	R	119	267	Q	156	9	16	114	Q	248	53	Q	Q
06 316	185	102	29	90	14	Q	141	189	£633	43	185	115	9	2	Q	24	41	R	100	22	2	R
06 317	278	155	20	24	=	2	197	210	1588	4	148	540	2	9	Q	2	2	2	9	R	2	QN
06 318	1/1	8	12	21	-	9	141	157	16000	ł	32	170	2	136	34	40	126	Q	364	15	2	Q
00 319	376	16	96	25	18	9	232	105	13100	¥ :	175	194	24	8	Ð	Q	106	2	255	48	2	Q
06 320	69	158	22	19	5	2	220	8	19800	20	110	345	32	148	2	9	138	9	352	74	2	2
06 321	159	101	54	20	12	CN	179	115	15400	10	120	156	2	8	R	8	115	¥	199	15	2	Q

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Table 5 - Ongombo Prospect: trace element geochemistry of the ore-bearing sequence in drill cores. (Boreholes no. 068 151A, 068 153A, 068 154A).

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BOREHOLE DEB 151A

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sample no.	3	2	5	E and	3	2	2 12	in the second		-	- mou	*		mun	a mon	mon	i u	mon		in the second		DOM:
	medd	Indd	mdd	intd	mid	nerd	and	1	ł		-		L		Ł	Ł		Ł		Ł		1
	F	9	0			5	g	222	200	yo	113	1 03	Ş	PA.	Ş	S	UEC.	UEI	106	54	G	9
585 30	59	11	44	7 M	14	2 9	9	211	340	96	6	0.92	2	84	2	2	261	32	144	2	2	15
06 386	54	11	5	40	16	2	50	282	552	89	153	1.13	ON	11	2	QN	284	123	143	CIN	IE	53
06 387	35	19	60	36	11	9	Q	302	462	105	107	0.99	Q	72	QN	QN	242	104	153	21	33	46
06 388	9	22	99	\$	25	2	42	362	416	96	118	0.96	Q	92	Q	Q	224	68	184	R	9	65
0G 389	5	19	15	84	32	Q	50	483	53	185	257	1.09	9	84	25	QN	82	119	63	Q	9	84
06 390	21	19	23	HE.	22	2	QN	544	534	16	105	1.00	9	88	9	QN	287	96	170	20	26	47
162 301	8	20	110	48	23	GN	20	544	111	96	144	1.06	Q	61	2	Q	225	143	116	2	20	19
06 392	10	2	61	36	11	Q	Q	574	165	86	32	1.05	9	15	9	QN	389	139	181	UN	35	62
06 393		16	19	IE	11	2	QN	483	574	64	2	0.90	2	69	Q	R	270	81	171	R	ą	42
06 394	*	14	99	32	15	8	R	473	295	34	98	0.93	9	33	2	Q	258	92	188	QN	53	3
06 395	m	15	18	148	21	Q	QN	252	55	167	134	0.71	R	92	2	Q	66	18	23	QN	9	24
06 396	146	12	06	45	21	2	2	362	349	108	102	96-0	Q	68	9	Q	214	73	129	Q	QN	8
165 30	6	25	118	134	35	Q	47	644	74	130	170	0.89	9	모	9	Q	144	15	36	DN	23	40
06 398	152	12	82	45	19	Q	40	574	219	84	138	0.95	9	86	Q	QN	220	29	35	ND	R	39
06 399	47	19	18	46	16	Q	Q	674	219	63	164	0.96	Q	57	Q	QN	207	44	82	20	21	12
06 400	28	15	88	42	14	Q	DN	473	384	88	140	1.04	1	42	2	Q	262	49	100	QN	30	37
00 601	399	E	8	20	21	1	QN	185	504	63	134	0.93	Q	80	QN	21	212	99	131	QN	Q	31
06 602	0.21%	15	101	145	11	1	47	195	119	101	85	0.83	Q	31	Q	QN	170	18	107	QN	27	34
06 603	110	16	106	ES	18	9	42	1/5	371	16	133	0.99	Q	62	Q	ND	246	49	135	QH	9	43
06 604	194	14	134	108	17	Q	42	126	107	166	134	0.68	Q	62	Q	QN	113	11	36	R	2	22
06 605	88	11	83	35	11	9	Q	559	221	87	115	1.03	Q	72	Q	QN	311	34	142	R	QN	41
06 606	14	19	110	6	27	9	Q	838	154	140	198	1.02	Q	GN	Q	IE	111	39	88	R	26	36
06 607	37	20	20	47	22	2	Q¥	569	204	19	133	0.96	R	15	Q	CN	243	43	150	R	36	53
06 608	33	19	55	11	53	Q	QN	778	132	EOI	289	1.25	9	20	25	Q	134	25	104	N	CN	æ
00 609	19	19	26	20	11	9	40	469	215	94	129	0.98	Q	09	Q	DN	223	43	139	ON	24	95
019 00	96	20	99	40	H	9	ND	379	683	84	139	0.96	9	51	9	QN	198	16	117	₽ ₽	Ð	48
Drs 2008																						
207 5 10																						
06 611	47	23	46	42	20	Q	QN	279	660	75	156	1.11	Q	16	Q	QN	218	129	131	Ð	9	54
06 612	40	22	8	47	11	2	9	319	635	84	114	1.07	QN	100	g	QN	231	127	107	Q	25	26
06 613	9	19	19	44	16		QN	269	704	87	127	1.13	Q	36	Q	QN	227	130	126	Q	33	8
06 614	13	20	19	34	12	2	Q	409	195	75	49	1.06	Q	47	20	QN	249	102	189	Q	30	68
06 615	6	20	75	38	14	9	R	- 582	522	58	100	1.02	Q	101	Q	QN	237	16	162	22	9	46
06 616	98	30	59	139	24	8	Q	359	603	65	127	1.08	9	29	R	QN	232	130	129	QN	32	54
06 617	29	25	6	55	23	2	\$	339	191	87	141	1.26	2	22	Q	QN	253	161	157	24	56	65
06 618	9	12	105	54	25	2	15	309	540	59	145	1.49	9	3	2	QN	381	155	310	R	48	16
05 619	31	23	81	20	18	2	47	309	136	67	183	1,19	Q	88	2	9	239	171	133	8	46	13
06 620	19	51	46	42	12	Q	Q	319	284	51	16	0.94	2	37	2	N	232	16	129	2	œ	51
06 621	118	22	2	53	15	Q	99	354	831	87	179	1.22	R	65	51	Q	240	173	140	12	25	99
06 622	20	19	68	43	15	Q	QN	364	418	11	123	1.06	2	QN	Q	Q	285	108	162	9	2	89
06 623	42	23	108	99	20	9	58	304	1268	19	186	1.23	R	55	Q	9	257	161	163	32	31	29
06 624	160	18	s	99	20	T	63	283	878	65	173	0.94	9	62	25	Q	209	161	196	Q	2	3
06 625	54	25	56	42	27	9	Q	324	166	60	142	1.18	9	66	Q	QN	246	159	140	Q	2	67
06 626	56	22	62	38	22	2	Q	506	880	20	122	1.12	Ð	8	21	QN	320	114	158	Q	23	8
06 627	12	20	13	33	22	2	48	265	831	70	118	1.10	9	14	23	Q	300	131	179	52	2	55
06 628	48	24	09	33	19	9	Q	445	548	53	134	16.0	₽	15	22	QN	298	103	169	9	¥	5
06 629	96	12	78	42	21	Q	Q	395	138	48	155	1.10	9	53	Q	QN	259	131	156	2	26	55

