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Geochemistry of Palaeoproterozoic Kajrahat Limestone, Vindhyan Supergroup, central India: insights into depositional conditions and sources of rare earth elements

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Abstract

The Lower Vindhyan (Semri) Group substantially consists of unmetamorphosed and undeformed sedimentary sequences of Palaeoproterozoic age. A detailed geochemical study has been carried out to infer the paleoenvironment and the source of REEs for the Palaeoproterozoic Kajrahat Limestone, Vindhyan Supergroup, Central India. This study investigates the geochemical composition of major, trace and rare earth elements (REEs). The studied limestones, have a high percentage of CaO (ranging from 30.87 to 48.59) implies that calcite was the primary mineral phase in these carbonates. SiO₂ is second most abundant major oxide. Sr is dominant trace element showing a negative correlation with CaO. All trace elements exhibit depletion with respect to Post-Archean Australian Shale (PAAS). The PAAS-normalized REE pattern of studied limestone has a relatively uniform pattern, with slightly enriched LREE compared to HREE with negative Ce anomaly. Collectively, low concentration of U and U/Th, Ce/Ce* ratios clearly indicates an oxic depositional condition for the samples. The ΣREE shows a positive correlation with SiO₂, TiO₂ and Al₂O₃ (r^2 =0.87, 0.84 and 0.91 respectively) and a week positive correlation with CaO (r^2 =0.12) suggested that siliciclastic sediments also serve as a source for the REEs beside seawater. This, in turn, indicates that the deposition of this limestone occurred in a coastal/shallow marine environment with some contribution from continental part.

Keywords Geochemistry · Palaeoproterozoic limestone · Depositional environment · Vindhyan Supergroup

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Introduction

One of the prime focus of sedimentological study is to concede ancient sedimentary environments, and the present study attempts to decode the depositional environment of Palaeoproterozoic Kajrahat Limestone of Central India. Characteristics of sedimentary rocks depend on formation, transport and depositional processes and include physical, chemical and biological factors in any depositional environment (Nichols 2009). Geochemistry is an important tool used in ascertaining the depositional imprints and overprints of post-depositional processes on sediments. The chemical composition of carbonate rocks helps to infer the physicochemical conditions in the deposition basin. Major oxides and trace elements act as indicators, revealing the composition of the carbonate rocks and provide information about the environmental conditions during deposition (Srivastava and Singh 2018). The source of the chemical elements to the oceans are various such as river, wind, hydrothermal sources, groundwater, etc. (Martin and Whitfield 1983; Birgel et al. 2011; Coimbra et al. 2015). Carbonate deposition and seafloor redox dynamics act as depositional sinks because major and trace elements are incorporated into the calcium carbonate lattice, as seen in modern neritic and oceanic environments (Takematsu et al. 1993). Therefore major oxides and trace elements act as indicators of depositional environment of carbonate rocks. The low solubility of rare earth elements (REEs) in seawater means they are entirely incorporated into clastic sediments due to their short residence time in the water. Additionally, REEs tend to remain immobile during subsequent processes like diagenesis (Chaudhuri and Cullers 1979; Gong et al. 2018) and metamorphism (Muecke et al. 1979), chemical weathering and transport (Yang et al. 2017). Consequently, REE concentrations in ancient carbonate rocks serve to distinguish the sources of REEs, whether they originate from marine or non-marine milieu (Banner et al. 1988; Frimmel 2009; Zhao et al. 2009). The average provenance compositions can potentially be determined through REE patterns (Taylor and McLennan 1985; Bakkiaraj et al. 2010). Carbonate rocks typically contain low concentrations of REEs (Goldberg et al. 1963; Haskin and Haskin 1966; Tlig and M'Rabet 1985; Nogueira et al. 2019) and the distribution of REEs in these rocks is mostly affected by the conditions of deposition (Murray et al. 1990, 1992; Patra and Singh 2017; Srivastava and Singh 2018) and diagenetic processes (Scherer and Seitz 1980; Schieber 1988; Armstrong-Altrin et al. 2003; Morad et al. 2010; Fu et al. 2011). REEs are useful for determining the type of depositional environments including widespread marine anoxia (Liu et al. 1988; German and Elderfield 1990), oceanic palaeo-redox conditions (Wang et al. 1986; Elderfield and Pagett 1986; Kato et al. 2002), changes in surface productivity (Toyoda et al. 1990), distance from the source area (Murray et al. 1991), lithology and diagenesis (German and Elderfield 1990; Nath et al. 1992; Madhavaraju and Lee 2009) and palaeogeography and depositional models (Kamber and Webb 2001; Kemp and Trueman 2003).

