

## Geochemical characterization and palynological studies of some Agbada Formation deposits of the Niger Delta basin: implications for paleodepositional environments

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**Abstract:** Forty-two ditch cutting samples of the KR-1 offshore well from depths of 9660 ft to 10,920 ft composited at 90-ft intervals were subjected to sedimentological, micropaleontological, and geochemical analyses using standard procedures and the laser ablation-induced coupled plasma mass spectrometry technique, respectively. Sedimentological analysis revealed the presence of glauconites and the rare occurrence of framboidal pyrites, indicative of deposition in a slightly anoxic marine environment. Palynomorph percentage distribution shows that there are more terrestrially derived miospores (dominated by *Zonocostites ramonae* (*Rhizophora* spp.), *Psilatricolporites crassus* (*Tabernaemontana crassa*), *Acrotichum aureum*, and *Laevigatosporites* sp.) than marine phytoplanktons. Rare occurrence of *Globoquadrina venezuelana*, *Globigerinoides promordius*, and *Globigerina* sp. denotes an Early Miocene age and proximal shelf. These indicate that the main environment of deposition in the KR-1 well is coastal to marginal marine consisting of coastal deltaic-inner neritic, made up of tidal channel and shoreface deposits. Geochemical results show that the average concentrations of considered rare earth elements are less than their concentrations in world average shale. Trace metal ratios (such as Th/Cr, Cr/Th, Th/Co, and Cr/Ni) suggest that the investigated sediments were derived from felsic source rocks. Rare earth element patterns (such as La/Yb, Gd/Yb, La/Sm, and Eu/Eu) and Th data established the felsic composition of the source rocks. Ratios of U/Th, Ni/Co, Cu/Zn, and V/Sc suggest a well-oxygenated bottom water condition. Estimated europium and cerium anomalies of the studied samples suggest an oxidizing environment of deposition. Nonetheless, the ratios of V/Cr suggest a range of environmental conditions. Moreover, ratios of V/(V+Ni) suggest the rare occurrence of suboxic to anoxic environments of deposition.

**Key words:** Sedimentology, palynomorphs, traces elements, rare earth element, environment of deposition, Niger Delta, Nigeria

### 1. Introduction

The Niger Delta basin is one of the sedimentary basins in Nigeria (Figure 1). It is an important basin because it contains large hydrocarbon resources. This makes Nigeria the most prolific oil producer in Sub-Saharan Africa, ranking as the third largest producer of crude oil in Africa and the tenth largest in the world. Nigeria's economy is predominantly dependent on its oil sector; oil supplies 95% of Nigeria's foreign exchange earnings and 80% of its budgetary revenues (Olayiwola, 1987; Adenugba and Dipo, 2013). This petroliferous nature has made the basin, for many years, the subject of continuous, consistent, and extensive geologic investigations both for academic and economic purposes (Adebayo, 2011). Intensive exploration and exploitation of hydrocarbon in the basin has been ongoing since the early 1960s due to the discovery of oil in commercial quantity in the Oloibiri-1 well in 1956 (Nwajide and Reijers, 1996). Biostratigraphy

played an important role in the exploration of oil and gas in the Niger Delta basin. Microfossils were employed among other things to reconstruct the paleoenvironment of the studied sections. This is important because different depositional settings imply different reservoir qualities in terms of architecture, connectivity, heterogeneity, and porosity-permeability characteristics (Simmons et al., 1999).

Trace element abundances in sedimentary rocks have added significantly to our understanding of crustal evolution with rare earth element (REE) patterns and Th being particularly useful in determining provenance (Ganai and Rashid, 2015). The geochemical behavior of trace elements in modern organic-rich, fine-grained sedimentary rocks (i.e. shales) and anoxic basins has often been documented to determine paleoenvironmental conditions of deposition (Brumsack, 1989; Calvert and Pedersen, 1993; Warning and Brumsack, 2000; Algeo

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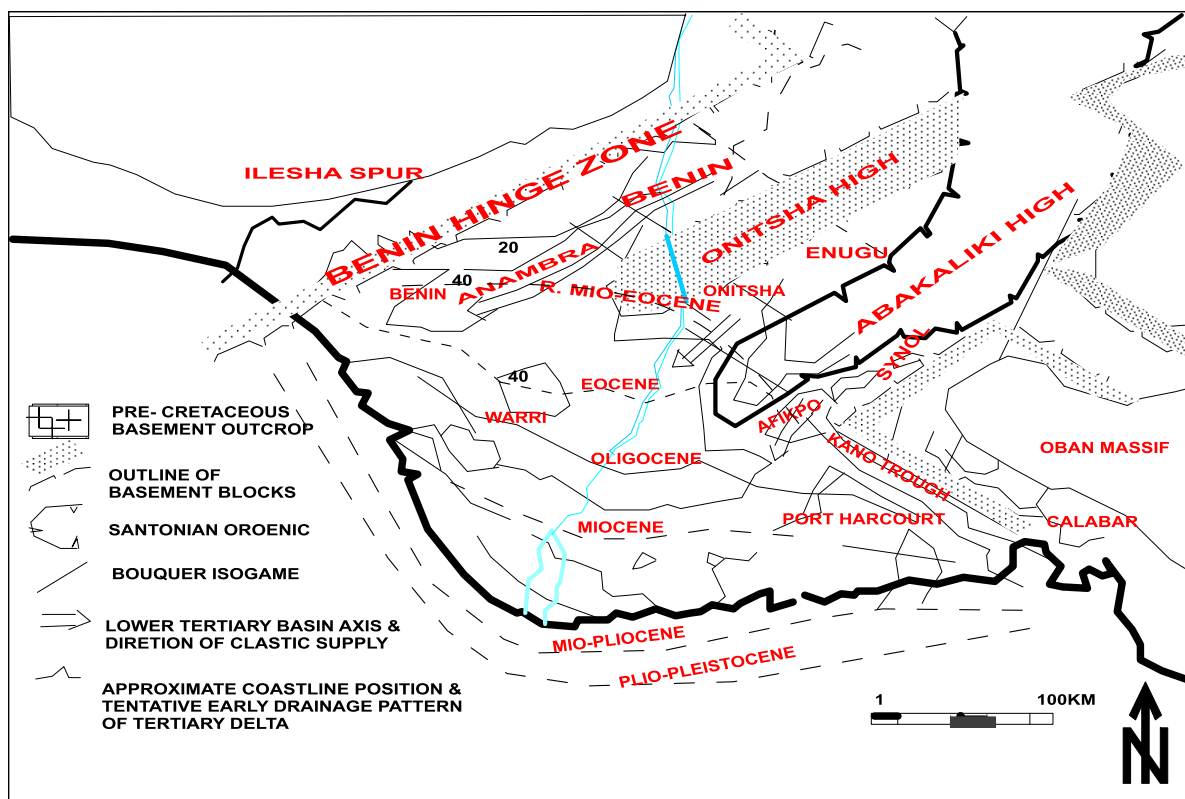


Figure 1. Geological map of the Niger Delta (Weber and Daukoru, 1975).

and Maynard, 2004). Redox-sensitive trace element (TE) concentrations or ratios are among the main extensively used indicators of redox conditions in modern and ancient sedimentary deposits (e.g., Calvert and Pedersen, 1993; Jones and Manning, 1994; Crusius et al., 1996; Dean et al., 1997, 1999; Yarincik et al., 2000; Morford et al., 2001; Paillet et al., 2002; Algeo and Maynard, 2004). Enrichments of redox-sensitive elements replicate the depositional environment of ancient organic carbon-rich sediments and sedimentary rocks as well and can consequently be used to reveal the likely paleodepositional conditions leading to their formation (Brumsack, 1980, 1986; Hatch and Leventhal, 1992; Piper, 1994). The degree of enrichment/depletion is usually based on the element/Al ratio in a sample, calculated relative to the respective element/Al ratio of a common standard material, e.g., average marine shale (Turekian and Wedepohl, 1961). The purpose of this paper is to interpret the paleoenvironmental changes during the deposition of the sediments in the studied section of the Niger Delta basin. To achieve the objective, a multidisciplinary approach combining sedimentological features and palynological and geochemical analyses was employed.

## 2. The geologic setting of the basin

The present-day Niger Delta Complex is situated on the continental margin of the Gulf of Guinea in the southern part of Nigeria. It lies between longitudes 4 °E and 8.8 °E and latitudes 3 °N and 6 °N (Figure 1). The onshore portion of the basin is delineated by the geology of southern Nigeria and southwestern Cameroon. It is bounded in the north by outcrops of the Anambra Basin and the Abakaliki Anticlinorium, and delimited in the west by the Benin Flank, a northeast-southwest trending hinge line south of the West African basement massif. The Calabar Flank, a hinge line bordering the Oban massif, defines the northeastern boundary. The offshore boundary of the basin is defined by the Cameroon volcanic line to the east and the eastern boundary of the Dahomey Basin (the easternmost West African transform-fault passive margin) to the west. The evolution of the delta is controlled by pre- and synsedimentary tectonics as described by Evamy et al. (1978), Ejedawe (1981), Knox and Omatsola (1987), and Stacher (1995). It is a large arcuate delta covering an area of about 300,000 km<sup>2</sup> (Kulke, 1995), with a sediment volume of 500,000 km<sup>3</sup> (Hospers, 1965) and a sedimentary thickness of over 10 km in the basin depocenter (Kaplan et al., 1994).

The evolution of the basin has been linked to that of a larger sedimentary complex called the Benue-Abakaliki Trough. The trough, a NE-SW trending aborted rift basin with folded sedimentary fill, runs obliquely across Nigeria (Figure 1). The Niger Delta basin is actually the youngest and the southernmost subbasin in the trough (Murat, 1972; Reijers et al., 1997).