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BOREHOLE OGB 153A

Sample no.	ē	44	70	1N	Co	A	W	ų	Ba	5	~	1102	As	HO3	Q.	S	Ir	8	Sr	81	ND	*
	bod	đ	udd	udd	Hod I	Edd	qdd	udd	udd	udd	udd	+	udd	uxid	udd	udd	udd	udd	udd	mqq	ndd	udd
CCE JU	36		g	80	BC	UN	90	202	769	69	081	00 1	s	8	ŝ	Ş	346	130	125	G	16	67
770 00	20	36	3 2	2 2	8	2 5	2 2	222	103	101	2	101	2 5	3	2 5	2 9	84.6	5	161			2
06 324	F 8	2 2	83	38	20	2	9	282	551	11	121	0.99	2 9	2	2 9	2 9	239	108	139	N N	22	5 13
06 325	6	28	09	49	21	2	NO	242	783	8	167	1.07	9	15	9	9	225	142	104	Q	Q	3
06 326	98	21	83	33	12	Q	QN	212	311	94	122	0.89	9	86	Q	R	257	16	129	QN	Q	44
06 327	24	30	73	56	21	Q	54	232	616	69	217	0.91	Q	109	Q	R	177	149	72	22	QN	63
06 328	46	24	95	45	22	Q	QN	272	455	42	127	1.03	Q	75	9	Q	277	120	150	QN	Q	29
06 329	TE	24	20	44	52	QN	N	212	826	69	182	0.94	Q	103	2	UN	218	142	103	QN	24	35
06 330	33	21	8	0	11	2	2	242	390	16	102	0.96	9	55	Q	Q	265	105	132	26	QN	15
06 331	75	24	118	47	19	9	Q	282	645	52	158	1.02	Ð	36	R	QN	228	133	123	QN	QN	53
OG 332	21	23	68	37	11	9	g	232	456	8	107	0.97	2	74	2	QN	252	88	143	22	Q	56
06 333	15	25	66	42	81	2	2 9	242	555	22	8	0.98	2	46	2	2 1	227	118	128	9	37	62
06 334	R	24	2	98	= 1	2 9	9	232	100	5	2	1.05	2	47	2	Q 1	288	68	149	9	2	8
06 335	27	54	11	4 5	51	2 9		252	113	5 8	95	1.09	2 9	5	2 9	2 9	221	139	106	2 9	2 5	5
00 230	14	5	120	20	3 8	2 9	5 5	202	LUL	8 8	EVI	1 00		2 5	2 5	2 5	090	120	116	2 5		2 1
166 JU		3 2	05	35	15	2 9	99	2635	101	8	501	96.0	2 9	2 2	2 5		250	105	122	2 9	6 9	h (A
DEE SU	24	52	1 2	19	1		e Qu	EDE	608	14	145	0.98	9	1 58	9	Q	223	121	109	9	9	25
06 340	18	28	8	68	18	9	9	373	631	59	8	1.05	2	59	2	QN	248	113	125	Q	32	40
06 341	65	24	70	64	21	QN	QN	373	593	11	98	1.32	Q	65	Q	ON	435	109	229	ON	42	73
06 342	6	22	52	191	25	ND	DN	464	455	319	161	0.75	9	15	P	QN	26	113	14	QN	QN	47
06 343	20	20	74	49	18	Q	QN	393	682	74	66	1.11	Ø	8	R	QN	251	8	129	QN	50	46
06 344	4	22	45	42	11	QV .	QN	353	528	240	131	0.83	9	16	9	R	115	106	82	Q	32	36
06 345	1	21	323	27	6	2	QN	322	153	104	303	1.34	Q	ON	R	QN	119	24	135	QN	24	74
06 346	m	25	88	38	14	Q	QN	393	251	64	325	1.56	Q	74	ę	Q	119	88	83	g	Q	19
06 347	19	52	02		1	2	8	203	629	18	96	1.20	2	8	9	9	243	140	186	54	2	5
00 348		02	20	9	5	2 9	8 9	433	840	11	121	0.98	2	5	2	9	233	1	18	2 1	2 3	4
06 369	0 ¥	3 8	10	5	17		04	12G	104	20		1.01	2 9	24	2 9		877	8 8	5	2 5	97	18
06 351	14	15	6	S R	1	9	2	504	186	84	352	191	2 2	2		2 9	131	8 62	115	2 2	9 9	25
256 30	8	18	75	05	23	2	43	484	449	11	141	1.07	2	69	9	9	221	102	139	9	36	66
06 353	14	18	88	37	14	QN	40	383	226	114	81	0.81	9	8	2	Q	229	46	129	Q	2	42
06 354	115	18	131	55	26	9	8	569	311	146	184	1.17	9	88	25	ON	182	59	106	Q	QN	39
06 355	19	18	64	35	15	QN	QN	413	300	106	09	16.0	9	48	Q	QN	258	41	114	Q	42	48
06 356	8	18	15	38	15	9	Q	403	414	109	68	0.92	Q	TE	Q	Q	225	45	Ħ	NO N	R	33
00 357	34	18	99	38	12	Ø	55	393	234	106	88	0.87	R	Q	Q	QN	234	26	130	Q	20	43
06 358	81	23	120	23	24	2	48	433	906	66	191	1.16	2	4	24	CN I	227	142	164	Q I	12	5
06 359	-	12	132	97	3 :	2 9	59	524	099	5/5	555	1.21	2 9	2 9	2 9	2 9	18	16	2 5	2 9	20	8 8
196 200	8 %	t si			1	2 5		000	H OIL	20	COT	WL 1		8 5	2 5	2 9	130	ND N	211	2 9	2 9	
06 362	59	28	128	43	17	2	9	308	291	SH SH	III	0.94		25		9	204	64	169	Q	22	15
06 363	33	21	42	38	21	9	QN	318	585	105	128	1.09	9	11	9	QN	260	62	116	N	34	49
DG 364	IE	52	151	45	11	Q	42	368	550	122	157	21-1	9	QN	2	Q	215	114	110	Q	2	44
ore zone																						
06 365	16	56	11	37	61	2	8	269	610	S	108	0.92	9	36	8	Q	213	112	133	9	28	63
06 366	5	20	8	33	11	9	QN	318	684	NS	171	1.08		65	9	QN	232	135	131	NO	Q	85
06 367	12	23	64	43	18	Q	Q	249	394	NS	2	0.93	QN	8	Q	QN	228	98	168	DN	9	40
00 368	EI	24	40	47	20	2	Q	249	676	NS	129	1.15	2	83	R	Q	233	132	133	Q	2	22
00 369	52	24	52	34	15	2	2	209	400	96	100	0.99		Q	Q	Q	250	16	169	Q	20	43
00 370	52	20	8	43	2	9 9		502	540	98	6EI	66.0	2	61	2	2	195	125	114	2	24	S (
06 371	=	DZ S	5 :	2	11	2 9	2 9	662	101	21	132	1.15	2	8	53	9 1	235	168	147	9 9	9	5
2/5 50	g •	51	14	8	19	2 9	7	517	070	60	5	1.1	2 9	2 (2 1	2 1	215	5	112		s :	\$:
C/C 20	•	1	2	2	9	2	e	6	-	10	R	1.02	5	2	2	R	633	M	TAD		2	6

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Sample no.	3	qd	Zn	Ni	3	W	W	Min	Ba	5	>	1102	R	K03	¥	Sn	21	42	Sr	18	qu	*
	udd	wdd	udd	udd	udd	udd	qdd	wdd	udd	uxdd	udd	•	udd	udd	undd	udd	udd	udd	wdd	udd	udd	udd
06 374	14	21	84	40	16	Ð	ON	293	670	02	133	1.17	9	98	9	ON NO	162	149	129	2	QN	38
06 375	13	16	62	44	16	Q	Q	413	687	02	127	1.15	2	83	23	QN	235	140	123	2	Q	55
06 376	12	23	93	39	11	QN	Q	634	548	11	35	1.06	9	49	R	QN	268	113	145	UN	22	44
06 377	6	24	132	15	22	9	QN	745	858	96	154	1.15	9	87	32	QN	216	176	35	QN	36	89
06 378	15	20	11	32	15	2	QN	569	571	120	ш	1.16	9	36	9	Q	317	93	140	R	38	49
06 379	4	20	18	28	15	Q	Q	664	169	63	134	1.06	2	32	Q	QN	231	106	120	9	29	M
06 380	30	20	64	33	18	Q4	QN	564	495	105	67	1.09	Q	11	æ	Q	318	104	141	Q	39	63
06 381	10	20	108	42	10	2	QN	564	764	86	155	1.08	Q	54	2	Q	266	147	128	9	9	99
06 382	33	21	88	36	16	9	Q	372	417	11	121	1.05	Q	47	2	Q	294	104	151	9	27	47
06 383	6	19	110	42	15	Q	Q	393	744	88	100	1.33	Q	40	9	9	389	141	135	Q	9	52
BOREHOLE DGB 154A																						
in the second	2	4	-	10	5			-	-	2		CULT.	2	cut	4	2	1	-	2			
-out and inc			MOM	in mun	B mod	2 2		and and	Do muu	-	, mon	*			2	ue un	LT Dren	2	1			-
	add	mdd	und	mdd	udd	und	add	udd	inde	und	udd	•	mdd	udd	R		udd	udd	Se l	Rd	udd	Indd
06 630	16	20	92	38	11	Q	QN	395	687	63	124	1.12	R	55	9	QN	285	109	120	ON	26	8
06 631	108	19	113	4	21	9	45	526	432	46	163	0.99	9	62	Q	QN	239	81	196	9	24	52
06 632	35	16	11	177	28	9	QN	210	Q	150	205	0.81	Q	43	9	QN	69	ND	19	Q	Ð	26
06 633	9	15	35	48	14	g	QN	213	22	130	222	0.87	R	QN	Q	DN	104	QN	174	Q	Q	32
00 634	248	19	101	202	43	2	Q	496	26	113	235	1.04	Q	105	2	Q	67	Q	44	QN	Q	50
06 635	19	19	82	88	2	2	2	536	319	59	132	0.96	2	55	2	Ð	308	48	112	R	23	19
06 636	-	16	5	\$	2	2	Q :	51/	136	\$ 1	120	0.93	2	8	2	9	247	52	113	Q	88	8
06 637	8 1	2	151	99 :	2 3	2 9	a	166	351	33	157	0.14	2 9	3 5	2 9	2	161	13	78	9	22	Ħ
06 030	10	17	5 5	1	12	2 5		950	120	5 9	60	19.0	2 5	3 =	2 9	2 :	162	7	111		2	44
00 640	150	CT BI	124	19	1	2 -	2 5	UEB	626	Se Se	200	1 71	2 5	102	2 %	17	0/2	47	20	2 9	9 8	2
06 641	ye.	8	105	44	50	Ş	9	115	254	3 9	111	101	2 5		3 5	2 5	OVC	10	8 8	2 9	3 5	7 8
06 647	9	11	8	40	28	9	3 13	466	673	25	149	00.0	2 5	20	2 9	2 5	145	3 9	145		2 2	2 5
06 643	65	1	96	23	15	2	9	213	ON	-	380	1.52	9	34	9 8	2	142	3 9	134	2 9	2 9	1 9
06 644	31	13	8	20	12	9	Q	253	Q	20	397	1.77	9	Q	12	9	147	Q.	144	2	9	42
06 645	£	15	92	88	12	9	48	466	113	61	300	1.38	Q	82	Q	Q	130	33	110	Q	QN	43
06 646	5	15	89	40	19	Q	Q	395	251	30	140	0.94	Q	80	9	QN	242	47	131	QN	Q	40
00 647	=	15	82	89	24	Q	QN	455	212	101	184	0.99	Q	40	2	Ŋ	186	25	121	ND	39	44
00 648	12	14	25	4	81	2 9	2 9	466	210	29	156	0.94	8	41	9 1	2	263	36	101	모	37	23
DC 650	9 9	10		5	9 5	2 9		100	183	2 22	911	0.09	2 9	43	2 9	2 9	162	10	120	2 9	2 9	5
06 651	8	25	102	3	12	9	-	850	317	65	214	1 04		3 4	2 5	36	226	9 0		2 9		3 9
06 652	42	11	19	49	22	2	45	466	262	69	113	0.91	9	15	2	9	248	96	121	2 2	1 9	2
06 653	11	18	28	40	16	9	Q	409	425	61	114	1.06	9	25	9	QN	248	81	66	24	9	14
06 654	55	25	25	54	24	g	43	370	815	72	169	1.10	2	16	23	QN	216	133	82	Q	21	62
0G 655	19	22	8	44	11	2	3	469	773	29	111	1.02	9	81	9	Q	235	96	87	20	34	19
06 656	10	19	45	9	9	9	Q	389	205	74	61	0.78	Q	65	Ð	Q	210	49	11	Q	Q	27
00 657	9	11	11	33	12	2	9	629	488	5	107	1.05	9	18	P	Q	335	63	212	DN	Q	61
06 658	81	24	8	36	4	9	9	389	778	69	104	0.65	Q	82	2	23	126	26	2	Q	32	23
06 659	489	22	£	51	19	2	45	448	328	2	76	0.61	2	20	2	QN	141	32	9	Q	Q	41
ore zone																						
				1	2		1	100	And .	P												
06 660	8	2	14	8	51 22	2 9	1	280	000	5	138	0.94	9 9	112	9		230	8	16	9	55	8
T00 90	OT I	9	8	9	5	2	ş	NCC	640	R	167	60'T	8	2	2	Q	515	133	53	N	33	8