Examining rare earth elements (REEs), mainly LREE, in sedimentary rocks offers insights into the initial composition of seawater. As a result, ancient sedimentary rocks have been recognized as trustworthy sources for REE proxies (Holser 1997). The Rare Earth Elements (REEs) in carbonate rocks preserved valuable information on paleo-seawater chemistry and local-regional redox conditions (Zhang and Shields 2023). REEs are substituted for Ca^{2+} in the carbonate lattice. It is believed that REEs exhibit greater stability compared to oxygen or carbon isotopes in the carbonate lattice during diagenetic processes (Zhong and Mucci 1995). Furthermore, diagenetic fluids typically exhibit low concentrations of REEs ranging from 10^{-6} to 10^{-4} ppm (Sholkovitz et al. 1989; Banner and Hanson 1990). Consequently, REE concentrations in ancient carbonate rocks are likely to remain unchanged, even during periods of considerable diagenesis (Webb and Kamber 2000; Frimmel 2009). Seawater REE patterns found in marine chemical sediments, regardless of their age of deposition, show a uniform LREE depletion; enrichment of La, depletion of Ce and slight enrichment of Gd, in shale-normalized patterns (Shields and Webb 2004; Bolhar and Kranendonk 2007; Zhang and Nozaki 1996).

The Indian subcontinent is crucial in both Columbian and Rodinian reconstructions (Li et al. 2008; Meert and Santosh 2017). It has numerous Proterozoic sedimentary basins, mostly undeformed and unmetamorphosed, holding potential clues to tectonic, climatic and biological evolution on Earth's Proterozoic Eon. Only a few studies, explore evidence for Proterozoic global events from the carbonate formations in these basins (e.g., Ray et al. 2003; Mohanty et al. 2015; Ansari et al. 2018; George et al. 2018; Choudhuri et al. 2023). The Vindhyan Basin provides most of the data, boasting the largest Proterozoic sedimentary sequence in India. Ray et al. (2002, 2003) analyzed isotope systematic in carbonate sequences of the Vindhyan Supergroup, India to explore their geological significance, and age. According to their study of Pb-Pb dating and Sr isotope stratigraphy established the Kajrahat Limestone as Palaeoproterozoic age $(1600 \pm 130 \text{ Ma})$. Furthermore, there has been an investigation into the depositional environment by stable isotope composition of the Kajrahat Limestone (Banerjee et al. 2007). Singh et al. (2020) have recorded soft sedimentary deformation structures (SSDs) in the Kajrahat Limestone and have envisaged that earthquakes might have occurred multiple times between 1700 and 1600 Ma in Central India. Furthermore, this implies that the Vindhyan Basin experienced instability and the seismicity was primarily a result of rift-related tectonics in this intra-cratonic basin. Even after years of research, a comprehensive study of the REE geochemistry of Palaeoproterozoic Kajrahat Limestone of Lower Vindhyan succession remains unavailable. Therefore, the present study major, trace and REEs geochemistry of Kajrahat Limestone are analyzed and their variations are recorded to gain insights into the physico-chemical conditions that were prevailed during formation of these carbonates of the Vindhyan Basin. This study aims to investigate the depositional setting, identify potential REE sources, and provide insights into the variations in Eu and Ce anomalies within the studied limestones. Additionally, this study will contribute to understand the seawater chemistry during Palaeoproterozoic times in Central India.

Geological setting

The Vindhyan Supergroup of Central India records a considerable portion of Proterozoic time and holds the distinction of being the largest of the Proterozoic sedimentary basins in the Indian subcontinent, known as the Purana Basins (Kale and Phansalkar 1991). This sedimentary basin holds unmetamorphosed and slightly deformed layers that are approximately 4500 m thick. These beds have wellpreserved sedimentary structures and provide evidence of paleo-environmental conditions during a significant portion of the Proterozoic eon (~1800-900 Ma) with good time resolution (Deb et al. 2002; Ray 2006; Malone et al. 2008; Gopalan et al. 2013). The Proterozoic Vindhyan Supergroup is found in Central India and covers an area of ~ 100,000 km². A substantial portion of this Supergroup is concealed under the Gangatic alluvium and the Deccan traps (Gopalan et al. 2013). The western boundary of this Supergroup is defined by the Great Boundary Fault (GBF), while the southern limit is marked by the Narmada-Son lineament (Fig. 1A). These sequences have been found in two areas: the western region (Rajasthan) and the eastern region (Son valley) (Fig. 1A). The Supergroup overlies the Bundelkhand Granite massif (2.5 Ga; Azmi et al. 2008). The Semri Group occurs in the lower part of the Vindhyan succession and is primarily composed of carbonates and shales with subordinate sandstones and volcaniclastic units (Ramakrishnan and Vaidyanadhan 2008). The Upper Vindhyan succession comprises three groups, i.e., Kaimur, Rewa and Bhander.

Vindhyan sequences were deposited mostly in shallow marine environments (Chanda and Bhattacharyya 1982). According to Bose et al. (1997, 2001), the Lower Vindhyan sedimentation occurred in an intracratonic rift basin, and the change from a rifted basin to an intracratonic sag basin occurred during the Upper Vindhyan deposition. Well preserved sections of the Semri and Kaimur Groups of the Vindhyan Supergroup are exposed in Son Valley, Central India. The Semri Group is primarily characterised by its volcanogenic and biochemical sediments, whereas the overlying Kaimur Group is predominantly composed of siliciclastic deposits. The Kajrahat Limestone of the Semri Group is conformably underlain by the Basal Conglomerate and, in turn, is overlain by the Porcellanite Formation (Table 1). The Porcellanite Formation contains volcanic tuffs and rhyolite indicating significant felsic volcanic activity in the early Mesoproterozoic Era. The Kajrahat Limestone is exposed along the southern side of the Vindhyan basin in the Son Valley region (Rasmussen et al. 2002) (Fig. 1B).