The evolution of the trough, which began in the Cretaceous, during the opening of the South Atlantic, led to the separation of the African and South American plates. The tectonic framework of the continental margin along the western coast of Africa is controlled by Cretaceous fracture zones expressed as trenches and ridges in the deep Atlantic. The fracture zone ridges subdivided the margin into individual basins and, in Nigeria, form the boundary faults of the Cretaceous Benue-Abakaliki Trough, which cuts far into the West African Shield. The rifting greatly diminished in the Late Cretaceous in the Niger Delta region (Ako et al., 2004). A well section through the Niger Delta basin generally displays three vertical lithostratigraphic subdivisions, namely a prodelta lithofacies, a delta front lithofacies, and upper delta top facies (Nwajide and Reijers, 1996). These lithostratigraphic units correspond respectively to the Akata Formation (Paleocene-Recent), Agbada Formation (Eocene-Recent), and Benin Formation (Oligocene-Recent) (Short and Stauble, 1967).

### 3. Materials and methods

Forty-two ditch cutting samples of the KR-1 offshore well (Figure 2) were taken from depths of 9660 to 10,920 ft at 90-ft interval (Figure 3). These were processed and analyzed for sedimentological, palynological, micropaleontological, and geochemical studies.

#### 3.1. Sedimentological analysis

The samples were subjected to sedimentological analysis using visual inspection and a binocular microscope. Physical characteristics such as color, texture, hardness, fissility, and rock types were noted. Dilute HCl (10%) was added to identify the calcareous samples. Fossil contents, presence of accessory minerals, and postdepositional effects such as ferruginization were determined.

#### 3.2. Palynological preparation

Ten grams of each dry sample were crushed into small fractions between 0.25 mm and 2.5 mm. Standard palynological processing procedures were employed (Faegri and Iversen, 1989; Wood et al., 1996). These included the digestion of the mineral matrix using dilute HCl for carbonates and concentrated HF for silicates. Removal of the fluoride gel (formed during the HF treatment) was done using hot concentrated HCl and wet sieving the residue using a 10- $\mu$ m polypropylene Estal Mono sieve. The residues were oxidized and inorganic

materials were separated from the organic ones using  $ZnCl_2$  of specific gravity 2.0. Slides were mounted using Norland adhesive mounting medium and dried under UV light. One slide per sample was analyzed under the optical microscope and the photomicrographs of well-preserved palynomorph specimens were taken using an Olympus CH30 transmitted light microscope (Model CH30RF200) with an attached camera. Palynomorph identifications were done using the works of Germeraad et al. (1968) and Evamy et al. (1978) (i.e. Shell Oil Company Scheme, 1978). The data were plotted using *StrataBugs* software at 1:5000 scale with depth on the y-axis and the identified taxa on the x-axis.

#### 3.3. Foraminiferal preparation

Twenty-five grams of each sample was processed for their foraminiferal content using the standard preparation techniques. The weighed samples were soaked in kerosene and left overnight to disaggregate, followed by soaking in detergent solution overnight. The disaggregated samples were then washed-sieved under running tap water over a 63- $\mu$ m mesh sieve. The washed residues were then dried over a hot electric plate and sieved (when cooled) into three main size fractions, namely coarse, medium, and fine (250-, 150-, and 63- $\mu$ m meshes). Each fraction was examined under a binocular microscope. All the foraminifera, ostracodes, shell fragments, and other microfossils observed were picked with the aid of a picking needle and counted. Foraminifera identification was made to genus and species levels where possible using the taxonomic scheme of Loeblich and Tappan (1964) and other relevant foraminiferal literature such as the works of Fayose (1970), Postuma (1971), Petters (1979a, 1979b, 1982), Murray (1991), and Okosun and Liebau (1999).

#### 3.4. XRF and LA-ICPMS analyses

The pulverized ditch cutting samples were analyzed with X-ray fluorescence (XRF) and laser ablation-induced coupled plasma mass spectrometry (LA-ICPMS) techniques. The elemental data for this work were acquired using XRF and LA-ICPMS analyses.

The analytical procedures were as follows:

Pulverized ditch cutting samples were analyzed for major elements using an Axios instrument (PANalytical) with a 2.4-kW Rh X-ray tube. The same set of samples was further analyzed for trace elements using LA-ICPMS instrumental analysis. LA-ICPMS is a powerful and sensitive analytical technique for multielement analysis. The laser was used to vaporize the surface of the solid sample, while the vapor and any particles were then transported by the carrier gas flow to the ICP-MS. The detailed procedures for sample preparation for both analytical techniques are reported below.

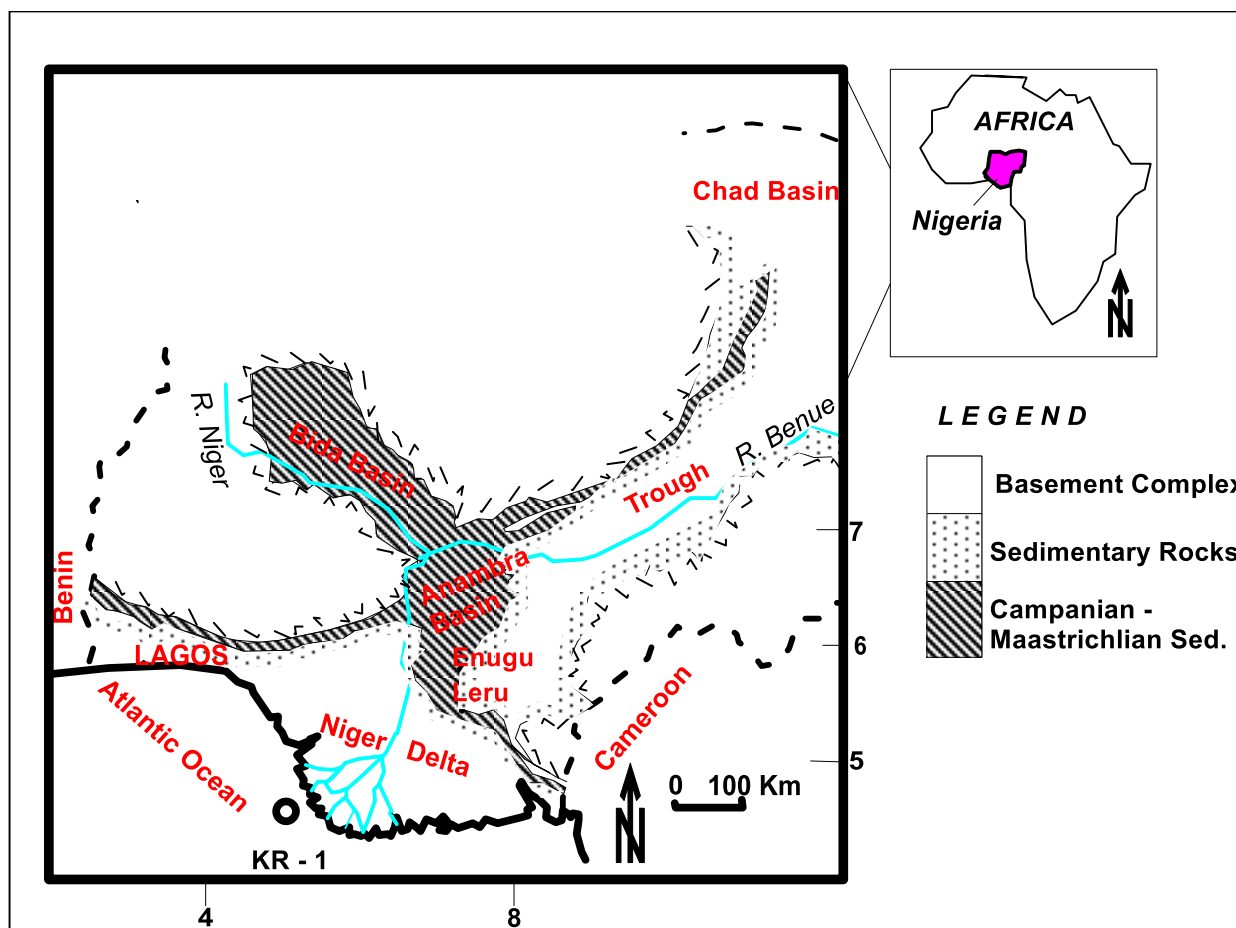


Figure 2. Simplified geologic map of Nigeria and location of KR-1 well (Adebayo et al., 2015b).

### 3.4.1. Fusion bead method for major element analysis

- Weigh  $1.0000 \pm 0.0009$  g of milled sample.
- Place in oven at  $110\text{ }^{\circ}\text{C}$  for 1 h to determine  $\text{H}_2\text{O}+$ .
- Place in oven at  $1000\text{ }^{\circ}\text{C}$  for 1 h to determine LOI.
- Add  $10.0000 \pm 0.0009$  g of Claissé flux and fuse in M4 Claisséfluxer for 23 min.
- Add 0.2 g of  $\text{NaCO}_3$  to the mix and preoxidize the sample+flux+ $\text{NaCO}_3$  at  $700\text{ }^{\circ}\text{C}$  before fusion.
- Flux type: Ultrapure Fused Anhydrous Li-Tetraborate-Li-Metaborate flux (66.67%  $\text{Li}_2\text{B}_4\text{O}_7$  + 32.83%  $\text{LiBO}_2$ ) and releasing agent Li-iodide (0.5% LiI).