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BOREHOLE DGB 153A

iample no.																					
	C.	ab Zn	N1	3	Pg.	Au	W	Ba	5	>	1102	As	604	Po	Sn	Zr	Rb	Sr	81	QN N	*
	id udd	udd wo	bba	udd	udd	dqq	udd	udd	bba	udd	*	udd	undd	udd	udd	udd	mqq	edd	udd	mdd	udd
06 662	114 1	8 31	æ	18	9	QN	280	1129	62	129	1.04	Q	83	g	Q	266	120	19	QN	45	50
00 663	27 2	ES 11	36	16	GN	20	340	1405	74	152	1.01	2	69	QN	NO	208	124	52	QN	Q	61
00 664	26 1	15 66	37	18	Q	QN	260	559	54	114	1.01	Q	60	QN	Q	263	110	16	R	ą	48
06 665	29 2	55 53	42	19	QN	QN	260	724	69	138	1.10	Ŗ	11	QN	QN	238	146	100	QN	34	64
06 666	13 2	19 Ei	40	24	Q	Q	230	704	25	152	1.05	QN	36	QN	QN	230	132	100	QN	QN	59
06 667	19 1	13 73	34	61	QN	CH	300	406	72	84	0.98	QN	37	NO	24	273	92	159	QN	01	59
00 668	14 2	101 102	4	22	Q	QN	280	660	64	159	1.07	Q	83	Q	Q	228	136	150	R	31	57
06 669	2	104	48	22	Q	ND	240	1071	69	212	1.17	Q	13	Q	9	228	191	114	QN	2	56
06 670	25 1	18 75	M	18	Q	Q	200	387	104	72	0.94	Q	55	21	Q	243	66	151	Q	52	48
06 671	42 1	16 45	33	50	QN	QN	190	405	104	106	0.79	QN	ND	QN	QN	218	36	114	ON	ON	47
06 672	10 2	19 61	40	56	Q	QN	280	166	94	145	1.08	P	72	Q	QN	237	111	152	QN	29	41
06 673	9	12 76	40	21	Q	QN	200	486	51	22	1.00	Q	80	QN	ND	249	86	176	QN	22	58
06 674	7 2	37	56	22	Q	R	280	765	74	204	1.17	QN	96	R	2	218	11	33	QN	R	69

BOREHOLE DGB 154A

trace element geochemistry analyses by Scientific Services laboratories}.
Table 6 - Organio Prospect: of the are-zone rock types and Tsumeb Corporation Ltd.