The lower most and middle part of the Kajrahat Limestone Formation is exposed around Dala town and Kajrahat village, which is 1 km east of Dala town. The formation composed of alteration of thinly bedded and laminated limestones and characterized by the presence of stylolites along the stromatolitic layers. The upper division of this limestone formation consists of several shallowing-upward stromatolite cycles. The general trend is larger stromatolites are followed by smaller one at the top of the cycle (Banerjee et al. 2007). The middle part of the succession is characterized by Soft Sedimentary Deformation (SSD) structures. These are convolute bedding, contorted cross-bedding, autoclastic breccia, and small-scale folds characterized by narrow anticlines and broad synclines (Singh et al. 2020). The studied limestones are exposed in the vicinity of Dala town with a notable presence of different lithologies. The succession starts with thinly bedded grey coloured limestone which turns into laminated and then thickly bedded. After thickly bedded unit there is again alteration of laminated and thinly bedded unit (Fig. 2). In the study area three different facies



Fig.1 A Generalized geological map of the Vindhyan Supergroup, India (after Shukla et al. 2020). The samples of Kajrahat Limestone of Semri Group analyzed in the present study were collected

from nearby Dala Town. **B** Detailed geological map of the Vindhyan Supergroup in Son Valley (after Auden 1933). The yellow square represents the sampling area

Table 1	Lithostratigraphic succession of	Vindhyan Supergroup of Son valley	, Central India (modified after Auden 1933; Kumar 1978)
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	Kaimur Group		
	Unconformity		
Vindhyan Supergroup	Semri Group	Rohtas Limestone Formation Kheinjua Formation	1599±48 Ma (Sarangi et al. 2004)
		Chopan Porcellanite Formation	1631±5 Ma (Rasmussen et al. 2002); 1628±8 Ma (Ray et al. 2002); 1640±4 Ma (Bickford et al. 2017)
		Kajrahat Limestone Formation Basal Conglomerate Formation	1600±130 Ma (Ray et al. 2003)

Fig. 2 Litho-column displaying vertical organisation of different lithofacies in the Palaeoprote-rozoic Kajrahat Limestone with sample locations



are identified: laminated, thinly bedded and thickly bedded limestone. Gradational contacts are observed in between the different unit throughout the succession. Elephant skin weathering and stylolites are commonly found within these limestones.

Methodology

Unweathered limestone samples were collected from the exposed outcrops of Kajrahat Limestone Formation, Vindhyan Basin located near Dala town. The selected samples were grinded in an agate mortar after washing with distilled water and subsequent air drying. The chemical analyses of seventeen representative samples were performed for major, trace and rare earth elements (REEs) at Wadia Institute of Himalayan Geology (WIHG), Dehradun, India. Sample preparation of major elements used pressed powder pellet mode and analyzed using X-Ray Fluorescence (XRF). The samples were dissolved using standard HF-HNO₃ dissolution technique for trace elements and REEs. About ~ 50 mg powdered samples were completely dissolved using ultrapure acids. The final solutions were prepared in 2% HNO₃ Then to avoid matrix effect highly diluted samples were analyzed using ICP-MS along with USGS rock standard BHVO-2.

Results

Major elements

Major element concentrations are listed in Table 2. The dominant oxide among all the oxide is CaO making up 30.87–48.59% (Table 2). CaO also shows a positive correlation with SiO_2 and TiO_2 (Table 4). On the other hand, MgO, shows variation ranging from 0.51 to 19.17%, (Table 2) and has a positive correlation with Fe₂O₃, MnO and Na₂O (Table 4). The SiO₂ content ranges from 1.14 to 15.42%(Table 2) and shows negative correlation with Fe_2O_3 , MgO, MnO, but positive correlation with TiO₂, Al₂O₃, CaO, Na₂O, K_2O and P_2O_5 (Table 4). Al₂O₃ ranges from 0.23 to 4.09% (Table 2), and there is a negative correlation with CaO, MgO, and MnO (Table 4). In these limestones, the concentration of TiO₂ ranges from 0.02 to 0.11%. The amount of Fe_2O_3 is relatively low ranging from 0.6 to 1.86% (Table 2). Additionally, Fe₂O₃ and CaO have a negative correlation (Table 4). These limestones have very low concentrations of MnO (0.01–0.05), K₂O (0.34–1.27), Na₂O (0.02–0.05) and P_2O_5 (0.12–0.78) (Table 2). The dolomitization ratio, as suggested by Singh et al. (2013) varies from 0.01 to 0.62 in the studied limestones. The PAAS-normalized major elements concentrations exhibit notable variations. The CaO content is higher, while the Na_2O content is moderately depleted in studied limestones (Fig. 3).