### 3.4.2. Pressed pellet method for trace element analysis

- Weigh  $8 \pm 0.05$  g of milled powder.
- Mix thoroughly with 3 drops of Mowiol wax binder.
- Press pellet with pill press to pressure of 15 t.
- Dry in oven at  $100\text{ }^{\circ}\text{C}$  for 30 min. before analyzing.

These analytical methods yielded data for 11 major elements, reported as oxide percent by weight [ $\text{SiO}_2$ ,  $\text{TiO}_2$ ,  $\text{Al}_2\text{O}_3$ ,  $\text{Fe}_2\text{O}_3$ ,  $\text{MgO}$ ,  $\text{MnO}$ ,  $\text{CaO}$ ,  $\text{Na}_2\text{O}$ ,  $\text{K}_2\text{O}$ ,  $\text{Cr}_2\text{O}_3$ , and

$\text{P}_2\text{O}_5$ ] and 21 trace elements [Ni, Cu, Zn, Ga, Rb, Sr, Y, Zr, Nb, Co, V, Pb, Th, U, Ti, Cr, Ba, La, Ce, Nd, and P] reported as mg/kg (ppm).

## 4. Results and discussion

### 4.1. Sedimentological analysis

Lithologically, the sequence is characterized by the alternation of shale and sandy shale facies (Figure 3). The shales are light gray, fissile, effervescent and slightly ferruginized while the sandy shales are light gray and ferruginous. These sediments contain muscovite flakes. There are few to common occurrences of glauconite while pyrite and shell fragments are rare to few. Quartz grains within the sediments vary from fine to medium, subangular to well-rounded and moderately sorted.

### 4.2. Palynological assemblage

Palynomorph preservation in the analyzed sediments is fairly good with high concentration and diversity (see Figures 4 and 5). All the samples yielded common to

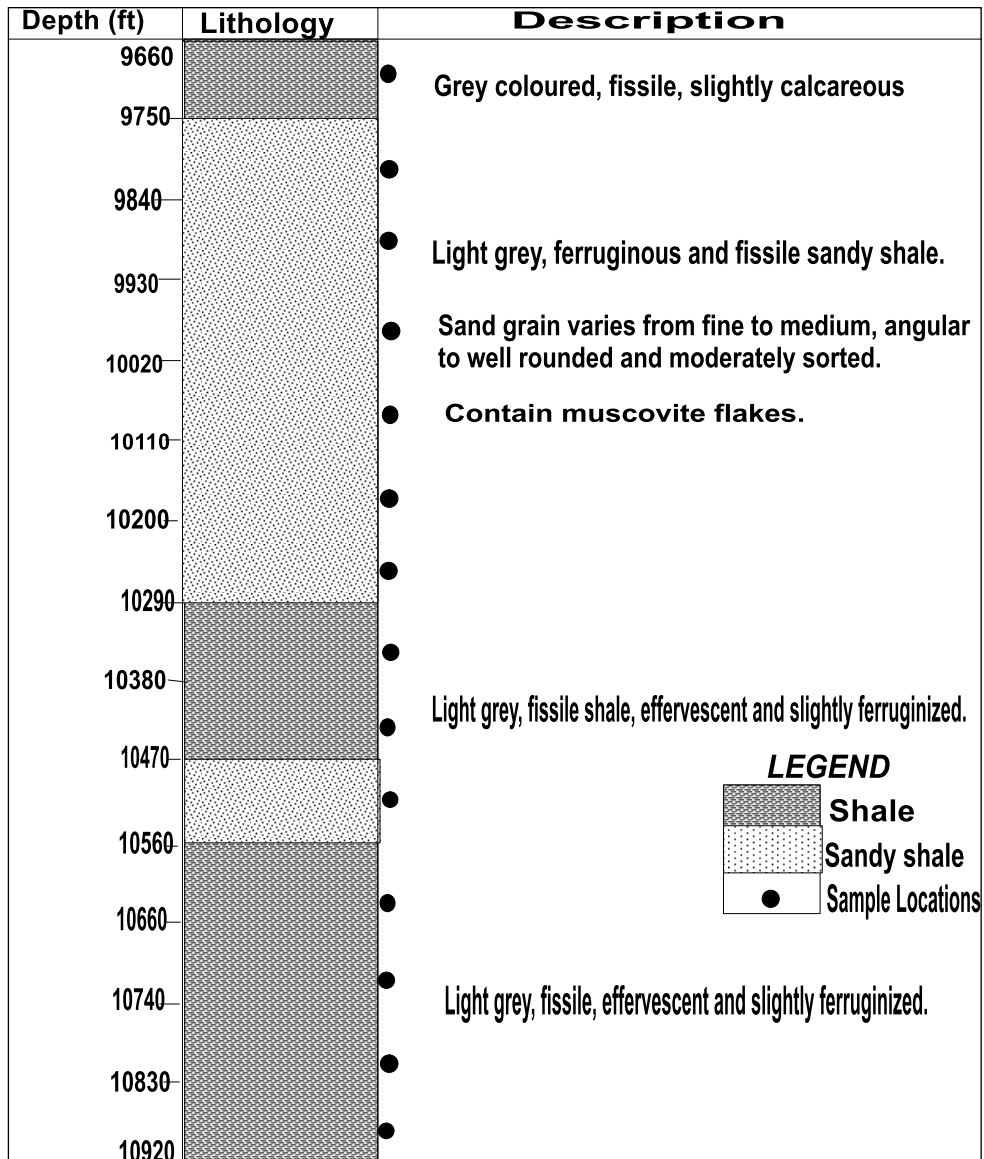


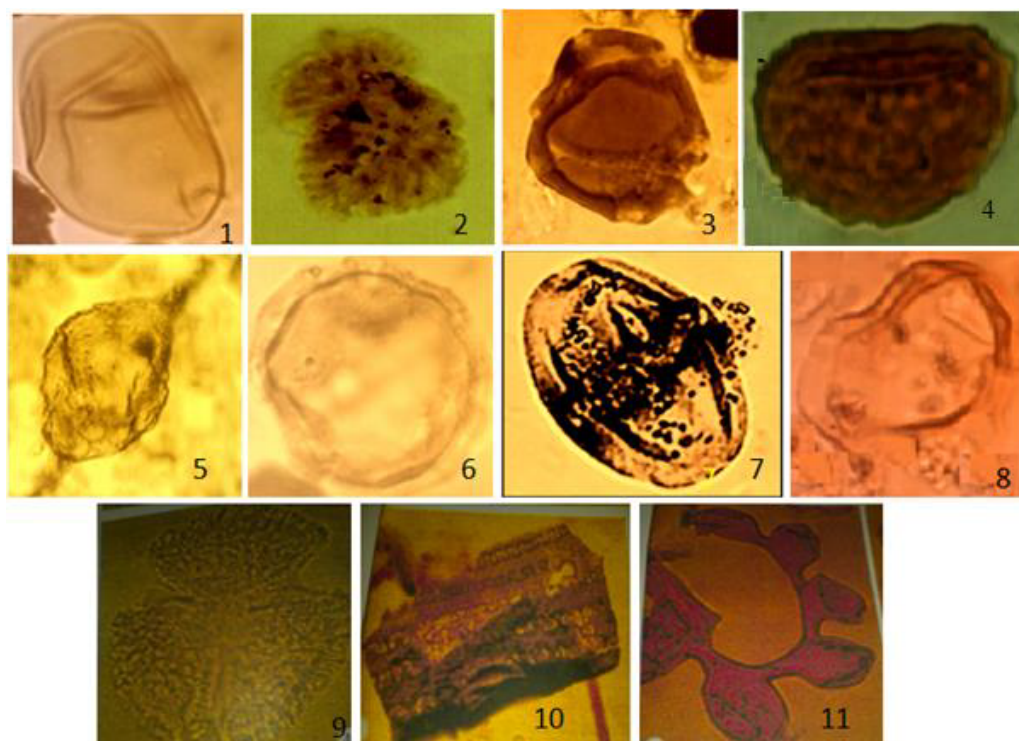
Figure 3. The lithology of the KR-1 well (after Adebayo et al., 2015b).

abundant assemblages that range from moderate to well preserved. Dinoflagellate cysts are sporadically present and range in abundance from very rare to few and do not occur in all the samples.

There are 256 pollen grains, 220 spores, 231 *Botryococcus* and *Pediastrum*, 9 dinoflagellate cysts, and 3 microforaminiferal wall linings, making a total of 719 recovered palynomorphs. The assemblage is dominated by angiospermous pollen with an equally significant occurrence of pteridophyte spores. The angiosperms consist mainly of *Tricolporites*, *Tetraporites*, and *Monoporites* while *Laevigatosporites*, *Verrucatosporites*, and *Polypodiaceoisporites* are the dominant pteridophyte spores (Figures 4 and 5). The biostratigraphically

important palynomorphs recovered from the well are *Zonocostites ramonae* (*Rhisophora* sp.), *Psilatricolporites crassus* (*Tabermaemontana* sp.), *Pachydermites diederixi* (*Symphonia globulifera*), *Retitricolporites irregularis* (*Amanoa* sp.), *Praedapollis africanus*, and *Verrucatosporites usmensis* (*Polypodium* sp.) (Figures 4 and 5). The palynomorph assemblage as a whole shows strong similarities with those previously identified in the San Jorge Gulf Basin, southern Patagonia, Argentina (Palamarczuk and Barreda, 1998), and especially those from the Mazarredo Subbasin (Barreda and Palamarczuk, 2000), dated as Early Miocene and Latest Oligocene-Early Miocene, respectively. The KR-1 well assemblage is also closely comparable to the Early Miocene interval