BOREHOLE No. 0G8 151A

1 1																							
000 000 <td></td> <td>*</td> <td>udd</td> <td>udd</td> <td>udd</td> <td>udd</td> <td>Edd</td> <td>mdd</td> <td>udd</td> <td>udd</td> <td>udd</td> <td>und</td> <td>ppr</td> <td>udd</td> <td>udd</td> <td>bbu</td> <td>udd</td> <td>udd</td> <td>udd</td> <td>udd</td> <td>mdd</td> <td>udd</td> <td>•</td>		*	udd	udd	udd	udd	Edd	mdd	udd	udd	udd	und	ppr	udd	udd	bbu	udd	udd	udd	udd	mdd	udd	•
061 103 2 2 0 1 2 0 1 2 0 1 2 0 1 2 0 1 2 0 <td>448 GF</td> <td>0.13</td> <td>30</td> <td>208</td> <td>36</td> <td>23</td> <td>1</td> <td>N/S</td> <td>268</td> <td>486</td> <td>21</td> <td>76</td> <td>552</td> <td>U/N</td> <td>34</td> <td>N/D</td> <td>N/D</td> <td>110</td> <td>31</td> <td>22</td> <td>N/D</td> <td>N/D</td> <td></td>	448 GF	0.13	30	208	36	23	1	N/S	268	486	21	76	552	U/N	34	N/D	N/D	110	31	22	N/D	N/D	
010 1.8 3 <td>449 GF</td> <td>0.45</td> <td>32</td> <td>265</td> <td>36</td> <td>33</td> <td>m</td> <td>0.11</td> <td>318</td> <td>557</td> <td>9</td> <td>130</td> <td>611</td> <td>Q/N</td> <td>11</td> <td>22</td> <td>N/D</td> <td>142</td> <td>58</td> <td>54</td> <td>N/D</td> <td>Q/N</td> <td>×</td>	449 GF	0.45	32	265	36	33	m	0.11	318	557	9	130	611	Q/N	11	22	N/D	142	58	54	N/D	Q/N	×
010 110 10 0	450 GF	1.58	28	315	31	114	6	0.28	377	704	56	64	538	N/D	Q/N	31	Q/N	16	62	12	N/D	N/D	
010 010 <td>451 GF</td> <td>1.48</td> <td>16</td> <td>261</td> <td>37</td> <td>66</td> <td>8</td> <td>0.63</td> <td>298</td> <td>435</td> <td>11</td> <td>8</td> <td>612</td> <td>N/D</td> <td>N/D</td> <td>Q/N</td> <td>N/D</td> <td>66</td> <td>36</td> <td>13</td> <td>N/D</td> <td>N/D</td> <td>N</td>	451 GF	1.48	16	261	37	66	8	0.63	298	435	11	8	612	N/D	N/D	Q/N	N/D	66	36	13	N/D	N/D	N
0.2 0.13	452 GF	2.42	11	199	42	244	13	0.12	59	R	51	R	n	N,	UR	N	S	B	R	N	R	N	
65 7.17 90 90 30 90 30 90 30 90 30 9	453 GF	0.18	39	138	38	30	2	0.04	333	N	168	UR	n	R	R	UR	S	5	M	R	N	R	
65 ff 0.11 30 35 30 <t< td=""><td>454 GF</td><td>2.72</td><td>20</td><td>949</td><td>28</td><td>127</td><td>21</td><td>1.08</td><td>199</td><td>962</td><td>53</td><td>32</td><td>352</td><td>Q/N</td><td>O/N</td><td>12</td><td>O/N</td><td>995</td><td>39</td><td>Q/N</td><td>N/D</td><td>N/D</td><td>z</td></t<>	454 GF	2.72	20	949	28	127	21	1.08	199	962	53	32	352	Q/N	O/N	12	O/N	995	39	Q/N	N/D	N/D	z
65 F 0.1 21 23 <th2< td=""><td>455 GF</td><td>E1.0</td><td>30</td><td>206</td><td>38</td><td>R</td><td>2</td><td>10.0</td><td>298</td><td>1900</td><td>23</td><td>20</td><td>870</td><td>0/W</td><td>55</td><td>Q/N</td><td>O/H</td><td>164</td><td>43</td><td>69</td><td>N/D</td><td>20</td><td></td></th2<>	455 GF	E1.0	30	206	38	R	2	10.0	298	1900	23	20	870	0/W	55	Q/N	O/H	164	43	69	N/D	20	
9.6 0.3 2.3 2.3 3.1 0.0 3. 0.0 3. 0.0	456 GF	0.27	28	215	30	53	2	0.05	308	2157	50	11	376	N/D	60	27	U/N	176	49	48	N/0	D/N	
64 F 0.03 32 1 1 0.13 13 14 15 14 15 14 15 14 15 14 15 14 15 16 <th< td=""><td>457 GF</td><td>m</td><td>22</td><td>323</td><td>25</td><td>611</td><td>20</td><td>0.31</td><td>199</td><td>1125</td><td>19</td><td>23</td><td>5/T</td><td>U/N</td><td>N/D</td><td>48</td><td>Q/N</td><td>30</td><td>12</td><td>N/D</td><td>N/D</td><td>N/D</td><td>1</td></th<>	457 GF	m	22	323	25	611	20	0.31	199	1125	19	23	5/T	U/N	N/D	48	Q/N	30	12	N/D	N/D	N/D	1
65 F 0.08 36 9 7 1 N/2 23 144 34 135 100 75 101 130 101 130 101 130 101 130 101 130 101 130 101 130 101 130 101 130 101 130 101 130 101 130 101 130 101 130 101	458 GF	0.93	32	168	21	47	8	0.14	137	N	186	R	5	RI	UR	5	ŝ	s	s	ß	R	M	
International state Internatindindicate Internatindindicate	459 GF	0.05	26	86	48	27	-	N/S	228	1444	34	126	110	Q/N	36	O/N	Q/N	233	ш	129	0/N	24	
Up is in the pair pair pair pair pair pair pair pair	LEHOLE No. 06	AE21 B																					
4 PPP	ple no.	З	qd	Zn	N	9	Ag	Au	W	Ba	5	>	1102	As	NO3	Mo	Sn	Ir	đã	Sr	9	đ	
60 6 0.01 3 10		r.	udd	udd	udd	udd	wdd	udd	udd	udd	udd	udd	udd	udd	udd	udd	udd	udd	udd	wdd	udd	wdd	
616 2.71 20 63 2 1 0.44 53 15	460 GF	0.04	38	83	47	23	N/D	N/S	288	678	38	150	110	N/D	G	Q/N	U/D	206	105	100	N/D	32	
66 ff 1.02 2 1.00 4.1 1.00 4.1 1.00 1.01 1.0	461 GF	2.71	20	405	24	67		0.14	66	IS	IS	15	1	IS	IS	IS.	SI	51	IS	IS	IS	15	
616 3.8 3.8 56.1 2 113 14 100 101 100	462 GF	0.22	24	170	64	42	2	0.06	338	1046	63	122	887	N/D	44	0/14	Q/N	161	69	37	N/0	U/N	
646 10.2 71 205 71 205 71 205 71 205 71 205 71 205 71 205 71 205 71 205 71 205 71 205 71 205 71 205 <t< td=""><td>463 GF</td><td>3.89</td><td>28</td><td>199</td><td>22</td><td>112 -</td><td>16</td><td>0.25</td><td>228</td><td>333</td><td>46</td><td>19</td><td>191</td><td>R/N</td><td>d/N</td><td>23</td><td>O/N</td><td>24</td><td>11</td><td>N/D</td><td>N/D</td><td>N/D</td><td></td></t<>	463 GF	3.89	28	199	22	112 -	16	0.25	228	333	46	19	191	R/N	d/N	23	O/N	24	11	N/D	N/D	N/D	
656 11.3 54 100 53 11 0.0 150 0.0	464 GF	0.52	74	275	21	54	m	0.06	407	1552	9	11	855	N/D	11	22	Q/N	212	106	99	N/D	N/D	
656 0.03 56 13 70 0.06 236 125 25 13 910 70 71 70 71 70 <	465 GF	11.75	44	1700	43	41	99	3.78	139	253	п	O/N	1631	N/D	D/N	31	N/D	0	N/D	D/N	N/D	N/D	
60F 5.3 13 130 25 13 130 13 130 13 130 13 130 13 130 13 130 13 130 13 130 13 130 13 130 13 130 13	466 GF	0.03	26	159	98	8	0/N	0.08	248	1257	52	83	066	N/D	99	0/N	Q/N	252	59	156	N/D	21	
668 1.13 4.9 1.43 1.3 2.9 3.0 6.9 1.13 6.9 6.9 1.13 6.9 7.9 6.9 7.9 6.9 7.9 6.9 7.9	467 64	8.45	2	1900	26	120	37	0.16	199	S/N	M	N/S	-/N	S/N	S/N	S/N	N/S	N/S	N/S	N/S	N/S	N/S	
000 0.10 030 23 <th< td=""><td>400 LT</td><td>3.13</td><td>14</td><td>914</td><td>8</td><td>5</td><td>5 9</td><td>1.13</td><td>196</td><td>8 9</td><td>99</td><td>8</td><td>5 5</td><td>5 9</td><td>8</td><td>5 9</td><td>¥ :</td><td>5</td><td>g i</td><td>5</td><td>an l</td><td>5</td><td></td></th<>	400 LT	3.13	14	914	8	5	5 9	1.13	196	8 9	99	8	5 5	5 9	8	5 9	¥ :	5	g i	5	an l	5	
717 0.11 60 27 35 40 10 0.01 20 34 20 10	470 GF	1 72	RA BA	510	3 2	136	5 9	0.75	167	5	UE UE	5 9	5 3	5 9	5 9	5 9	¥ 9	5 9	¥ 9	5 9	X S	5 9	
472 F 0.56 51 334 32 43 17 16 17 16 17 16 17 16 17 16 17 16 17 16 17 16 17 16 17 16 16 17 16 16 17 16 17 16 17 16 17 16 17 16 17 16 17 16 17 16 16 17 16 16 16 16 16 17 16 17 16	471 GF	0.14	09	247	36	40	3	0.04	284	5	216	M	5 3	1 9	g	5 9	5 9	5 8	5 9	5 9	5 9	5 9	
173 C 0.61 66 20 16 42 18 </td <td>472 GF</td> <td>0.56</td> <td>15</td> <td>334</td> <td>32</td> <td>43</td> <td>¥</td> <td>0.19</td> <td>206</td> <td>5</td> <td>204</td> <td>R</td> <td>5</td> <td>3</td> <td>3</td> <td>s</td> <td>R3</td> <td>9</td> <td>i an</td> <td>9</td> <td>M</td> <td>i M</td> <td></td>	472 GF	0.56	15	334	32	43	¥	0.19	206	5	204	R	5	3	3	s	R3	9	i an	9	M	i M	
474 E 1.41 61 325 30 60 18 0.4 161 174 18 19 19 19 19 18 11	473 GF	0.61	46	240	91	42	R	0.4	167	¥,	192	R	Sec.	RR N	R	M	5	N	ß	R	M	B	
475 (F 0.82 31 339 31 93 UR 0.24 343 UR 133 UR <td>474 GF</td> <td>16.1</td> <td>19</td> <td>325</td> <td>30</td> <td>99</td> <td>N</td> <td>0.4</td> <td>167</td> <td>R</td> <td>174</td> <td>N</td> <td>LIR.</td> <td>95</td> <td>R</td> <td>B</td> <td>M</td> <td>g</td> <td>BU</td> <td>UR</td> <td>UR</td> <td>R</td> <td></td>	474 GF	16.1	19	325	30	99	N	0.4	167	R	174	N	LIR.	95	R	B	M	g	BU	UR	UR	R	
476 F 0.15 35 138 36 21 UR 0.05 225 UR 162 UR	475 GF	0.82	IE	339	31	66	M	0.24	343	R	138	R	R	S	M	M	B	5	R	ŝ	NR N	un	
FMO.E No. 0GB 154A ple No. Cu Pb Zn NI Co Ag Au Pa Ba Cr V 1102 As NO3 No Sn Zr Rb Sr Bi Nb Ple No. Cu Pb Zn NI Co Ag Au Pn Ba Cr V 1102 As NO3 No Sn Zr Ri Nb ppm	476 GF	0.15	35	138	36	21	¥	90.0	225	ä	162	R	UR	s	UR	NI I	ä	SI I	M	¥.	ur,	R	
qle ro. Cu Pb Zn NI Co Ag Au Hn Ba Cr V 1102 As W3 Ho Sn Zr Rb Sr Bi Sn Sn Zr Rb Sr Bi Sn Sn Zr Rb Sr Bi Sn Zr N Sn Zr Bi Sn Zr Bi Sn Zr N Sn Zr Bi Sn Sn Zr Sn Sn <th< td=""><td>EHOLE No. OG</td><td>B 154A</td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td></th<>	EHOLE No. OG	B 154A																					
* Ppm	ple no.	3	Pb	Zn	W	3	by	Au	Ŧ	84	5	A	1102	As	EOM	£	5	12	ßb	Sr	18	ą	
425 GF 0-17 18 160 42 20 ND 0-03 288 447 17 95 7839 N/0 N/0 N/0 N/0 174 48 46 N/0 N/0 N/0 42 67 1.05 22 411 39 38 8 0.32 348 575 21 76 6635 N/0 N/0 N/0 N/0 115 87 41 N/0 N/0 N/0 42 67 2.77 32 1100 29 121 17 0.39 89 107 29 24 1133 21 N/0 23 N/0 13 N/0 110 N/0 N/0 N/0 42 67 0.03 24 35 46 23 00 0.07 407 549 186 82 647 N/0 52 N/0 12 89 32 N/0 10 12 87 41 0.00 10 42 67 0.03 24 23 66 0.07 24 2972 110 23 110 22 100 23 100 23 20 0.07 100 24 24 24 24 25 113 9972 N/0 52 N/0 12 23 100 20 10 10 10 10 10 10 10 10 10 10 10 10 10		*	udd	udd	uxdd	udd	udd	udd	udd	udd	udd	udd	udd	udd	wdd	udd	udd	uxid	udd	mqq	udd	mdd	
426 GF 1.05 Z2 411 39 38 8 0.32 348 575 21 76 6635 N/D N/D N/D N/D 115 87 41 N/D N/D N/D 427 GF 2.77 32 1100 29 121 17 0.39 89 107 29 24 1133 21 N/D 23 N/D 13 N/D 170 N/D N/D 426 GF 0.02 29 31 40 31 2 0.07 407 549 186 82 647 N/D 52 N/D 102 89 32 N/D 24 436 7 0.07 25 113 9972 N/D 52 N/D 52 N/D 25 106 80 N/D 0.07 107 107 107 107 107 107 107 107 107 1	425 GF	0.12	18	160	42	20	Q	E0.03	288	447	17	95	7839	N/D	N/D	N/D	N/D	174	48	46	N/O	N/D	
427 GF 2.77 32 1100 29 121 17 0.39 89 107 29 24 1133 21 N/D 23 N/D 13 N/D N/D N/D N/D N/D 124 20 0.25 29 313 40 31 2 0.07 407 549 186 82 6427 N/D 62 N/D N/D 132 89 32 N/D 24 429 GF 0.03 24 85 46 25 N/D 0.07 218 777 25 113 9972 N/D 52 N/D 20 25 106 80 N/D N/D N/D N/D 20 25 106 20 N/D N/D 20 25 20 25 20 25 20 20 20 20 20 20 20 20 20 20 20 20 20	426 GF	1.05	22	411	39	88	8	0.32	348	575	21	36	6635	N/D	D/N	N/D	Q/N	115	87	41	0/1	N/D	
428 GF 0.25 29 313 40 31 2 0.07 407 549 186 82 6427 N/D 62 N/D N/D 132 89 32 N/D 24 429 GF 0.03 24 85 46 25 N/D 0.07 218 777 25 113 9972 N/D 52 N/D N/D 255 106 80 N/D N/D	427 GF	2.77	32	1100	29	121	17	0.39	68	107	29	24	1133	21	N/D	53	N/D	13	N/D	N/D	Q/N	Q/N	z
429 GF 0.03 24 85 46 25 NO 0.07 218 777 25 113 9972 N/O 52 N/O N/O 225 106 80 N/O N/O	428 GF	0.25	29	313	40	31	2	0.07	407	549	186	82	6427	N/D	62	N/D	N/D	132	68	32	N/D	24	
	429 GF	0.03	24	85	46	25	QN	0.07	218	111	25	113	0072	N/N	63	N/D	N/0	225	106	BU	NIN.	NIN	