Trace elements

The concentrations of trace elements are shown in Table 2. These trace elements are normalized using PAAS values (Taylor and McLennan 1985) and these values are important to know the minute enrichment and deficiencies in certain elements (Rollinson 1993). Additionally, their distribution is represented in the spider diagram (Fig. 4). The concentration of Cu and Sr shows large variations while the concentrations of other elements are relatively minimal (Table 2). As compared to PAAS, Ba and Co are significantly depleted, whereas Cu concentrations are roughly similar in some samples. The PAAS normalized patterns of studied limestones shows an enrichment in V, Sc, Cu, Zn, U and Sr while depletion in Ba, Co, Rb, Th, Zr (Fig. 5). The Sr concentrations in these limestones ranging from 126 to 514 ppm are lower than the average lithospheric carbonate value of 610 ppm. The Ba concentrations range from 10.1 to 38.4 ppm while U varies from 0.3 to 1.3 ppm (Table 2).

Rare earth elements

The concentrations of REEs are reported in Table 2 and are normalized to PAAS values (Taylor and McLennan 1985) in Fig. 5. The limestones show seawater like REE pattern with negative Ce and Eu anomaly (except for three samples which shows positive Eu anomaly). The Σ REE content varies significantly among the limestone samples, ranging from 13.7 to 68.3 ppm. The LREE ranges from 13 to 64.2 ppm, while the HRRE ranges from 0.7 to 4.1 ppm. The higher content of REE than typical marine carbonate indicates the presence of silt and clay fractions because REE accommodated easily in the clay structure (McLennan 1989). The Σ REE shows a positive correlation with SiO₂, TiO₂, Al₂O₃ ($r^2 = 0.87, 0.84$, 0.91, respectively) and a week positive correlation with CaO $(r^2=0.12)$ (Table 4). Correlation pattern indicates that the concentration of these elements may control by the influence of terrigenous material. The PASS-normalized REE patterns of these limestones (Fig. 5) exhibit a minor enrichment in LREE relative to HREE. Additionally, these samples exhibit a negative Ce anomaly (Ce/Ce*=0.66-0.89) (Table 3). Similarly, most of the studied samples show negative Eu anomalies (Eu/Eu*=0.88-0.94), barring three samples (Eu/ $Eu^* = KJ5 = 1.07$, KJ1 = 1.09, LB10 = 1.43; Table 3). The majority of the samples show positive La anomalies, negative Ce anomalies, positive Gd anomalies and negative Eu anomalies (Fig. 5). The $(Gd/Yb)_N$ ratio varies from 1.53 to 2.33 with an average of 1.81, whereas the (La/Yb)_N ratio varies from 0.98 to 2.13 (Table 3) with an average value of 1.48.