**Figure 5.** Plates of recovered palynomorphs from the investigated intervals from the KR-1 well (1000×). **1.** *Laevigatosporites* sp.; **2.** *Botryococcus braunii* Kützing, 1849; **3.** *Pachydermitesdiederixi* Germeraad, Hopping & Muller, 1968; **4.** *Verrucatosporites* sp.; **5.** *Palaeocystodinium* sp.; **6.** *Monoporites annulatus* van der Hammen, 1954; **7.** Sapotaceae; **8.** *Psilatricolporites crassus* van der Hammen & Wijmstra 1964; **9.** *Retitricolporites irregularis* van der Hammen & Wijmstra, 1964; **10.** charred Gramineae; **11.** microforaminiferal wall lining.

important marker species such as *Zonocostites ramonae* (mangrove pollen), *Monoporites annulatus* (Poaceae pollen suggesting open vegetation found in coastal Savannah), *Magnastriatites howardi* (a small aquatic fern of alluvial plain and coastal swamps), *Pachydermites diderixi* (an angiosperm of coastal swamps), foraminiferal wall linings, and dinocysts are recovered. Lithologically, glauconite and pyrite are the most important accessory minerals in the studied well that can be used for environmental deductions. Glauconite forms only as an authigenic mineral during the early stage of the diagenesis of marine sediments. It is extremely susceptible to subaerial weathering and is not known as a reworked second cycle detrital mineral (Selley, 1976). The presence of glauconite in the sandy shales therefore indicates a marine origin. On the other hand, rare occurrence of pyrite in the shale bodies probably suggests a reducing condition during deposition.

The studied sequence can be categorized into three sections based on significant changes in the occurrence of the recovered taxa (Figure 4). The lowermost section, which lies between depths of 10,920 and 10,560 ft, constituted a paleoecological zone. It is characterized by the appreciable occurrence of organic wall microplanktons such as foraminiferal wall linings and

dinocysts (*Palaeocystodinium* spp.), uphole decrease in the population of *Monoporites annulatus*, rare occurrence of *Botryococcus braunii*, and the paucity of fresh water forms represented by *Pediastrum* (Figures 4 and 5). This section is assigned to a marginal marine environment (Sarjeant, 1974; Durugbo, 2013). The depth between 10,560 and 9930 ft belongs to a continental-mangrove environment based on the dominance of terrestrially derived taxa (*Psilatricolporites crassus* and *Pachydermites diderixi*), the acme of *Zonocostites ramonae*, and the absence or rarity of microplanktons. The topmost section, which lies between 9930 and 9750 ft, is a mixed environment that ranges from back-mangrove to brackish water swamp to marshes. Though this section of the well is dominated by *Botryococcus braunii* and *Zonocostites ramonae*, the significant presence of *Psilatricolporites crassus* and *Acrotichum aureum* (similar to *Deltoidospora adriennis*) (Figures 4 and 5) and the occurrence of microplanktons enable the suggestion of back-mangrove-brackish water swamp-marshes (Tomlinson, 1986; Thanikaimoni, 1987).

#### 4.4. Trace element/Al ratios and enrichments

The enrichment factor (EF) for an individual element is equal to  $(\text{element}/\text{Al})_{\text{sample}} / (\text{element}/\text{Al})_{\text{shale}}$ , where the

ratio in the numerator is that for the shale in question and the ratio in the denominator is that for a “typical” shale (using data from Wedepohl, 1971, 1991). Any relative enrichment is then expressed by  $EF > 1$ , whereas depletion elements have  $EF < 1$ . This approach has been used by various authors to evaluate trace-element enrichments in modern and ancient sediments (e.g., Calvert and Pedersen, 1993; Arnaboldi and Meyers, 2003; Rimmer, 2004; Brumsack, 2006). Generally, comparisons of V/Al ratios in the Agbada Formation samples with world average shale (Wedepohl, 1971) show high enrichment factors ( $EF_V = 5.74-1.15$ ) at some depth intervals such as 9660–9750 ft, 9750–9840 ft, 9840–9930 ft, and 9930–10,020 ft (Table 1). In contrast, other investigated intervals were marked by low enrichment factors ( $EF_V = 0.40-0.05$ ). Compared with average shale, Mo/Al ratios in the studied Agbada Formation samples show high enrichment factors ( $EF_{Mo} = 115.45-5.56$ ) in all the investigated depth intervals. The observed variability in Mo/Al and V/Al ratios in the studied Agbada Formation samples are indicative of a mixed environment of deposition (i.e. paralic setting). Compared with world average shale, Ni/Al ratios in the Agbada Formation samples show high enrichment factors ( $EF_{Ni} = 5.21-1.21$ ) at 9660–9750 ft, 9750–9840 ft, 9840–9930 ft, and 9930–10,020 ft depth intervals (Table 1). Alternatively, other investigated depth intervals show low enrichment factors ( $EF_{Ni} = 0.81-0.27$ ). In comparison with the world average shale, Co/Al ratios in the studied samples show high enrichment factors ( $EF_{Co} = 14.56-1.64$ ). Variability in the enrichment of Ni/Al and Co/Al ratios in the Agbada Formation samples indicate a mixed environment of deposition. U/Al ratios compared with average shale show high enrichment factors ( $EF_U = 5.28-1.11$ ) in samples taken at depth intervals such as 9660–9750 ft, 9750–9840 ft, 9840–9930 ft, 9930–10,020 ft, 10,650–10,740 ft, 10,740–10,830 ft, and 10,830–10,920 ft (Table 1). Conversely, other investigated depth intervals show low enrichment factors ( $EF_U = 0.82-0.20$ ). Compared with average shale, Cr/Al ratios show high enrichment factors ( $EF_{Cr} = 12.16-0.52$ ), with the exception of the sample taken at the depth interval of 10,020–10,110 ft. Lower U/Al and Cr/Al ratios imply oxic bottom water conditions during deposition. Compared with world average shale, Sr/Al ratios in the Agbada Formation samples show high enrichment factors ( $EF_{Sr} = 4.60-1.01$ ) at 10,650–10,740 ft, 10,740–10,830 ft, 10,830–10,920 ft, 9840–9930 ft, and 9930–10,020 ft depth intervals. Conversely, low enrichment factors ( $EF_{Sr} = 0.99-0.19$ ) were observed in other investigated depth intervals. Ba/Al ratios in the studied samples compared with world average shale show high enrichment factors ( $EF_{Ba} = 54.71-1.31$ ) in all the investigated depth intervals. Furthermore, a relatively high enrichment of Ba/Al and Sr/Al ratios suggest well-oxygenated bottom water conditions during

deposition. The Cu/Al ratios in Agbada Formation samples compared with world average shale show high enrichment factors ( $EF_{Cu} = 6.97-1.64$ ) at 10,380–10,470 ft, 10,650–10,740 ft, 10,740–10,380 ft, 10,380–10,920 ft, 9660–9750 ft, 9750–9840 ft, 9840–9930 ft, and 9930–10,020 ft depth intervals. Other investigated depth intervals show low enrichment factors ( $EF_{Cu} = 0.95-0.38$ ). Zn/Al ratios in the studied samples compared with world average shale show high enrichment factors ( $EF_{Zn} = 12.92-0.79$ ) with the exception of the sample taken at the 10020–10110 ft depth interval. Compared with world average shale, Pb/Al ratios for all samples show high enrichment factors ( $EF_{Pb} = 18.35-0.80$ ), with the exception of samples taken at 10,020–10,110 ft and 10,200–10,290 ft depth intervals.

Going by the world average shale standard, Rb/Al ratios show evidence of low enrichment factors ( $EF_{Rb} = 3.34-0.10$ ) with the exception of the sample taken at the 9660–9750 ft depth interval. Similarly, compared with world average shale, the Y/Al ratios in Agbada Formation samples show low enrichment factors ( $EF_Y = 4.38-1.04$ ). Alternatively, low enrichment factors ( $EF_Y = 0.98-0.09$ ) were obtained in samples taken at 9660–9750 ft, 9840–9930 ft, 9930–10,020 ft, 10,650–10,740 ft, and 10,740–10,830 ft depth intervals. Zr/Al ratios in Agbada Formation samples compared with world average shale show high enrichment factors ( $EF_{Zr} = 11.42-0.92$ ), with the exception of the sample taken at the 10,020–10,110 ft depth interval. The studied Agbada Formation samples exhibit different degrees of trace-element enrichment, with the approximate order of enrichment relative to world average shale as follows: Mo > Ba > Pb > Cr > Co > Zn > Zr > Cu > V > U > Ni > Sr > Rb.

#### 4.5. Provenance and paleoredox conditions

Armstrong-Altrin et al. (2004) revealed that low contents of Cr imply a felsic provenance, and high levels of Cr and Ni are essentially found in sediments derived from ultramafic rocks. Nickel concentrations are lower in the Agbada Formation sediments compared with world average shale (WSA) (Table 2), but chromium shows higher contents. Accordingly, the low Cr/Ni ratios in Agbada Formation samples are between 1.32 and 10.93. This indicates that felsic components were the major components among the basement complex source rocks. Some authors showed that ratios such as La/Sc, Th/Sc, Th/Co, and Th/Cr are significantly different in felsic and basic rocks and may possibly allow constraints on the average provenance composition (Wronkiewicz and Condie, 1990; Cullers, 1994, 1995, 2000; Cox et al., 1995; Cullers and Podkovyrov, 2000; Nagarajan et al., 2007). The ratios of Th/Cr (~0.03–0.09; average = ~0.05), Cr/Th (~10.70–30.64; average = ~20.37), Th/Co (~0.01–0.48; average = ~0.25), and Cr/Ni (~1.32–10.93; average = ~5.13) (Table 3) imply that the Agbada Formation sediments recovered from the KR-1 well were derived from felsic source



**Table 1.** Trace element ratios and enrichments in the Agbada Formation Sediments compared to world average shale (WSA) (Wedepohl, 1971).