Note: UR - unavallable result; N/S - insufficient sample; N/D - below detection limit.

Table 7 - Trace element geochemistry of the unmineralized schists.

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sample no.	Cu	Pb	Zn	NI	Co	Ag	Au	Hn	Ba	Cr	V	Ti	As	HO3	Но	Sn	Zr	Rb	Sr	Bi	Nb	۲
	ppm	ppm	ppm	ppm	ppm	ppm	ppb	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm
OG 71	80	29	92	41	29	ND	115	310	724	62	146	9900	ND	56	ND	63	252	128	128	36	ND	ND
OG 119	160	24	131	58	33	2	95	351	838	7	204	11000	ND	107	ND	32	242	130	98	36	ND	ND
OG 129	34	36	79	52	27	ND	66	260	750	17	119	10600	ND	ND	ND	ND	209	131	105	ND	ND	60
OG 130	42	41	108	67	28	ND	46	260	971	10	227	10200	ND	ND	ND	ND	203	157	86	ND	ND	ND
OG 131	54	41	95	85	25	ND	46	150	952	7	213	9480	NO	46	ND	ND	245	149	85	ND	ND	65
OG 132	21	26	77	59	26	ND	63	210	909	7	205	9757	NO	ND	ND	ND	254	141	83	ND	ND	ND
OG 133	44	38	114	39	27	ND	63	190	776	10	227	10400	ND	ND	ND	ND	213	133	107	ND	ND	ND
OG 134	59	36	122	47	24	ND	74	130	732	17	183	10800	NO	ND	ND	ND	229	139	111	ND	ND	56
OG 135	64	38	115	56	20	ND	61	190	940	27	218	9945	ND	ND	ND	ND	204	139	90	ND	ND	ND
OG 136	32	34	111	51	21	ND	63	160	672	22	152	9836	ND	ND	ND	ND	212	128	96	ND	ND	ND
x	59	30.6	104.4	55.3	26	0.1	69.3	221.1	758.4	18.6	189.4	10200	ND	20.9	ND	9.5	226.3	137.5	98.9	7.2	ND	18

Table 8 - Ongeama Prospect: trace element geochemistry of the magnetic schists from the "I anomaly" area.

sample no.	Cu	Pb	Zn	NS.	Ca	Ag	Au	Mn	Ba	Cr	v	Tİ	As	HO3	Ho	Sn	Zr	Rb	Sr	Bi	Nb	Y
	ppm	ppm	ppm	ppm	ppm	ppm	ppb	ppm	ppm	ppm	ppm	ppm	рря	ppm								
0G 137	44	32	90	56	17	NO	58	350	568	12	124	10400	NO	ND	ND	ND	264	100	91	ND	ND	ND
OG 138	92	33	84	43	16	ND	53	579	201	17	146	8873	ND	38	ND	ND	192	25	B1	ND	ND	ND
OG 139	38	33	78	34	13	ND	46	300	532	12	133	9583	ND	ND	ND	ND	231	100	93	ND	ND	67
OG 140	43	30	104	44	17	ND	51	459	715	17	141	10600	ND	ND	ND	ND	235	108	93	ND	ND	ND
OG 141	43	31	85	45	11	ND	43	409	533	7	101	9985	ND	ND	ND	ND	294	104	111	ND	ND	ND
OG 142	38	31	76	29	11	ND	56	330	624	12	97	9269	ND	ND	ND	ND	246	96	107	ND	ND	55
OG 143	45	30	74	34	11	ND	66	310	487	7	110	10400	ND	ND	ND	ND	259	96	122	ND	ND	ND

Table 9 - X-Ray fluorescence analy	cal conditions	and rock	standards	used.
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Element	Emission	Line	Tube	kV	mA	Cry	ystal	Time	Counter	Collimator	Spec imen
Si	K		Cr	55	40		PET	40	flow	coarse	fusion disc
Ti	К		Cr	55	40	LIF	(200)	10	flow	fine	fusion disc
A1	К		Cr	55	40		PET	40	flow	coarse	fusion disc
Fe	К		Cr	55	40	LIF	(200)	20	flow	fine	fusion disc
Mn	к		Cr	55	40	LIF	(200)	20	flow	coarse	fusion disc
Mg	к		Cr	55	40		TLAP	100	flow	fine	fusion disc
Ca	К		Cr	55	40	LIF	(200)	10	flow	fine	fusion disc
Na	K		Cr	55	40		TLAP	100	flow	fine	powder pellet
К	K		Cr	55	40	LIF	(200)	10	flow	fine	fusion disc
Р	К		Cr	55	40		Ge	20	flow	coarse	fusion disc

Rock standards

Major elements AGV-1, BCR-1, GSP-1, NIM-D, NIM-N

Na BCR-1, GSP-1, AGV-1, G-2

Table	10	-	Ongeama	Prospe	ect:major	element	geochemistry	of	selected
			samples	s from	the alte	ration p	ipe.		

	OG 37 (FS)	OG 43 (MM)	OG 52 (MM)	OG 62 (MM)	OG 75 (MM)	0G 77 (CS)	0G 80 (CS)	0G 84 (CS)	0G 87 (CS)
Si02	19.176	58.314	36.315	72.412	64.328	60.672	63.178	64.164	65.983
Ti02	0.417	0.606	0.54	0.67	0.639	0.793	0.785	0.773	0.773
A1203	3.618	7.931	4.171	7.258	9.838	13.516	12.107	11.494	12.075
Fe203	66.407	30.103	50.594	16.395	19.032	19.951	19.772	17.675	16.099
Fe0	0	0	0	0	0	0	0	0	0
MnO	0.61	0.091	0.083	0.091	0.071	0.171	0.083	0.11	0.091
MgO	0.246	0.277	0.699	0.354	2.053	3.42	2.955	3.085	2.576
CaO	0.231	0	0.105	0	0	0.034	0	0.018	0.003
Na20	4.198	0.381	0.118	0.542	0.331	0.091	0.142	0.09	0.308
K20	0	1.125	0.856	1.274	1.047	0.054	0.153	0.054	0.0726
P205	0.285	0.146	0.214	0.082	0.087	0.136	0.131	0.101	0.102
LOI	7.68	3.55	6.07	2.54	2.5	2.6	2.71	2.54	2.71
H20-	1.2	0.6	1	0.4	0.18	0.18	0.17	0.16	0.15
	103.619	103.125	100.496	102.018	100.106	101.617	102.186	101.265	101.598

Note: FS = ferruginous schist; MM = magnetic muscovite schist; CS = magnetic chlorite schist.

	MWE 202	MWE210	MWE 219	MWE 226	MWE 234	
S 102	70.529	94.834	63.606	51.81	76.579	
T102	0.802	0.021	0.312	0.741	0.302	
A1203	11.71	0.274	3.687	8.837	1.451	
Fe203	7.611	3.257	26.514	29.701	18.931	
Fe0	0	0	0	0	0	
MnO	0.095	0.014	0.045	0.096	0.031	
MgO	3.446	0.013	0.214	0.414	0.016	
CaO	0.047	0	0.02	0.02	0.011	
Na20	0.373	0.013	0.176	0.321	0.0529	
K20	2.125	0	0.806	2.425	0	
P205	0.184	0.077	0.16	0.208	0.05	
LOI	3.04	0.53	3.65	3.85	1.87	
H20-	0.52	0.17	0.7	0.91	0.49	
	100.482	99.318	99.892	99.312	100.26	

Table 11 - Matchless West Extension deposit:major element geochemistry of selected samples from the alteration pipe.

Note:

MWE 202 = phlogopite-bearing muscovite-chlorite schist (marginal zone)

MWE 210 = pyrite-bearing muscovite quartzite (axial zone)

MWE 219 = pyrite-bearing quartz-muscovite schist (marginal zone)

MWE 226 = pyrite-bearing quartz-muscovite schist (marginal zone)

MWE 234 = pyrite-bearing quartz-muscovite schist (marginal zone)

Table 12 - Major element geochemistry of unmineralized metapelitic schists (Kuiseb Formation).