Table 2 Majo	r (wt %)), trace a	and rare	earth e	lements c	concentr	ations (ppm) fo	or the Ka	ajrahat I	Limesto	ne of Vi	ndhyan	basin, C	Central I	India	
Sample	KJ16	KJ7	KJ5	KJ1	SKJ14	SKJ9	SKJ3	SKJ1	B28	B15	B9	B1	LB20	LB16	LB10	LB6	LB1
SiO ₂	7.28	14.97	8.80	14.35	12.55	12.54	11.21	11.37	13.19	12.95	9.48	9.51	15.42	12.08	1.14	9.22	10.27
TiO ₂	0.05	0.11	0.05	0.08	0.09	0.08	0.08	0.07	0.10	0.11	0.08	0.08	0.13	0.08	0.02	0.10	0.08
Al_2O_3	1.09	2.76	1.24	2.41	2.47	2.35	2.19	2.24	2.79	2.94	2.10	1.80	4.09	2.40	0.23	1.89	2.49
Fe ₂ O ₃	0.88	0.89	0.74	1.36	1.23	0.95	1.29	1.24	1.47	1.65	1.11	1.46	1.35	1.45	1.86	0.60	1.03
MnO	0.01	0.01	0.01	0.03	0.02	0.01	0.02	0.02	0.01	0.01	0.01	0.01	0.02	0.01	0.05	0.01	0.01
CaO	46.56	43.78	48.59	44.08	41.85	44.63	42.37	41.94	39.89	40.54	43.93	41.89	37.2	39.88	30.87	46.70	42.26
MgO	2.39	0.87	0.51	0.71	3.28	1.07	3.35	3.16	4.32	3.84	2.94	4.36	4.96	4.99	19.17	0.80	3.31
Na ₂ O	0.02	0.03	0.02	0.03	0.03	0.03	0.03	0.03	0.04	0.04	0.03	0.03	0.05	0.03	0.02	0.02	0.03
K ₂ O	0.39	0.64	0.34	0.60	0.65	0.58	0.58	0.57	0.81	0.86	0.7	0.68	1.27	0.81	0.14	0.64	0.79
P_2O_5	0.16	0.41	0.12	0.44	0.37	0.19	0.25	0.23	0.50	0.66	0.78	0.38	0.60	0.43	0.03	0.35	0.45
SUM	58.83	64.46	60.41	64.09	62.55	62.43	61.37	60.87	63.12	63.6	61.16	60.19	65.08	62.16	53.54	60.32	60.73
LOI	39.81	35.15	38.67	36.07	37.40	36.80	38.00	42.78	36.94	36.84	38.77	39.04	35.54	37.79	45.72	38.00	37.43
D.R	0.05	0.02	0.01	0.02	0.08	0.02	0.08	0.07	0.11	0.09	0.07	0.10	0.13	0.12	0.62	0.02	0.08
Ва	23.90	30.90	21.10	32.60	31.20	30.90	27.40	28.70	15.50	38.40	13.50	34.80	11.60	31.10	10.10	30.50	32.10
v	19.60	23.60	19.60	22.50	24.50	25.30	22.60	23.90	20.30	28.90	20.20	19.60	35.70	28.40	18.70	28.00	29.40
Sc	2.80	3.80	3.20	3.20	3.40	3.00	2.80	2.80	3.30	4.10	3.30	3.20	5.10	3.60	2.10	3.90	3.90
Co	1.60	2.50	2.60	2.70	2.30	2.40	2.00	2.30	2.50	3.00	2.70	2.20	3.40	2.50	1.60	2.90	2.20
Cu	16.10	53.40	33.50	31.80	9.50	9.40	6.60	6.90	4.20	9.20	3.60	3.60	12.30	9.80	8.60	11.90	8.40
Rb	15.40	27.40	13.90	23.90	27.70	24.10	21.40	23.50	30.00	36.30	26.30	24.30	57.70	31.40	4.30	29.80	32.60
Zn	18.10	21.70	17.70	21.10	23.90	18.30	15.90	17.10	25.10	47.9	18.80	34.00	31.60	24.50	14.30	22.60	20.40
Pb	4.40	5.40	4.70	6.10	4.90	4.90	4.40	4.60	5.50	5.70	4.40	5.60	6.00	4.70	3.50	5.80	4.80
Th	1.40	3.10	1.40	2.20	2.40	2.60	2.10	2.40	3.20	3.10	2.30	2.20	4.20	2.20	0.40	2.70	2.30
U	0.30	0.80	0.40	0.40	0.40	0.60	0.30	0.40	0.60	0.50	1.30	0.50	0.50	0.30	0.50	0.60	0.60
Zr	16.40	35.40	18.60	21.80	24.60	24.90	18.00	21.70	28.60	34.60	21.70	20.70	41.40	25.70	4.40	34.80	33.10
Sr	137.0	139.0	134.0	134.0	154.0	144.0	144.0	145.0	316.0	347.0	426.0	240.0	145.0	126.0	480.0	514.0	417.0
La	6.66	9.90	7.42	8.83	8.46	9.23	7.05	8.41	8.75	9.34	8.11	6.68	14.68	8.89	2.48	8.77	8.85
Ce	12.38	20.95	14.66	17.11	15.51	18.50	14.79	15.62	17.94	18.18	15.21	13.25	28.24	18.12	5.61	15.24	16.07
Pr	1.57	2.80	2.12	2.40	1.93	2.37	1.87	1.91	2.38	2.38	2.03	1.78	3.59	2.46	0.88	1.90	2.07
Nd	4.84	9.75	7.93	8.42	6.13	7.59	6.06	6.74	8.26	8.16	6.73	6.04	12.65	8.86	2.92	5.92	6.51
Sm	0.87	1.97	1.59	1.59	1.12	1.42	1.12	1.12	1.55	1.56	1.24	1.16	2.35	1.67	0.51	1.10	1.20
En	0.17	0.36	0.37	0.37	0.20	0.24	0.20	0.23	0.29	0.30	0.26	0.21	0.42	0.31	0.15	0.20	0.23
Gd	0.79	1.86	1 64	1 59	1.01	1.26	1.00	1 17	1 43	1 51	1 19	1 10	2.26	1 57	0.47	1.03	1 11
Th	0.10	0.27	0.25	0.23	0.13	0.18	0.13	0.14	0.19	0.21	0.16	0.15	0.32	0.21	0.06	0.14	0.15
Dv	0.53	1 47	1 40	1 19	0.68	0.90	0.67	0.77	1.02	1 10	0.83	0.81	1.65	1 11	0.00	0.76	0.76
Но	0.55	0.30	0.28	0.23	0.00	0.18	0.13	0.17	0.21	0.22	0.05	0.01	0.33	0.22	0.06	0.15	0.15
Fr	0.28	0.77	0.20	0.25	0.15	0.10	0.35	0.30	0.55	0.6	0.45	0.44	0.88	0.57	0.15	0.43	0.40
Tm	0.20	0.11	0.10	0.01	0.05	0.47	0.05	0.05	0.05	0.0	0.45	0.17	0.00	0.08	0.15	0.45	0.40
Vh	0.04	0.11	0.10	0.00	0.05	0.07	0.05	0.05	0.00	0.09	0.00	0.07	0.15	0.00	0.02	0.00	0.00
IU Iu	0.23	0.00	0.50	0.49	0.55	0.42	0.51	0.57	0.01	0.04	0.59	0.41	0.78	0.49	0.12	0.40	0.59
Lu ΣPEE	0.04 28 60	51.20	30.00	0.00 13 10	36.00	12 80	33 70	37 10	13 20	14 20	36.90	32 20	68 20	0.00 11 60	13 70	36.10	37.00
	20.00	J1.20	25 70	40.20	34.40	42.00	22 10	25 20	40.60	44.20	24.00	32.30	64 20	41.00	12.00	24.20	26.00
LKEE	27.30 1.20	47.00	2 20	40.50	34.40 1.60	40.00	52.10 1.60	55.20 1.00	40.00 2 60	41.40 2 80	24.80 2.00	2 10	04.20 4 10	41.90 2.70	15.00	34.20 1.00	1.00
LIKEE	1.50	3.00	3.30	2.00	1.00	2.20 10.45	1.00	1.90	2.00	2.00	2.00	2.10	4.10	2.70	10 57	1.90	1.90
LKEE/HKEE	21.00	15.22	10.82	14.39	21.50	18.45	20.06	18.55	15.62	14./9	17.40	14.58	15.66	15.52	18.5/	18.00	18.95