Element	WSA	9660– 9750 ft	9750– 9840 ft	9840– 9930 ft	9930– 10,020 ft	10,020– 10,110 ft	10,110– 10,200 ft
Ni (ppm)	68	46.25	32.92	30.60	36.28	18.17	23.77
(Ni/Al)*104	7.7	40.15	9.73	10.89	9.99	2.04	2.85
EF		5.21	1.26	1.41	1.30	0.27	0.37
Co (ppm)	19	35.22	125.82	51.60	53.42	149.73	104.40
(Co/Al)*104	2.1	30.57	37.17	18.36	14.71	16.85	12.52
EF		14.56	17.70	8.74	7.01	8.02	5.96
Cu (ppm)	45	40.97	32.68	23.53	37.58	17.07	29.29
(Cu/Al)*104	5.1	35.56	9.65	8.37	10.35	1.92	3.51
EF		6.97	1.89	1.64	2.03	0.38	0.69
Zn (ppm)	95	163.73	170.23	196.74	280.27	77.06	165.94
(Zn/Al)*104	11	142.12	50.29	70.02	77.21	8.67	19.90
EF		12.92	4.57	6.37	7.02	0.79	1.81
V (ppm)	130	99.13	57.92	66.57	74.24	25.74	50.14
(V/Al)*104	15	86.05	17.11	23.69	20.45	2.90	6.01
EF		5.74	1.14	1.58	1.36	0.19	0.40
Cr (ppm)	90	142.85	150.93	135.35	196.18	46.71	100.47
(Cr/Al)*104	10.2	124.00	44.59	48.17	54.04	5.26	12.05
EF		12.16	4.37	4.72	5.30	0.52	1.18
Ba (ppm)	580	896.83	1063.47	515.00	556.06	766.98	3455.12
(Ba/Al)*104	66	778.49	314.17	183.29	153.18	86.30	414.36
EF		11.80	4.76	2.78	2.32	1.31	6.28

**Table 1.** (Continued).

Element	WSA	9660– 9750 ft	9750– 9840 ft	9840– 9930 ft	9930– 10,020 ft	10,020– 10,110 ft	10,110– 10,200 ft	10,200– 10,290 ft
Rb (ppm)	140	61.60	31.90	36.73	41.58	14.02	32.27	25.40
(Rb/Al)*104	16	53.47	9.42	13.07	11.45	1.58	3.87	3.11
EF		3.34	0.59	0.82	0.72	0.10	0.24	0.19
Sr (ppm)	300	180.24	114.33	113.34	124.33	57.11	150.23	87.92
(Sr/Al)*104	34	156.45	33.77	40.34	34.25	6.43	18.02	10.77
EF		4.60	0.99	1.19	1.01	0.19	0.53	0.32
Zr (ppm)	160	236.75	185.52	195.99	208.18	147.83	250.34	179.02
(Zr/Al)*104	18	205.51	54.81	69.75	57.35	16.63	30.02	21.93
EF		11.42	3.04	3.88	3.19	0.92	1.67	1.22
Pb (ppm)	22	23.48	15.96	23.69	35.42	17.70	21.82	18.74
(Pb/Al)*104	2.5	20.38	4.71	8.43	9.76	1.99	2.62	2.30
EF		8.15	1.89	3.37	3.90	0.80	1.05	0.92
U (ppm)	3.7	2.56	1.78	1.96	2.07	0.76	1.59	1.20
(U/Al)*104	0.42	2.22	0.52	0.70	0.57	0.09	0.19	0.15
EF		5.28	1.25	1.66	1.35	0.20	0.45	0.35
Mo (ppm)	1	13.30	10.68	7.77	12.86	4.94	5.69	4.70
(Mo/Al)*104	0.1	11.55	3.16	2.76	3.54	0.56	0.68	0.58
EF		115.45	31.55	27.64	35.41	5.56	6.82	5.76
Y (ppm)	41	23.22	13.12	15.40	17.38	3.60	10.82	8.67
(Y/Al)*104	4.6	20.15	3.87	5.48	4.79	0.41	1.30	1.06
EF		4.38	0.84	1.19	1.04	0.09	0.28	0.23

**Table 1.** (Continued).

Element	WSA	10,290– 10,380 ft	10,380– 10,470 ft	10,470– 10,560 ft	10,560– 10,650 ft	10,650– 10,740 ft	10,740– 10,830 ft	10,830– 10,920 ft
Ni (ppm)	68	36.60	52.90	41.83	41.82	39.77	57.15	54.48
(Ni/Al)*104	7.7	4.52	6.23	4.96	5.04	9.68	10.93	9.32
EF	0.59	0.59	0.81	0.64	0.65	1.26	1.42	1.21
Co (ppm)	19	59.08	30.04	29.01	30.34	47.51	31.11	29.66
(Co/Al)*104	2.1	7.30	3.54	3.44	3.66	11.57	5.95	5.07
EF		3.48	1.69	1.64	1.74	5.51	2.83	2.42
Cu (ppm)	45	39.01	72.07	32.86	38.02	34.75	75.20	54.90
(Cu/Al)*104	5.1	4.82	8.49	3.90	4.58	8.46	14.39	9.39
EF		0.95	1.67	0.76	0.90	1.66	2.82	1.84
Zn (ppm)	95	184.82	336.85	354.01	218.42	263.23	374.79	222.20
(Zn/Al)*104	11	22.85	39.69	42.01	26.32	64.09	71.71	38.02
EF		2.08	3.61	3.82	2.39	5.83	6.52	3.46
V (ppm)	130	55.22	107.15	98.92	95.68	93.75	105.60	103.66
(V/Al)*104	15	0.68	1.26	1.17	1.15	2.28	2.02	1.77
EF		0.05	0.08	0.08	0.08	0.15	0.13	0.12
Cr (ppm)	90	168.83	282.78	456.95	316.78	278.54	413.02	193.70
(Cr/Al)*104	10.2	20.87	33.32	54.23	38.18	67.81	79.02	33.14
EF		2.05	3.27	5.32	3.74	6.65	7.75	3.25
Ba (ppm)	580	1248.17	1150.91	4258.03	4015.67	3826.19	18,872.96	5193.40
(Ba/Al)*104	66	154.31	135.61	505.35	483.96	931.54	3610.98	888.61
EF		2.34	2.05	7.66	7.33	14.11	54.71	13.46

**Table 1.** (Continued).

Element	WSA	10,290– 10,380 ft	10,380– 10,470 ft	10,470– 10,560 ft	10,560– 10,650 ft	10,650– 10,740 ft	10,740– 10,830 ft	10,830– 10,920 ft
Rb (ppm)	140	35.49	63.37	64.55	64.09	60.71	61.74	59.05
(Rb/Al)*104	16	4.39	7.47	7.66	7.72	14.78	11.81	10.10
EF		0.27	0.47	0.48	0.48	0.92	0.74	0.63
Sr (ppm)	300	124.35	236.67	249.57	217.20	200.15	524.14	260.04
(Sr/Al)*104	34	15.37	27.89	29.62	26.18	48.73	100.28	44.49
EF		0.45	0.82	0.87	0.77	1.43	2.95	1.31
Zr (ppm)	160	168.12	288.00	310.47	264.10	302.52	229.66	278.13
(Zr/Al)*104	18	20.78	33.93	36.85	31.83	73.65	43.94	47.59
EF		1.15	1.89	2.05	1.77	4.09	2.44	2.64
Pb (ppm)	22	23.92	42.81	37.66	46.27	37.02	95.92	43.54
(Pb/Al)*104	2.5	2.96	5.04	4.47	5.58	9.01	18.35	7.45
EF		1.18	2.02	1.79	2.23	3.61	7.34	2.98
U (ppm)	3.7	1.51	2.87	2.90	2.69	2.60	2.52	2.73
(U/Al)*104	0.42	0.19	0.34	0.34	0.32	0.63	0.48	0.47
EF		0.45	0.81	0.82	0.77	1.50	1.15	1.11
Mo (ppm)	1	9.84	12.28	6.18	5.44	4.32	30.54	26.61
(Mo/Al)*104	0.1	1.22	1.45	0.73	0.66	1.05	5.84	4.55
EF		12.16	14.47	7.33	6.56	10.51	58.42	45.52
Y (ppm)	41	12.09	29.33	25.78	25.26	24.02	24.95	26.35
(Y/Al)*104	4.6	1.49	3.46	3.06	3.04	5.85	4.77	4.51
EF		0.32	0.75	0.67	0.66	1.27	1.04	0.98

**Table 2.** Major element (wt. %) and trace element (mg/kg) abundances of Agbada Formation sediments and world shale average (WSA). *nd*: Not determined.