*	OG 131	OG 135	OG 138	e
5102	55,323	6.079	69,912	-
Ti02	1.048	1.02	0.856	
A1203	19.298	17.215	12.339	
Fe203	9.189	8.264	7.996	
Fe0	0	0	0	
MnO	0.151	0.093	0.111	
MgO	4.146	3.701	2.273	
Ca0	0.993	0.982	2.62	
Na20	0.916	1.251	2,597	
K20	4.185	3.87	0.492	
P205	0.211	0.276	0.254	
LOI	3.07	2.37	1.64	
H20-	0.93	0.22	0.24	
	99.458	100.052	101.33	

Table 13 - Subdivision of the alteration pipe at Ongeama in sectors:average of the trace element content for each sector.

3	Pb	μŢ	N1	8	Ag	W	ራ	¥	Ba	A	1	8	r	Zr	Sn	*	£	As	Ð	٨	81
- 1	udd	sid	udd	Edd	udd	udd	udd	bpm	udd	udd	udd	đ	Ma i	E d	udd	iidd	Ma	udd	wdd	udd	udd
0	101.3	133	28.7	1.15	0.03	0.164	42.3	303	1451	107.7	IE0%	22	34	127.3	10.7	27.3	0	7.3	0	•	6.7
2	111.7	232.7	26.7	34.3	0.36	0.336	52.3	319.7	1203	142	6500	26	31.3	6.7	13.33	•	•	0	10	0	14
20	106.3	193.3	23.7	24.7	0.7	0.328	52.3	342.7	3402	100.3	5466.7	20.7	50	126.7	25.7	47.3	۰	16.3	0	0	24
20	22	73	30.7	26.7	0.05	0.166	53	309.7	1800	128	4175	10	28	156	22.7	24.7	0	173.7	0	0	41.3
20	28	53.5	34	17.5	0.05	0.122	15	305	1522	138	2000	w.	23	149.5	4	0	•	179.5	0	0	E
20	31	53.5	Ħ	19.5	0.05	0.153	32.5	195.5	1018	109	05E9	п	54	173.5	38	0	0	161	0	0	26
8	21	32.5	31.5	11	0.05	0.14	п	291	155	82	5600	5	12	146	18	37.5	•	144	19	•	30

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Sector I includes samples 06 43, 06 46, 06 49 Sector II includes samples 06 22, 06 55, 06 58 Sector III includes samples 06 50, 06 77, 06 75 Sector VI includes samples 06 50, 06 77, 06 78 Sector V includes samples 66 80, 06 71, 06 78

Table 14 - Subdivision of the alteration pipe at Matchless West Extension in sectors: average of the trace element content in each sector.

18	0	0	9.5	•	0	27.2	
¥ wdd	0	•	•	•	15	•	
qu udd	0	•	•	0	0	•	
As an	0	•	12.25	12	18.5	12.4	
왉쎫	8.25	0	7.8	86.75	60.25	24	
mqq	19.5	0	0	55.25	13.5	28.2	
Sn	0	a	•	•	30.25	13.2	
77 bpg	206.5	196	132.25	145	145.25	84.8	
r2 mg	19.75	16	12	٥	36.75	32	
dh mqq	58.75	11	39.75	57.5	38.25		
11 Wdd	7440.5	8172	5207.75	6101.3	967.75	3806.8	
A mdd	109.5	127.5	06	136	249.25	105.8	
Ba	246.35	1710.5	1681.5	2758.5	3947	4135.8	
w	159.5	65	294.25	72.5	169.25	141.8	
Ь ud	10.25	27	18.5	23	23.5	20	
W	0.052	0.054	0.064	0.11	0.095	560.0	HHC 205
Ag	0.75	0.05	5.25	0.5	1.25	1.4	ME 204, 1
Co ppm	18.75	12.5	14.75	17.75	18.5	17.4	MHE 202,
in	27.25	19.5	22.25	26.5	22.5	24.2	ME 201.
mdq	43	23.5	54.3	34.25	101.25	851.16	samp les
9d dd	37	19	174.5	82.25	65.5	574.6	finc ludes
bpm	1600	1600	1900	1800	5400	4100	actor 1
Sector	-	н	ш	M	٨	IA	Note: St

Sector II includes samples MKE 206, MKE 209 Sector III includes samples MKE 210, MKE 211, MKE 215 Sector IV includes samples MKE 210, MKE 211, MKE 215 Sector V includes samples MKE 221, MKE 224, MKE 226 Sector VI includes samples MKE 232, MKE 234, MKE 235, MKE 238

Table 15 - Ongeama Prospect: barium and zinc contents in the schists of the alteration pipe.

UNGEAMA TRACE ELEMENT GEOCHEMISTRY - ALTERATI	UN	PIPE
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distance	Ba	Zn	
	ppm	ppm	
-101 4	204	32	
-84.3	106	33	
-61.4	1648	55	
-47.1	388	52	
-32.9	2338	54	
-25.7	706	53	
-16.7	2662	61	
-10	414	48	
-8.6	2324	110	
5.7	4200	145	
37.1	2875	188	
62.9	3130	247	
78.6	1676	181	
95.7	1100	249	
112.9	834	268	
134.3	988	144	
150	1308	115	
170	2057	140	

							ONGOMBO PRE	SPECT - T	HE ORE-HO	STING SEQUENC	E IN DRILL COF	ES							Table 16.h		
ahle 16 a			14016 10.0				Table 16.e				Table 16.1				Table 16.0						
			ONGEAMA PROSPE	ECT: MAGNETIC S	STR151S							and the second se							I major FOR SC	HISTS FROM A	LTERATION
UNHINERAL IZED SCHIS	۶ I		FROM THE "I AN	NOHALY" AREA			BOREHOLE OF	151A			BOREHOLE NO	. 068 153	- 1		BOREHOLE	NGB 154A			AND UNMINERAL I	ZED SCHISTS	(*)
l north	trace	ш	samle no.	1 trace	11		sample no.		trace	77	sample no.		trace	17	sample no.		I trace	11	sample no.	I major	17
sample no.	51919	:	- All Station		1				ġ					1		1			nr 33 -	0.0	
11 50	0.802	95.0	06 137	0.793	0.39		06 384	(H/H)	5/-0	0.448	00 322	(H/H)	0.724	0.420	06 630	(4/4)	0.78	141 U	06 43 *	0.962	0.485
06 119	0.818	Ch-0	06 138	0.770	0.40		982 90	(H/H)	0.72	0.398	06 324	(H/H)	0.735	0.412	06 635	(M/H)	EL-0	0.312	• 25 • 06	0.979	1.13
00 120	0.877	50	001 140	0.816	0.45		06 387	(H/H)	0.69	0.408	06 325	(H/H)	0.781	0.475	06 636	(H/H)	0.65	0.374	• 29 90	0.903	0.425
06 131	0.83	0.39	141 00	0.763	0.34		06 388	(H/H)	0.65	0.43	06 326	(H/H)	0.665	0.348	06 638	(H/H)	0.71	0.385	06 75 *	0.939	0.358
06 132	0.822	0.38	06 142	162.0	0.38		062 390	(H/H)	0.71	0.347	0G 327	(H/H)	0.826	0.512	06 639	(H/H)	0.69	0.315	- 11 - 90	0,992	0.396
06 133	0.802	0.49	06 143	0.745	0.4		16E 90	(H/H)	0.77	0.471	06 328	(H/H)	. 169.0	0.372	06 641	(H/H)	0.77	0.42	• 08 90	0.987	0.418
06 134	167.0	0.47					265 50	(H/H)	69.0	0.27	06 329	(H/H)	0.795	0.433	06 642	(H/H)	0.8	0.411	* 56 90	0.988	0.376
00 135	0.835	0.49					06 393	(W/H)	2/-0	0.355	155 20	(14/14)	122 0	205-0	040 040	(#/#)	0.00	0.300		01.5.0	8/2.0
06 136	167.0	0.46					465 00	(H/H)	0.75	0 45	155 00	(H/H)	0 71	0.387	00 648	(4/4)	0.00	955.0	MKE 202 *	0.814	0.35
							865 90	(H/M)	6.0	0.43	06 333	(H/H)	0.738	0.433	06 650	(H/H)	0.64	0.349	MAE 210 *	0.962	0.17
							06 399	(H/H)	0.75	0.463	06 334	(H/H)	0.661	0.365	00 651	(H/H)	0.75	0.46	MHE 219 *	0.964	0.57
							06 400	(H/H)	0.78	0.397	06 335	(H/H)	0.773	0.493	06 652	(H/H)	0.7	0.368	MME 226 *	116.0	0.49
							00 601	(H/H)	0.84	0.438	00 336	(M/H)	0.805	0.445	00 653	(H/H)	0.73	0.427		1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1	20.0
Table 16.c			Table	e 16,d			06 602	(H/H)	0.94	0.487	00 337	(H/H)	0.785	0.379	06 654	(11/10)	0.81	0.509	00 13E X	0.689	0.39
	and a second			and and and	THE FOR		06 603	(H/H)	0.75	0.402	00 338	(H/H)	0.755	285.0	06 055	(H/H)	0.63	0.434	x 671 90	0.645	55 G
ONGEAMA PROSPECT -	AL TERATION	BIPE	MATCH	HLESS WEST EXI	ENSION		PUG 004	(1/1)	50 TO	1111	065 340	(H/H)	192.0	0.473	00 000	(11/11)	0.66	515.0		-	01.0
							00 00	(H/H)	0.7	0.576	06 341	(H/H)	0.688	0.303	06 658	(H/H)	0.94	0.519			
sample no.	I trace	11	samp	le no.	I trace	17	06 607	(H/H)	0.64	0.395	06 343	(H/H)	0.792	0.442	06 659	(H/H)	16-0	0.433			
							00 609	(H/H)	0.66	0.438	0G 347	(H/H)	0.695	0.494							
2 50	0.995	0.669	WE	201	0.99	0.41	00 610	(H/H)	0.8	0.483	06 348	(H/H)	0.825	0.422		Ore z	one				
06 25	0.988	0.568	He is	202	16.0	0.35					00 349	(#/#)	61/10	0.476 0	DC 660	VE MAY		0,400			
06 30 or 33	0.993	1.61	ME	204	15.0	0.36		7 a.lo	200		06 352	(8/4)	E12-0	0.452	00 661	(E/M)	0.84	0.507			
26 20	1	2.1	- Here	206	16.0	0.47	06 611	(F/H)	0.75	0.509	06 353	(H/H)	0.651	0.352	06 662	(F/H)	0.87	166.0			
06 40	10.994	0.639	HHE	209	16.0	0.37	06 612	(F/H)	0.76	0.463	06 354	(H/H)	111.0	0.643	06 663	(F/H)	0.83	0.486			
06 43	0,988	0.485	MAE	210	1	0.17	06 613	(F/H)	0.75	0.498	00 355	(M/H)	0.721	0.351	06 664	(F/H)	0.76	0.384			
06 46	0.976	0.515	MME	211	0.98	0.47	06 614	(F/H)	0.69	0.426	OC 356	(H/H)	0.772	0.41	00 665	(F/H)	0.77	0.462			
06 49	186.0	0.672	HME	214	0.99	0.49	06 615	(F/H)	0.71	0.43	0C 357	(H/H)	0.693	0.373	06 666	(H/H)	0.78	0.456			
06 52	0.995	1.13	ME	215	0.98	0.34	00 616	(F/M)	0.75	0,465	00 358	(H/H)	0.787	0.51	06 667	(F/H)	0.67	0.359			
0G 55	0.976	0.65	WE	216	0.98	0.5	00 617	(F/H)	0.74	0.498	06 362	(H/H)	102.0	0.46	00 668	(F/W)	0.74	0.469			
88 53	0.982	0.477	A ST	217	0.98	0.39	00 019	(F/H)	65.0	195.0	COC 303	(H/H)	0 273	0 521	06 620	(H/H)	19.0	0.346			
00 52 DC	200.0	675-D		513	0 00	10.0	00 010	(E/M)	0.65	064-D		11/11	e11-0	170.0	00 010	(1/1)	0 73	0000.0			
06 67	0.992	0.558	MAE	221	66.0	0.43	06 621	(E/M)	0.77	0.508	Or6	zone		1	06 672	(F/M)	0.65	0.456			
06 75	0.989	0.358	MIE	224	0.99	1.2	06 622	(F/M)	0.67	0.372					06 673	(H/H)	0.68	0.401			
06 76	0.992	0.349	THE	226	0.99	0.49	06 623	(F/H)	0.8	0.479	06 365	(F/H)	0.748	0.431	06 574	(F/H)	0.87	0.537			
06 77	0.987	0.396	MAE	229	0.99	0.4	06 625	(F/H)	0.8	0.48	06 366	(F/H)	0.749	0.465							
06 78	0.989	0.327	W.	232	66'0	0.47	06 626	(F/M)	61.0	0.35	06 367	(F/H)	99.0	0.408							
08 30	0.986	0.418		234	66.0	0.57	170 00	(M/4)	2.0	105.0	00 300	(1/4)	P. 659	05.0	1						
06 81	696.0	0.3/5		552	1 00 0	n c	00 070	10/11	0 76	135.0	COL DO	(alla)	0 766	19 0							
00 84 07 85	1000	010.0	THE .	123	00.0		630 DA	Tur.	2.5	67.0	1/2 50	(F/M)	0.718	0.489							
DC 87	0.979	0.378		2							06 372	(F/H)	0.73	0.544							
00 88	0.983	0.529									06 373	(F/H)	0.627	0.458							
06 89	0.992	0.542									06 374	(F/H)	0.73	0.506		Tables 16	a-h - A	teration 1	ndex for trace eleme	offer (T trans	and have 1
16 50	566.0	0.401									06 375	(F/H)	0.747	0.489		elements	(1 major)	, and the	T102/2r Index (TZ) f	or unineral	ized and
06 92	0.988	1.09									06 376	(F/H)	0.724	0.395	Č	nineral iz	ed rock a	ypes.			
06 93	0.992	0.519									06 377	(H/H)	167.0	0.532							
96 90	0.97	0.369									06 378	(F/M)	0.769	0.366							
86 30	166.0	0.528									06 3/9	(1/1)	0./50	0.459						ę	
66 90	1	0.205									085 380	(#/4)	0 765	0.406							
101 00	1 003	COC.0									06 387	(E/M)	0.684	0.357							
101 JU	0.985	0.431									06 383	(F/M)	0.762	0.342							
00, 109	0.995	0.643																			
111 50	0.983	0,418																			
06 113	196.0	0.32																			