Discussion

Significance of major and trace elements

The studied limestones contain a significant amount of CaO

(Table 2). The primary carbonate phase is mainly calcite, as indicated by the high concentration of CaO and the low percentage of MgO, which is further supported by the minimal dolomitization ratio. The concentration of SiO_2 varies from 1.14 to 14.97%. The relatively higher concentration



Fig. 3 Plot of Post-Archean Australian Shale normalized major elements pattern of Kajrahat Limestone



Fig. 4 Plot of Post-Archean Australian Shale normalized trace elements pattern of Kajrahat Limestone

of SiO₂ may be due to the more continental influx during the deposition of the limestone in shallow marine condition. The low concentrations of other oxides such as TiO₂, Al₂O₃, Fe₂O₃, MnO, Na₂O, K₂O, and P₂O₅ (Table 2) and their positive correlation with SiO₂ indicates that these oxides are associated with a siliciclastic phase that was present in the examined limestones as impurities/detrital admixtures. Primarily Ca is biogenic in origin and inspite of its original distribution is basically behaved as a dilutant for all other constituents. This is evident by its negative correlations with most of the major and trace elements (Table 4). The relation between Al₂O₃ and Fe₂O₃ differentiates between inland and marine limestones by the negative and positive correlation, respectively (Zhang et al. 2017). The studied samples show positive relation between these two oxides (Table 4) which



Fig. 5 Plot of Post-Archean Australian Shale normalized rare earth elements pattern of Kajrahat Limestone

indicate its marine nature. In contrast, Al₂O₃ has a negative correlation with MnO for all the limestones ($r^2 = -0.37$), which indicates that the concentration of MnO is may not be controlled by clay minerals but other input to the limestone geochemistry has a significant role. The P2O5 contents are well correlated with TiO₂ ($r^2=0.72$), Fe₂O₃ ($r^2=0.16$) and $K_2O(r^2=0.78)$ (Table 4), implying that some, maybe most, of the P₂O₅ contents in these limestones may not be biogenic, because TiO₂ and K₂O are principally derived from aluminosilicate clastics and Fe₂O₃ from hydrothermal Fe-Mn-oxyhydroxides (e.g., Murray 1994). All of the major oxides except CaO, MnO, MgO showing positive correlation with each other. So, the geochemical pattern of major oxides reveals that the geochemistry of the present studied limestones is basically managed by the tectonic environment of the basin that may be volcanic activity, topographic elevation and the distance away from the continent.

The trace elements of the studied limestones viewed by their positive correlation with Al_2O_3 TiO₂ (Table 4), which indicates that their concentrations primarily controlled by detrital silisiclastic fractions. The concentrations of the highstrength-field elements (HSFEs; e.g., Th, U, Sc, Zr) and the transitional trace elements (e.g., Co, Cu), in the studied limestones are generally lower by one order of magnitude when compared to the PAAS (Fig. 4). However, the contents of the large-ion-lithophile elements (LILEs; e,g. Ba, Rb, Sr)) in these limestones exhibit large range. The studied limestones reveal that Sr is the most abundant trace element in them. The Sr concentration in sedimentary rocks varies due to various factors that affect Sr in depositional environments under low temperatures. The presence of Ca can make an impact on the distribution of Sr, causing additional Sr to be incorporated into diagenetic carbonate. Additionally, weathering of feldspars, specifically plagioclase can result in

Table 3 Elemental ratios in the Kajrahat Limestone of Vindhyan basin, Central India