Element	Al <sub>2</sub> O <sub>3</sub>	CaO	Cr <sub>2</sub> O <sub>3</sub>	Fe <sub>2</sub> O <sub>3</sub>	K <sub>2</sub> O	MgO	MnO	Na <sub>2</sub> O	P <sub>2</sub> O <sub>5</sub>	SiO <sub>2</sub>	TiO <sub>2</sub>	LOI	Total	As	Ni
WSA	16.7	2.20	<i>nd</i>	6.90	3.60	2.60	<i>nd</i>	1.60	0.16	58.90	0.78	<i>nd</i>	93.44	10	68
9660–9750 ft	15.68	1.51	0.02	7.31	1.59	1.16	0.05	0.55	0.22	58.80	0.95	10.93	98.76	<i>nd</i>	46.25
9750–9840 ft	7.76	1.12	0.02	4.40	0.91	0.76	0.03	0.30	0.13	76.21	0.50	7.52	99.66	<i>nd</i>	32.92
9840–9930 ft	9.87	0.87	0.02	4.40	0.99	0.79	0.03	0.36	0.12	73.20	0.61	8.23	99.50	<i>nd</i>	30.60
9930–10,020 ft	11.04	1.22	0.03	5.35	1.16	0.91	0.04	0.46	0.12	68.23	0.69	9.42	98.68	<i>nd</i>	36.28
10,020–10,110 ft	2.18	0.42	0.01	1.50	0.47	0.30	0.01	0.11	0.03	92.02	0.19	2.40	99.63	<i>nd</i>	18.17
10,110–10,200 ft	6.40	1.35	0.01	3.10	1.00	0.79	0.02	0.29	0.07	78.07	0.47	6.64	98.21	<i>nd</i>	23.77
10,200–10,290 ft	5.31	0.82	0.01	2.57	0.80	0.53	0.02	0.23	0.07	82.41	0.35	6.21	99.32	<i>nd</i>	19.77
10,290–10,380 ft	6.86	1.45	0.03	4.55	1.05	0.82	0.03	0.31	0.11	74.69	0.43	8.31	98.63	<i>nd</i>	36.60
10,380–10,470 ft	15.92	2.51	0.02	8.73	1.61	1.25	0.06	0.60	0.19	48.78	0.89	16.48	97.04	<i>nd</i>	52.90
10,470–10,560 ft	16.79	3.03	0.04	8.53	1.66	1.49	0.07	0.66	0.27	48.36	1.00	15.58	97.49	<i>nd</i>	41.83
10,560–10,650 ft	15.75	2.37	0.06	6.78	1.72	1.37	0.04	0.67	0.17	50.52	0.94	17.13	97.52	<i>nd</i>	41.82
10,650–10,740 ft	15.42	1.37	0.04	6.72	1.69	1.25	0.04	0.56	0.16	57.58	0.86	12.46	98.15	<i>nd</i>	39.77
10,740–10,830 ft	15.28	1.18	0.04	5.88	1.65	1.05	0.04	0.56	0.14	60.31	0.90	11.51	98.52	<i>nd</i>	57.15
10,830–10,920 ft	16.03	3.06	0.03	8.73	1.72	1.31	0.06	0.55	0.19	47.04	0.87	13.19	92.79	<i>nd</i>	54.48
Element	Mn	U	Mo	V	Cr	Co	Ba	Sr	Y	Zr	La	Rb	Cu	Zn	Pb
WSA	850	3.7	1	130	90	19	580	300	41	160	41	140	45	95	22
9660–9750 ft	75.59	2.56	13.30	99.13	142.85	35.22	896.83	180.24	23.22	236.75	51.28	61.60	40.97	163.73	23.48
9750–9840 ft	144.61	1.78	10.68	57.92	150.93	125.82	1063.47	114.33	13.12	185.52	27.96	31.90	32.68	170.23	15.96
9840–9930 ft	145.28	1.96	7.77	66.57	135.35	51.60	515.00	113.34	15.40	195.99	33.75	36.73	23.53	196.74	23.69
9930–10,020 ft	213.04	2.07	12.86	74.24	196.18	53.42	556.06	124.33	17.38	208.18	37.37	41.58	37.58	280.27	35.42
10,020–10,110 ft	523.07	0.76	4.94	25.74	46.71	149.73	766.98	57.11	3.60	147.83	7.41	14.02	17.07	77.06	17.70
10,110–10,200 ft	320.91	1.59	5.69	50.14	100.47	104.40	3455.12	150.23	10.82	250.34	24.10	32.27	29.29	165.94	21.82
10,200–10,290 ft	271.2	1.20	4.70	42.55	98.93	89.31	979.62	87.92	8.67	179.02	17.73	25.40	18.18	106.12	18.74
10,290–10,380 ft	274.14	1.51	9.84	55.22	168.83	59.08	1248.17	124.35	12.09	168.12	24.63	35.49	39.01	184.82	23.92
10,380–10,470 ft	470.64	2.87	12.28	107.15	282.78	30.04	1150.91	236.67	29.33	288.00	58.20	63.37	72.07	336.85	42.81
10,470–10,560 ft	452.8	2.90	6.18	98.92	456.95	29.01	4258.03	249.57	25.78	310.47	53.40	64.55	32.86	354.01	37.66
10,560–10,650 ft	413.91	2.69	5.44	95.68	316.78	30.34	4015.67	217.20	25.26	264.10	51.78	64.09	38.02	218.42	46.27
10,650–10,740 ft	232.35	2.60	4.32	93.75	278.54	47.51	3826.19	200.15	24.02	302.52	50.51	60.71	34.75	263.23	37.02
10,740–10,830 ft	213.23	2.52	30.54	105.60	413.02	31.11	18872.96	524.14	24.95	229.66	50.08	61.74	75.20	374.79	95.92
10,830–10,920 ft	280.62	2.73	26.61	103.66	193.70	29.66	5193.40	260.04	26.35	278.13	50.81	59.05	54.90	222.20	43.54

rocks. Rare earth element mobilization can occur during chemical weathering of bedrock, and source bedrock REE signatures are preserved in the weathering profile because there is no net loss of REE abundance (Condie et al., 1991; Cullers et al., 2000; Kutterolf et al., 2008). Therefore, REE ratios such as La/Yb, Gd/Yb, La/Sm, and Eu/Eu\* (where Eu\* = europium anomalies) of sediments are considered to be similar to provenance and are usually used to determine bulk source composition (Kutterolf et al., 2008; Dabard and Loi, 2012). REE patterns and Th data of the

investigated Agbada Formation sediments indicate the felsic composition of source rocks.

Trace element ratios like Ni/Co, V/Cr, Cu/Zn, and U/Th were used to evaluate paleoredox conditions (Hallberg, 1976; Jones and Manning, 1994). The ratio of uranium to thorium may be used as a redox indicator with the U/Th ratio being higher in organic-rich mudstones (Jones and Manning, 1994). U/Th ratios below 1.25 suggest oxic conditions of deposition, whereas values above 1.25 indicate suboxic and anoxic conditions (Dill et al., 1988;

**Table 3.** Trace and rare earth element ratios of the studied Agbada Formation sediments.

Sample name	Ni/Co	V/Cr	U/Th	Cr/Ni	V/Sc	La/Sc	La/Yb	Gd/Yb	La/Th	La/Sm	Th/Yb
9660–9750 ft	1.31	0.69	0.19	3.09	5.80	3.00	22.69	3.07	3.84	6.13	5.91
9750–9840 ft	0.26	0.38	0.23	4.58	3.92	1.89	21.13	2.84	3.62	6.23	5.83
9840–9930 ft	0.59	0.49	0.21	4.42	4.27	2.17	21.22	2.82	3.71	6.55	5.72
9930–10,020 ft	0.68	0.38	0.20	5.41	4.74	2.39	21.54	2.82	3.59	6.12	5.99
10,020–10,110 ft	0.12	0.55	0.33	2.57	1.98	0.57	13.81	1.96	3.17	6.31	4.35
10,110–10,200 ft	0.23	0.50	0.23	4.23	3.61	1.73	19.56	2.54	3.49	6.27	5.60
10,200–10,290 ft	0.22	0.43	0.26	5.00	3.18	1.32	20.27	2.74	3.77	6.62	5.37
10,290–10,380 ft	0.62	0.33	0.22	4.61	4.01	1.79	19.23	2.76	3.55	5.68	5.42
10,830–10,920 ft	1.76	0.38	0.19	5.35	6.45	3.51	20.07	2.87	3.84	5.83	5.23
10,380–10,470 ft	1.44	0.22	0.19	10.93	6.30	3.40	20.34	2.78	3.58	6.13	5.68
10,470–10,560 ft	1.38	0.30	0.18	7.57	5.90	3.19	19.46	2.81	3.55	5.80	5.48
10,560–10,650 ft	0.84	0.34	0.18	7.00	5.61	3.02	19.65	2.58	3.53	6.03	5.57
10,650–10,740 ft	1.84	0.26	0.18	7.23	6.72	3.19	21.31	3.07	3.56	5.73	5.99
10,740–10,830 ft	1.84	0.54	0.19	3.56	6.48	3.18	19.58	2.83	3.62	5.68	5.41
Minimum	0.12	0.22	0.18	2.57	1.98	0.57	13.81	1.96	3.17	5.68	4.35
Maximum	1.84	0.69	0.33	10.93	6.72	3.51	22.69	3.07	3.84	6.62	5.99
Average	0.94	0.41	0.21	5.40	4.93	2.45	19.99	2.75	3.60	6.08	5.54
Standart deviation	0.64	0.13	0.04	2.18	1.46	0.90	2.04	0.27	0.17	0.30	0.42