DNGOMBD PROSPECT - THE GRE-NOSTING SEQUENCE IN DRILL CORES

sample no.	Cu	Pb	Zn	Ni	Co	Ag	Au	Mn	Ba	Cr	V	Ti
	ppm	ppm	ppm	ppm	ppm	ppm	ppb	ppm	ppm	ppm	ppm	ppm
					INCOLUDES DES	*******	BESSAERS		495655564			*******
OTJ 675	17	22	35	252	32	ND	ND	230	10	417	159	5813
OTJ 676	19	21	38	256	41	ND	42	180	10	548	185	6233
OTJ 677	131	16	22	107	21	ND	ND	200	20	116	157	6155
OTJ 678	371	36	56	102	24	ND	ND	790	10	508	45	2340
OTJ 679	25	3	16	31	3	ND	ND	50	10	412	10	8300
OTJ 680	166	12	3	43	3	ND	ND	290	10	157	35	1630
OTJ 681	4	37	9	56	18	ND	ND	745	10	280	22	1910
OTJ 684	11	16	36	372	35	ND	ND	402	43	730	134	4344
OTJ 685	13	30	46	409	31	ND	ND	520	64	975	133	4759
OTJ 686	26	25	37	415	40	ND	ND	78	10	985	144	2527

Table 17 - Trace element geochemistry of the Alpine-type ultramafic rocks from the Otjihase Mine area and Elisenheim.

sample no	As	WO3	Mo	Sn	Zr	Rb	Sr	Bi	Nb	Y
	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm

OTJ 675	ND	ND	ND	ND	57	ND	ND	ND	ND	ND
OTJ 676	ND	45	ND	ND	44	ND	ND	ND	ND	27
OTJ 677	ND	94	ND	ND	116	ND	316	ND	ND	ND
OTJ 678	ND	52	ND	ND	20	19	52	ND	ND	ND
OTJ 679	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND
OTJ 680	ND	63	ND	ND	20	ND	15	ND	ND	ND
OTJ 681	ND	56	ND	ND	27	ND	71	ND	ND	ND
OTJ 684	ND	70	23	ND	45	ND	15	ND	ND	20
OTJ 685	ND	36	ND	ND	44	ND	15	ND	ND	24
OTJ 686	ND	76	ND	ND	29	ND	ND	ND	ND	ND

	MWE 239	MWE 245	0TJ 686
Si02	44.843	55.134	47.809
T102	0.079	0.044	0.374
A1203	4.883	1.903	7.164
Fe203	21.899	10.595	10.488
MnO	0.191	0.262	0.063
MgO	15.399	18.277	26.99
CaO	6.292	11.201	0.15
Na20	0.295	0.494	0.08
K20	0.272	0	0.034
P205	0.087	0.097	0.024
LOI	5.59	1.57	6.67
H20-	0.73	0.26	0.33
	98.885	99.836	100.175

Table 18 - Major element geochemistry of ultramafic rock types.

1 gossanous talc-actinolite schist (Matchless West Extension)

2 gossanous talc-actinolite schist (ultramafic body, Von Francois Ost 60)

Table 19 - Concentrations of selected elements in different ultramafic rock types from Matchless West Extension, Ongombo and Otjihase, and values of the Ex and Um indices.

sample no.	Cr	v	Ni	Co	Ba	Cu	Zn	Pb	Um	Ex
	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm		
MWE 227	12	37	20	19	3799	4100	156	242	88	8143
MWE 233	17	18	29	18	27500	14200	382	162	82	42164
MWE 236	2	272	23	7	77400	2300	201	241	304	80205
MWE 239	7	97	- 26	26	2160	12600	3200	1800	156	18116
MWE 243	13	57	18	6	2695	2800	341	148	94	5930
MWE 245	8	56	8	5	1549	2100	208	143	77	3934
MWE 246	13	38	14	33	1542	4500	2400	49	98	8540
MWE 248	23	375	30	132	2857	35600	8700	207	560	47717
MWE 252	15	78	30	18	65046	11800	6200	203	141	83187
MWE 255	20	196	24	25	3861	3400	3100	285	265	10626
MWE 258	20	30	19	6	2215	2900	370	42	75	5560
OG 359	373	333	136	53	680	7	132	27	895	1714
OTJ 675	417	159	252	32	10	17	35	22	860	922
OTJ 676	548	185	256	41	10	19	38	21	1030	1097
OTJ 677	116	157	107	21	20	131	22	16	401	574
OTJ 678	508	45	102	24	10	371	56	36	679	1116
OTJ 679	412	10	31	3	10	25	16	3	456	507
OTJ 680	157	35	43	3	10	166	3	12	238	417
OTJ 681	280	22	56	18	10	4	9	37	376	399
OTJ 684	730	134	372	35	43	11	36	16	1271	1361
OTJ 685	975	133	409	31	64	13	46	30	1548	1671
OTJ 686	985	144	415	40	10	26	37	25	1584	1657

APPENDIX F

Ongeama Prospect: geochemical profiles along the minor gossan outcrop

Charts 1 to 15

Elements: Cu, Pb, Zn, Co, Ni, Ag, Au, Mn, Ba, Cr, V, Ti, As, Bi, Sn Zr, Rb, Sr

Legend



massive sulphide gossan

trench

magnetite quartzite

. . grid reference 11000E



ferruginous schist

Kuiseb formation

351

concentration of the considered element (Note: ND indicates "below detection limit")