Sample	KJ16	KJ7	KJ5	KJ1	SKJ14	SKJ9	SKJ3	SKJ1	B28	B15	B9	B1	LB20	LB16	LB10	LB6	LB1
La/Sc	2.38	2.61	2.32	2.76	2.49	3.08	2.52	3.00	2.65	2.28	2.46	2.09	2.88	2.47	1.18	2.25	2.27
La/Co	4.16	3.96	2.85	3.27	3.68	3.85	3.53	3.66	3.50	3.11	3.00	3.04	4.32	3.56	1.55	3.02	4.02
Th/Sc	0.50	0.82	0.44	0.69	0.71	0.87	0.75	0.86	0.97	0.76	0.70	0.69	0.82	0.61	0.19	0.69	0.59
Th/Co	0.88	1.24	0.54	0.81	1.04	1.08	1.05	1.04	1.28	1.03	0.85	1.00	1.24	0.88	0.25	0.93	1.05
U/Th	0.21	0.26	0.29	0.18	0.17	0.23	0.14	0.17	0.19	0.16	0.57	0.23	0.12	0.14	1.25	0.22	0.26
La/Th	4.76	3.19	5.30	4.01	3.53	3.55	3.36	3.50	2.73	3.01	3.53	3.04	3.50	4.04	6.20	3.25	3.85
Er/Nd	0.06	0.08	0.09	0.07	0.06	0.06	0.06	0.06	0.07	0.07	0.07	0.07	0.07	0.06	0.05	0.07	0.06
(La/Yb) _N	2.13	1.07	0.98	1.33	1.89	1.62	1.68	1.67	1.26	1.27	1.53	1.20	1.39	1.34	1.52	1.62	1.67
(La/Nd) _N	1.16	0.86	0.79	0.88	1.16	1.02	0.98	1.05	0.89	0.96	1.01	0.93	0.98	0.84	0.72	1.25	1.14
(Gd/Yb) _N	2.05	1.63	1.74	1.93	1.82	1.79	1.92	1.88	1.67	1.67	1.82	1.60	1.73	1.91	2.33	1.53	1.70
(Pr/Yb) _N	2.15	1.30	1.19	1.54	1.84	1.78	1.90	1.62	1.47	1.39	1.64	1.37	1.45	1.58	2.31	1.49	1.67
(Nd/Yb) _N	1.84	1.25	1.24	1.50	1.63	1.58	1.71	1.59	1.42	1.32	1.51	1.29	1.42	1.58	2.13	1.30	1.46
(Dy/Yb) _N	1.47	1.38	1.59	1.55	1.31	1.36	1.38	1.32	1.27	1.30	1.35	1.26	1.35	1.44	1.54	1.21	1.24
Eu/Eu*	0.96	0.88	1.07	1.09	0.88	0.84	0.88	0.94	0.91	0.91	1.00	0.87	0.85	0.89	1.43	0.88	0.93
Ce/Ce*	0.77	0.81	0.80	0.77	0.80	0.78	0.80	0.89	0.81	0.81	0.77	0.78	0.86	0.82	0.66	0.79	0.77
Pr/Pr*	1.15	1.11	1.12	1.14	1.13	1.14	1.12	1.06	1.11	1.11	1.14	1.13	1.08	1.10	1.24	1.14	1.15

Sr fractionation (Fairbridge 1972). The negative correlation between Sr and CaO (Table 4) indicates that the Sr and CaO are not genetically connected. Therefore, the enrichment of Sr content may be attributed either from its own minerals or from the siliciclastic materials. The presence of appreciable amount of trace elements like Ba, V, Cu, Rb, Zn, and Zr suggests that either these elements have replaced certain cations or they have formed their own minerals in small quantities. The weak positive and negative correlations of these trace elements with CaO (Table 4), indicate that they are either present in their own minerals or because of the siliciclastic input. Additionally, the presence of siliciclastic impurities in limestone may be responsible for the presence of feldspars and mica, resulting in to the presence of barium. Reeder (1983) suggested that under specific conditions, certain divalent cations like Pb, Cu, Co, Ni and others can substitute for Ca in calcite. This suggests that during the deposition of limestone, Ca may have replaced by Pb, Ni, Co and Cu under appropriate conditions. The Mn-Sr correlation is useful for determining diagenetic changes in limestones (Brand and Veizer 1980). However, there is no significant correlation between Mn and Sr in this study ($r^2 = 0.14$). This indicates that the studied limestones have not experienced major diagenetic changes and may still preserve their original geochemical fingerprints.

Significance of uranium in the marine environment

The redox conditions in seawater influence uranium, causing uranium ions to remain in a higher oxidation state (U^{6+}) in an oxidizing environment. This leads to the formation of a soluble compound called uranyl tricarbonate [UO_2

 $(CO_3)_3^{4-}$]. They maintain a lower oxidation state (U^{4+}) under reducing circumstances, and forming insoluble uranous fluoride that becomes immobilized within marine carbonates (Wignall and Twitchett 1996). Uranium is mobile in aqueous solutions, whereas Th is generally immobile (Anderson et al. 1983; Nozaki et al. 1981; Wright et al. 1984). Within continental margin settings, uranium can undergo fractionation similarly to Ce, leading to its separation from other rare earth elements (REEs) (Whittaker and Kyser 1993). In the studied limestones, the U concentration is relatively low, ranging from 0.3 to 1.3 (Table 5). This contrasts with sediments originating from the oxygen minimum zone (Nath et al. 1997). However, the U concentrations in the studied limestones are comparable to those observed in shallowmarine carbonates and the Kudankulam Limestone which were deposited under oxygen-rich conditions (Madhavaraju and Ramasamy 1999; Table 5). This shows that the studied limestones with a low U concentration are affected by the oxygenation level in the water column. In an oxygen-rich conditions (oxic environment), U is readily leached from the sediments and transferred into the water column. Conversely, under reducing conditions, U is extracted from seawater and precipitates onto the sediments. In some instances, there is no significant reduction of U⁺⁶ to U⁺⁴ observed in anoxic and suboxic environments (Anderson et al. 1989). In this particular case, sedimentary geochemists attempted to use the U/Th ratio as a redox indicator rather than the U concentration (Wright et al. 1984; Jones and Manning 1994). K and Th are present in the detrital clay fraction of marine mudrocks shows close correlation (Myers and Wignall 1987). U also present in the detrital clay fraction but behaves differently than Th, it is also transfer to some extent