**Table 3.** (Continued).

Sample name	Th/U	U/Pb	Eu/Eu*	V/Ni	Cr/Th	Th/Co	Th/Cr	Cu/Zn	Th/Sc	V/(Ni+V)
9660–9750 ft	5.23	0.11	0.72	2.14	10.70	0.38	0.09	0.25	0.78	0.68
9750–9840 ft	4.35	0.11	0.67	1.76	19.56	0.06	0.05	0.19	0.52	0.64
9840–9930 ft	4.65	0.08	0.68	2.18	14.87	0.18	0.07	0.12	0.58	0.69
9930–10,020 ft	5.03	0.06	0.71	2.05	18.87	0.19	0.05	0.13	0.66	0.67
10,020–10,110 ft	3.07	0.04	0.78	1.42	20.00	0.02	0.05	0.22	0.18	0.59
10,110–10,200 ft	4.34	0.07	0.78	2.11	14.56	0.07	0.07	0.18	0.50	0.68
10,200–10,290 ft	3.91	0.06	0.72	2.15	21.05	0.05	0.05	0.17	0.35	0.68
10,290–10,380 ft	4.58	0.06	0.73	1.51	24.34	0.12	0.04	0.21	0.50	0.60
10,830–10,920 ft	5.28	0.07	0.68	2.03	18.65	0.50	0.05	0.21	0.91	0.67
10,380–10,470 ft	5.14	0.08	0.70	2.37	30.64	0.51	0.03	0.09	0.95	0.70
10,470–10,560 ft	5.43	0.06	0.71	2.29	21.74	0.48	0.05	0.17	0.90	0.70
10,560–10,650 ft	5.51	0.07	0.73	2.36	19.47	0.30	0.05	0.13	0.86	0.70
10,650–10,740 ft	5.59	0.03	0.87	1.85	29.32	0.45	0.03	0.20	0.90	0.65
10,740–10,830 ft	5.15	0.06	0.71	1.90	13.80	0.47	0.07	0.25	0.88	0.66
Minimum	3.07	0.03	0.67	1.42	10.70	0.02	0.03	0.09	0.18	0.59
Maximum	5.59	0.11	0.87	2.37	30.64	0.51	0.09	0.25	0.95	0.70
Average	4.80	0.07	0.73	2.01	19.83	0.27	0.05	0.18	0.68	0.66
Standart deviation	0.71	0.02	0.05	0.29	5.60	0.19	0.02	0.05	0.24	0.04

Nath et al., 1997; Jones and Manning, 1994). The studied sediments show low U/Th ratios (~0.18–0.33; average = 0.21) (Tables 3 and 4), which imply that the Agbada Formation sediments were deposited in an oxygenated bottom water condition. Th/U ratios in the sediments range between ~5.59 and 3.07 with an average value of ~4.80, which indicates oxidizing conditions. Th/U ratios are high in oxidizing conditions and low in reducing conditions (Kimura and Watanabe, 2001).

A few authors have used the V/Cr ratio as an indicator of bottom water oxygenated condition (Bjorlykke, 1974; Shaw et al., 1990; Nagarajan et al., 2007). Chromium is mainly incorporated in the detrital fraction of sediments and it may substitute for Al in the structure of clays (Bjorlykke, 1974). Vanadium may be bound to organic matter by the amalgamation of V<sup>4+</sup> into porphyrins, and it is normally found in sediments deposited in reducing environments (Shaw et al., 1990; Kimura and Watanabe, 2001). V/Cr ratios above 2 indicate anoxic conditions, whereas values below 2 imply oxic conditions (Jones and Manning, 1994). The V/Cr ratios in Agbada Formation sediments range from ~0.22 to 0.69, with an average value of ~0.41 (Tables 3 and 4), which indicates that Agbada Formation sediments were deposited in an oxic depositional condition. Numerous authors have used the Ni/Co ratio as a redox indicator (Bjorlykke, 1974; Brumsack, 2006; Nagarajan et al., 2007). Ni/Co ratios below 5 indicate oxic environments, whereas ratios above 5 suggest suboxic and anoxic environments (Jones and Manning, 1994). The Ni/Co ratios vary between ~0.12 and 3.58 with an average value of ~1.11 (Table 2), implying that Agbada Formation sediments were deposited in a well-oxygenated bottom water condition. The Cu/Zn ratio is also used as a redox parameter (Hallberg, 1976). High Cu/Zn ratios indicate reducing depositional conditions, while low Cu/Zn ratios suggest oxidizing conditions (Hallberg, 1976). Consequently, the low Cu/Zn ratios vary between ~0.05 and 0.22 with an average value of ~0.17 in the studied Agbada Formation sediments (Tables 3 and

4), suggesting sediment deposition under oxic conditions. V/(Ni+V) ratios below 0.46 indicate oxic environments, but ratios above 0.54 to 0.82 suggest suboxic and anoxic environments (Hatch and Levant, 1992). The V/(Ni + V) ratios in the Agbada Formation sediments encountered at the KR-1 well vary between ~0.59 and 0.70 with an average value of ~0.66, which suggests that there might be rare occurrence of suboxic to anoxic environments of deposition. V/Sc ratios below 9.1 indicate an oxic environment of deposition (Hetzl et al., 2009). The V/Sc ratios in the Agbada Formation sediments vary between ~1.98 and 6.72 with an average value of ~4.93, which indicates an oxic environment of deposition (Tables 3 and 4). Based on REE studies of the early Cretaceous sediments, numerous geoscientists convincingly argued that the REE patterns (including Eu\* anomalies), though mostly dependent on their provenance, can also be controlled by *f*O<sub>2</sub> and sedimentary environment (Ganai and Rashid, 2015). They observed that when *f*O<sub>2</sub> is low (a reducing environment), the sediments deposited should be characterized by low REE values and a positive europium anomaly (Eu\*), whereas sediments deposited in oxidizing conditions (i.e. *f*O<sub>2</sub> is high) should be characterized by high total REE and Eu depletion (Ganai and Rashid, 2015). As a result, it appears that the Agbada Formation sediments recovered from the KR-1 well, which are characterized by high total REEs and strong negative Eu anomaly, were deposited in an oxidizing environment.

#### 4.6. Rare earth element geochemistry

A comparison of the REE contents in this study and a number of works on the behavior of REEs in secondary environments is shown in Table 5. Standards that are normally used include the world shale average (WSA), as calculated by Piper (1974) from published analyses (Haskin and Haskin, 1964; Wedepohl, 1995); the North American Shale Composite (NASC), analyzed by Gromet et al. (1984); the Upper Continental Crust (UCC), with several slightly different values reported by several authors (e.g., Wedepohl, 1969–1978; McLennan,

**Table 4.** Some trace element ratios to evaluate paleoredox conditions.

Element ratios	Oxic	Dysoxic	Suboxic to anoxic	Euxinic
Ni/Co <sup>1</sup>	<5	5–7	>7	
V/Cr <sup>1</sup>	<2	2–4.25	>4.25	
U/Th <sup>1</sup>	<0.75	0.75–1.25	>1.25	
V/(Ni + V) <sup>2</sup>	<0.46	0.46–0.60	0.54–0.82	>0.84
V/Sc <sup>3</sup>	<9.1			

<sup>1</sup> Jones and Manning, 1994; <sup>2</sup> Hatch and Levant, 1992; <sup>3</sup> Akinyemi et al., 2013; Adebayo et al., 2015a.

**Table 5.** Comparison of normalized rare earth element (REE) contents with literature data.

Element	WSA	UCC	PAAS	NASC	Aver. chondrites	9660–9750 ft	9750–9840 ft	9840–9930 ft	9930–10,020 ft	10,020–10,110 ft	10,110–10,200 ft	10,200–10,290 ft
La	41	30	38.20	31.1	0.32	51.28	27.96	33.75	37.37	7.41	24.10	17.73
Ce	83	64	79.60	66.7	0.90	107.79	58.13	69.52	76.76	17.01	50.37	38.19
Pr	10.1	7.1	8.83	7.7	0.13	11.96	6.44	7.62	8.69	1.81	5.41	4.15
Nd	38	26	33.90	27.4	0.57	45.28	24.23	28.35	32.30	5.98	20.60	15.02
Sm	75	4.5	5.55	5.59	0.21	8.37	4.49	5.16	6.11	1.18	3.84	2.68
Eu	1.61	0.88	1.08	1.18	0.07	1.81	0.90	1.07	1.27	0.28	0.88	0.60
Gd	6.35	3.8	4.66	4.9	0.31	6.93	3.76	4.49	4.89	1.05	3.13	2.40
Tb	1.23	0.64	0.77	0.85	0.05	0.90	0.52	0.55	0.64	0.16	0.42	0.28
Dy	5.5	3.5	4.68	4.17	0.30	4.99	2.85	3.35	3.81	0.73	2.30	1.68
Ho	1.34	0.8	0.99	1.02	0.07	0.90	0.53	0.63	0.71	0.15	0.49	0.33
Er	3.75	2.3	2.85	2.84	0.21	2.40	1.48	1.64	1.94	0.42	1.10	0.88
Tm	0.63	0.33	0.41	0.84	0.03	0.35	0.18	0.24	0.26	0.10	0.18	0.14
Yb	3.53	2.2	2.82	3.06	0.18	0.90	0.52	0.55	0.64	0.16	0.42	0.28
Lu	0.61	0.32	0.43	0.46	0.03	0.33	0.19	0.21	0.25	0.08	0.18	0.14
ΣREE	271.65	146.37	184.773	157.81	3.393	244.1635	132.155	157.094	175.6255	36.525	113.41	84.49
Element	WSA	UCC	PAAS	NASC	Aver. chondrites	10,290–10,380 ft	10,380–10,470 ft	10,470–10,560 ft	10,560–10,650 ft	10,650–10,740 ft	10,740–10,830 ft	10,830–10,920 ft
La	41	30	38.20	31.1	0.32	24.63	58.20	53.40	51.78	50.51	50.08	50.81
Ce	83	64	79.60	66.7	0.90	52.94	124.13	111.31	109.18	103.82	105.18	107.23
Pr	10.1	7.1	8.83	7.7	0.13	5.79	13.69	12.48	12.35	11.61	11.90	12.12
Nd	38	26	33.90	27.4	0.57	22.76	51.97	47.27	47.16	43.64	45.20	46.62
Sm	75	4.5	5.55	5.59	0.21	4.34	9.98	8.72	8.92	8.38	8.74	8.95
Eu	1.61	0.88	1.08	1.18	0.07	0.94	2.04	1.83	1.90	1.77	2.28	1.88
Gd	6.35	3.8	4.66	4.9	0.31	3.54	8.33	7.31	7.47	6.62	7.22	7.34
Tb	1.23	0.64	0.77	0.85	0.05	0.48	1.12	1.02	1.00	0.94	1.00	0.98
Dy	5.5	3.5	4.68	4.17	0.30	2.49	6.39	5.76	5.46	5.00	5.40	5.70
Ho	1.34	0.8	0.99	1.02	0.07	0.49	1.16	1.01	1.03	0.96	0.99	1.00
Er	3.75	2.3	2.85	2.84	0.21	1.33	3.19	2.78	2.77	2.57	2.47	2.75
Tm	0.63	0.33	0.41	0.84	0.03	0.18	0.43	0.39	0.37	0.36	0.35	0.36
Yb	3.53	2.2	2.82	3.06	0.18	0.48	1.12	1.02	1.00	0.94	1.00	0.98
Lu	0.61	0.32	0.43	0.46	0.03	0.17	0.42	0.36	0.34	0.34	0.34	0.37
ΣREE	271.65	146.37	184.77	157.81	3.39	120.52	282.15	254.63	250.72	237.44	242.13	247.07