Key for the profiles

 massive gossan
 magnetite quartzite
 ferruginous schist
 country rocks (average content)







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APPENDIX G

Ongeama Prospect: geochemical profiles along the major gossan outcrop

Charts 1 to 16

Elements: Cu, Pb, Zn, Co, Ni, Ag, Au, Mn, Ba, Cr, V, Ti, As, W, Zr, Rb, Bi, Nb

Legend



massive sulphide gossan

semi-massive gossan

(anthophyllite)



farm track



magnetite quartzite

10300E

grid reference



ferruginous schist



magnetic muscovite schist

magnetic chlorite schist



concentration of the considered element (Note: ND indicates "below detection limits)

Key for the profiles

	massive sulphide gossan		т. Т
* * *	semi-massive gossan (anthophyllite)	- (- (-) -	magnetic muscovite schist
	magnetite quartzite	- c - c -	magnetic chlorite schist
	ferruginous schist		
	country rocks (average co	ontent)	



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-

-































APPENDIX H

Ongombo Prospect: geochemical profiles along the gossan outcrop

Charts 1 to 11

Elements: Cu, Pb, Zn, Ni, Co, Ag, Au, Mn, Ba, V, Ti, Sn, W, Zr, Sr, Mo, Bi

Legend

gossan surface outcrop

...

outcrop trace



fault

358

concentration of the considered element (Note: ND indicates "below detection limit")

Key for the profiles

mineralization

_____ <u>x</u> country rocks (average content)



•















CENTRAL SHOOT WESTERN EXTENSION (CENTRAL SHOOT) 1 3611 589 525 486 60 822 10000 E baseline 10000N 9000 E 8000 I 7000 E 261 3,58 479 183 255 182 558 244 642 1002 469 287 $\overline{\mathbf{x}}$ unmineralized schist = 10200 TiO₂ (out of scale)








APPENDIX I

ONGOMBO PROSPECT

Geochemical profiles of the ore-bearing sequence in drillcores

Boreholes no. OGB 151A, OGB 153A, OGB 154A

(in the back pocket)

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APPENDIX J

ONGOMBO PROSPECT

Geochemical profiles of the ore zone in drill cores

Boreholes no. OGB 151A, OGB 153A, OGB 154A Charts 1 to 3

Legend

BS	biotite selvedge zone
SD	sulphides in lenses and disseminations in schists (stringer-type ore)
SM	semi-massive sulphide layers
MQ	(sulphide-bearing) magnetite quartzite
hw sch	quartz-biotite- staurolite-chlorite schist

474 sample number



₹ţ.





APPENDIX L

Plates



Ongeama gossan outcrop: gossan after massive sulphide mineralization (below) overlain by rusty magnetic muscovite schist of the alteration zone.



Plate 2

Ongeama gossan outcrop: close-up of banded magnetite quartzite showing small-scale tight fold-like structures.



Ongeama gossan outcrop: gossan after massive sulphide mineralization (below) overlain by rusty magnetic muscovite schist of the alteration zone.



Plate 2

Ongeama gossan outcrop: close-up of banded magnetite quartzite showing small-scale tight fold-like structures.



Ongeama gossan outcrop, alteration zone: mottled magnetic stauroliteand garnet-bearing chlorite schist.



Plate 4

Ongombo Prospect: outcrop of magnetite quartzite in the sandy bed of the White Nossob River.



Ongeama gossan: coarse boxwork structure after pyrite. Note a small patch with fine-grained ladder-like boxwork after chalcopyrite towards the bottom-right corner. Acicular traces are probably anthophyllite crystals. White-grey material is Fe-hydroxide.

(Polished section OG 63; // Nicols; field of view = 2.5 mm).



Plate 6

Ongeama gossan: boxwork structure after shrinkage cracks, typical of bird's eye weathering in pyrrhotite. The light grey material is Fehydroxide.

(Polished section OG 38; // Nicols; field of view = 2.5 mm).



Plate 7

Ongombo gossan: boxwork structures after anthophyllite aggregates. The light grey material is Fe-hydroxide. (Polished section OG 196; // Nicols; field of view = 2.5 mm).



Plate 8

Ongeama gossan: radiating anthophyllite aggregates impregnated with Fe-hydroxides. White areas indicate voids. (Thin section OG 63; // Nicols; field of view = 4 mm).



Ongeama gossan: phlogopite (?)-bearing ferruginous schist (brownish) containing gahnite (?) (greenish transparent, fractured) and magnetite (opaque).

(Thin section OG 64; + Nicols; field of view = 4 mm).

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Plate 10

Ongeama deposit, alteration pipe, magnetic quartz-chlorite schist facies: aggregate of garnet (dark brown, containing light brown staurolite) and chloritoid (greenish-grey) in chlorite-bearing quartzrich groundmass. The opaque mineral is magnetite.

(Thin section OG 77B; // Nicols; field of view = 4 mm).



Plate 11

Ongeama deposit, alteration pipe, magnetic quartz-chlorite schist facies: coarse-grained skeletal garnet porphyroblast in chloritebearing quartzose groundmass.

(Thin section OG >7B; // Nicols; field of view = 4 mm).



Plate 12

Ongeama deposit, alteration pipe magnetic muscovite facies: Fehydroxide-stained quartz-muscovite schist, with disseminated magnetite (opaque). Sporadic staurolite (brownish, high relief, cracked) and chlorite (greenish, lamellar).

(Thin section OG 62; // Nicols; field of view = 4 mm).



Matchless West Extension deposit: particular of a talc-quartz-rich patch of the massive sulphide gossan. Talc flakes (pink to green) randomly grow on the Fe-hydroxide-impregnated quartz-rich groundmass. (Thin section MWE 207; + Nicols; field of view = 4 mm).



Plate 14

Matchless West Extension deposit: quartz-muscovite schist from the axial zone of the alteration pipe. Fe-hydroxide-rimmed cavities after euhedral pyrite crystals are disseminated. Fe-hydroxides impregnate the groundmass.

(Thin section MWE 205; // Nicols; field of view = 4 mm).



Plate 15

Matchless West Extension deposit, upper magnetite quartzite layer: thin magnetite strings and anhendral baryte blebs (colourless, high relief) in the sulphide-devoid quartz-rich groundmass. (Thin section MWE 253; // Nicols; field of view = 4 mm).



Plate 16

Matchless West Extension deposit, lower magnetite quartzite layer: typical banding induced by magnetite-rich laminae. Irregular sulphide and malachite patches and a weathered, anhedral amphibole grain are also shown.

(Thin section MWE 241A; // Nicols; field of view = 4 mm).



Matchless West Extension deposit, ultramafic exhalites: talcphlogopite-carbonate schist. Talc and phlogopite lamellae grow on a quartzose groundmass. Carbonate porphyroblasts are repleced by Fehydroxides.

(Thin section MWE 246A; // Nicols; field of view = 4 mm).



Plate 18

Matchless West Extension deposit: same rock type as in Plate 17. (Thin section MWE 246A; + Nicols; field of view = 4 mm).



Plate 19

Matchless West Extension deposit, ultramafic exhalites: assemblage of actinolite (yellowish), quartz (white to grey) and baryte (grey, in centre).

(Thin section MWE 236; + Nicols; field of view = 4 mm).



Plate 20

Matchless west Extension deposit, ultramafic exhalites: actinolite (yellowish) and baryte (grey, mottled) in gossanous quartzite. (Thin section MWE 245A; + Nicols; field of view = 4 mm).



Matchless West extension deposit, marble-like rock: anhedral, sievetextured carbonate crystals replacing the muscovite-bearing quartzitic-groundmass.

(Thin section MWE 259; + Nicols; field of view = 4 mm).



Plate 22

Von Francois Ost 60 farm area, massive talc-chlorite-actinolite schist: subhedral actinolite crystals (yellowish) and talc lamellae (pink to green) included in a fine-grained chlorite-rich groundmass. (Thin section OTJ 685; + Nicols; field of view = 4 mm).



Ongombo deposit, borehole no. OGB 153A: staurolite-bearing quartzbiotite-muscovite-chlorite schist in the hangingwall next to the ore zone. Note the brown colour of the biotite lamellae. (Thin section OG 716; // Nicols; field of view = 4 mm).



Plate 24

Ongombo deposit, borehole no. OGB 153A: biotite selvedge zone showing the coarse biotite groundmass, subhedral staurolite grains, coarse apatite crystals and part of pyrite "augen". Note the reddish-brown tinge of the biotite lamellae.

(Thin section OG 718; // Nicols; field of view = 4 mm).



Ongombo deposit, borehole no. OGB 153A: plagioclase-rich biotite schist facies within the ore zone.

(Thin section OG 735; + Nicols; field of view = 4 mm).



Plate 26

Ongombo deposit, borehole no. OGB 153A, sulphide-bearing magnetite quartzite: embayed intergrowth between recrystallized pyrite (white) and magnetite (grey). Chalcopyrite (greeny yellow) occurs as interstitial patches and inclusions within pyrite and magnetite. (Polished section OG 747; // Nicols; field of view = 2.5 mm).



Ongombo deposit, borehole no. OGB 153A: plagioclase-rich biotite schist facies within the ore zone.

(Thin section OG 735; + Nicols; field of view = 4 mm).



Plate 26

Ongombo deposit, borehole no. OGB 153A, sulphide-bearing magnetite quartzite: embayed intergrowth between recrystallized pyrite (white) and magnetite (grey). Chalcopyrite (greeny yellow) occurs as interstitial patches and inclusions within pyrite and magnetite. (Polished section OG 747; // Nicols; field of view = 2.5 mm).