WSA (Piper, 1974); UCC (Wedepohl, 1969–1978; McLennan, 1989; Rudnick and Gao, 2003); PAAS (McLennan, 2001); NASC (Gromet et al., 1984); average chondrites (Schmidt et al., 1963).

1989; Rudnick and Gao, 2003), but with rather similar interelement concentrations; the Post-Archean Australian Shale (PAAS), proposed by McLennan (2001); and finally an average of chondrites (Schmidt et al., 1963). The

concentrations of the REEs in these standards represent two compositional extremes of siliciclastic source-rocks, one felsic (WSA, UCC, PAAS, and NASC) and the second ultramafic (chondrites) (Piper and Bau, 2013).

Table 5 shows that the concentration range of La in Agbada Formation sediments is from ~58.20 to 7.41 ppm with an average value of ~38.50 ppm and standard deviation of ~15.96. The average value of La in Agbada Formation samples is lower than in WSA (Table 5) but higher than those of other standards such UCC, PAAS, NASC, and average chondrites.

Cerium contents in studied samples range between ~124.13 and 17.01 ppm with an average value of ~ 80.82 ppm and standard deviation of ~33.27. The average value of Ce in the studied samples is relatively lower than in WSA and less than that of UCC, PAAS, NASC, and average chondrites. The concentration range of Pr in the studied sediments is between ~13.69 and 1.81 ppm with an average value of ~ 9.00 ppm and standard deviation of ~3.78. The average concentration value of Pr in Agbada Formation samples is less than in WSA but higher than those of other standards. Concentration levels of Nd range from ~51.97 to 5.98 ppm with an average value of ~34.02 ppm and standard deviation of ~14.57. The average concentration level of Nd is less than in WSA but higher than those of other standards.

Concentration levels of Sm in Agbada Formation sediments range from ~9.98 to 1.18 ppm with an average value of ~6.42 and standard deviation of ~2.79. Concentration levels of Eu in the studied samples range between ~2.28 and 0.28 ppm with an average value of ~1.39 ppm and standard deviation of ~0.61. The average concentration of Sm is less than in WSA but higher than those of the other commonly used standards (Table 5).

The concentration levels of Gd range from ~8.33 to 1.05 ppm with an average value of ~5.32 ppm and standard deviation of ~2.28. The average value of Gd in the studied samples is less than in WSA but less high than those of the other commonly used standards. The levels of Tb in Agbada Formation sediments range between ~8.33 and 1.05 ppm with an average value of ~0.72 ppm and standard deviation of ~0.31. The average concentration level of Tb in studied samples is less than in WSA, PAAS, and NASC but higher than in UCC and average chondrites. Concentration levels of Dy in studied samples range between ~6.39 and 0.73 ppm with an average value of ~3.99 ppm and standard deviation of ~1.77. The average value of Dy in Agbada Formation sediments is higher than in UCC and average chondrites but lower than in the other considered standards. The content levels of Ho in studied samples range from ~1.16 to 0.15 ppm with an average value of ~ 0.74 ppm and standard deviation of ~0.31. The average concentration value of Ho in studied samples is less than those of all considered commonly used standards. Concentration levels of Er range from ~3.19 to 0.42 ppm with an average value of ~1.98 ppm and standard deviation of 0.31. The average concentration of Er in studied samples

is higher than in average chondrites but lower than in the other standards. Concentration levels of Tm in studied samples range between ~0.43 and 0.10 ppm with an average value of ~0.28 ppm and standard deviation of ~0.11. The average concentration of Tm in studied samples is higher than in average chondrites but lower than in other considered standards. The contents of Yb in Agbada Formation sediments range from ~1.12 to 0.16 ppm with an average value of ~0.72 ppm and standard deviation of ~0.31. The average concentration of Yb in studied samples is higher than in average chondrites but lower than in other considered standards. Contents of Lu in Agbada Formation sediments range from ~0.42 to 0.08 ppm with an average value of ~0.27 ppm and standard deviation of ~0.10. The average concentration level of Lu is higher than in average chondrites but lower than its concentration in other considered standards.

Cerium anomaly may perhaps be quantified by comparing the measured concentration (Ce) with an expected concentration (Ce\*) obtained by interpolating between the values of the neighboring elements. Ce anomalies in shales of the anoxic facies are attributed to eustatic sea level changes (Wilde et al., 1996). Similar to Mn, Ce<sup>4+</sup> is less soluble under oxic conditions, whereas under anoxic conditions it will be mobilized, leading to the depletion in Ce in anoxic sediments relative to those deposited under oxic conditions. A negative Ce anomaly would be indicative of postdepositional remobilization of Ce in the water column.

Table 6 shows two different values given for the Ce anomaly, which are based on different calculations. Taylor and McLennan (1985) suggested the use of the geometric mean  $Ce^* = \sqrt{(La \cdot Pr)}$ . The ratio  $Ce/Ce^*$  is then a measure of the anomaly, with values greater than unity being termed positive. Wilde et al. (1996) promoted the use of the arithmetic mean  $Ce^* = (La + Pr)/2$  and calculated the logarithm of the ratio  $Ce/Ce^*$ . Both calculations lead to the same values as  $Ce^*$  by KR-1 well samples mostly showing positive anomaly values. Therefore, the Agbada Formation sediments were deposited under oxic conditions, which indicates the incorporation of this cerium from the water column.

## 5. Conclusions

Lithologically, glauconite and rare pyrite are the most important accessory minerals in the studied well, indicating a slightly anoxic marginal marine environment of deposition while the palynomorph percentage distribution shows that there are more terrestrially derived miospores (dominated by *Zonocostites ramonae*, *Psilatricolporites crassus*, *Arostichum aurium*, and *Laevigatosporites* sp.) than marine phytoplanktons. These indicate that the main

**Table 6.** Ce anomaly for Agbada Formation sediments (two quantification approaches are given).

Authors	Equations	9660– 9750 ft	9750– 9840 ft	9840– 9930 ft	9930– 10,020 ft	10,020– 10,110 ft	10,110– 10,200 ft	10,200– 10,290 ft
Taylor and McLennan (1985)	$Ce^* = \sqrt{(La \cdot Pr)}$	24.76	13.42	16.04	18.02	3.66	11.42	8.57
	Ce/Ce*	4.35	4.33	4.34	4.26	4.64	4.41	4.45
Wilde et al. (1996)	$Ce^* = (La + Pr)/2$	57.26	31.18	37.56	41.71	8.32	26.80	19.80
	Log (Ce/Ce*)	0.64	0.64	0.64	0.63	0.67	0.64	0.65
Authors	Equations	10,290– 10,380 ft	10,380– 10,470 ft	10,470– 10,560 ft	10,560– 10,650 ft	10,650– 10,740 ft	10,740– 10,830 ft	10,830– 10,920 ft
Taylor and McLennan (1985)	$Ce^* = \sqrt{(La \cdot Pr)}$	24.63	58.20	53.40	51.78	50.51	50.08	50.81
	Ce/Ce*	2.15	2.13	2.08	2.11	2.06	2.10	2.11
Wilde et al. (1996)	$Ce^* = (La + Pr)/2$	36.95	87.30	80.09	77.66	75.77	75.11	76.22
	Log (Ce/Ce*)	0.33	0.33	0.32	0.32	0.31	0.32	0.32

Ce\* = Cerium anomaly.

environment of deposition in the KR-1 well is coastal to marginal marine consisting of coastal deltaic-inner neritic, made up of tidal channel and shoreface deposits (Adeigbe et al., 2013). Geochemical results show that the average concentrations of considered REEs are less than their concentrations in WSA. Trace metal ratios such as Th/Cr, Cr/Th, Th/Co, and Cr/Ni suggest that the sediments were derived from felsic source rocks. REE patterns such as La/Yb, Gd/Yb, La/Sm, and Eu/Eu\* and Th data confirmed the felsic composition of the sediments. Ratios of U/Th, Ni/Co, Cu/Zn, and V/Sc indicate well-oxygenated bottom water conditions. Estimated Eu\* and Ce\* anomalies obtained from the studied samples suggest an oxidizing

environment of deposition. Nonetheless, the ratios of V/Cr suggest a range of environmental conditions. Moreover, ratios of V/(Ni+V) suggest rare suboxic to anoxic environments of deposition.

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