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	SiO_2	TiO_2	Al_2O_3	$\mathrm{Fe}_{2}\mathrm{O}_{3}$	MnO	CaO	MgO	Na_2O	K_2O	P_2O_5	Ba	V	Sc	Co	Cu	Rb	Zn	Pb
SiO_2	1.00																	
TiO_2	0.85	1.00																
Al_2O_3	0.91	0.93	1.00															
Fe_2O_3	-0.11	-0.08	0.04	1.00														
MnO	-0.45	-0.47	-0.37	0.56	1.00													
CaO	0.24	60.0	-0.02	-0.85	-0.68	1.00												
MgO	-0.66	-0.49	-0.44	0.70	0.76	-0.88	1.00											
Na_2O	0.71	0.77	0.89	0.41	-0.15	-0.35	-0.09	1.00										
K_2O	0.74	0.88	0.93	0.11	-0.40	-0.11	-0.31	0.88	1.00									
P_2O_5	0.56	0.72	0.70	0.16	-0.38	-0.03	-0.28	0.68	0.78	1.00								
\mathbf{Ba}	0.36	0.24	0.17	-0.16	-0.39	0.41	-0.50	-0.08	0.07	0.03	1.00							
>	0.53	0.68	0.74	-0.03	-0.19	-0.14	-0.17	0.59	0.78	0.43	0.19	1.00						
Sc	0.65	0.85	0.81	-0.15	-0.43	0.06	-0.37	0.66	0.88	0.67	0.11	0.84	1.00					
Co	0.66	0.78	0.74	-0.11	-0.33	0.13	-0.42	0.62	0.74	0.68	0.04	0.65	0.83	1.00				
Cu	0.27	0.08	0.02	-0.40	-0.03	0.36	-0.33	-0.19	-0.19	-0.14	0.14	-0.09	0.13	0.13	1.00			
Rb	0.73	0.89	0.93	0.03	-0.37	-0.09	-0.31	0.84	0.98	0.72	0.07	0.86	0.93	0.79	-0.11	1.00		
Zn	0.40	0.61	0.52	0.37	-0.31	-0.15	-0.10	0.61	0.61	0.59	0.36	0.44	0.61	0.55	-0.15	0.59	1.00	
Pb	0.73	0.79	0.69	-0.10	-0.35	0.24	-0.53	0.54	0.66	0.52	0.37	0.47	0.71	0.75	0.24	0.68	0.63	1.00
Th	0.85	0.97	0.95	-0.07	-0.46	0.05	-0.47	0.83	06.0	0.68	0.13	0.69	0.83	0.78	0.02	0.91	0.55	0.76
Ŋ	0.02	0.21	0.12	-0.18	-0.22	0.07	-0.08	0.10	0.15	0.56	-0.30	-0.10	0.16	0.27	0.01	0.12	-0.03	0.00
Zr	0.75	0.92	0.85	-0.26	-0.55	0.17	-0.51	0.63	0.84	0.64	0.27	0.79	0.93	0.77	0.16	0.88	0.55	0.74
Sr	-0.51	-0.10	-0.27	0.10	0.14	-0.27	0.41	-0.18	-0.09	0.21	-0.22	-0.02	0.00	0.00	-0.38	-0.08	0.08	-0.12
La	0.85	0.87	0.92	-0.22	-0.44	0.16	-0.54	0.75	0.88	0.59	0.09	0.78	0.88	0.81	0.16	0.92	0.41	0.70
Ce	0.88	0.86	0.93	-0.15	-0.42	0.11	-0.50	0.78	0.86	0.57	0.07	0.73	0.84	0.78	0.23	0.89	0.41	0.68
Pr	0.87	0.82	0.90	-0.12	-0.39	0.10	-0.48	0.76	0.82	0.57	0.04	0.68	0.82	0.80	0.31	0.84	0.41	0.68
Nd	0.84	0.76	0.85	-0.06	-0.34	0.07	-0.43	0.74	0.77	0.53	0.00	0.62	0.78	0.80	0.36	0.79	0.40	0.66
Sm	0.82	0.74	0.81	-0.09	-0.37	0.10	-0.45	0.69	0.73	0.51	0.02	0.57	0.77	0.78	0.43	0.74	0.41	0.66
Eu	0.68	0.53	0.62	-0.05	-0.20	0.11	-0.37	0.54	0.53	0.43	-0.07	0.39	0.63	0.74	0.55	0.55	0.30	0.58
Gd	0.79	0.70	0.77	-0.09	-0.34	0.12	-0.45	0.66	0.69	0.50	0.01	0.54	0.75	0.80	0.46	0.71	0.40	0.66
Tb	0.75	0.65	0.71	-0.13	-0.32	0.16	-0.44	0.59	0.62	0.45	0.01	0.49	0.73	0.78	0.53	0.65	0.38	0.65
Dy	0.73	0.63	0.67	-0.16	-0.35	0.20	-0.46	0.55	0.58	0.42	0.02	0.45	0.70	0.77	0.56	0.61	0.37	0.64
Но	0.72	0.62	0.67	-0.17	-0.35	0.19	-0.46	0.55	0.57	0.41	0.00	0.43	0.69	0.76	0.56	0.60	0.35	0.62
Er	0.73	0.66	0.68	-0.17	-0.37	0.20	-0.47	0.57	09.0	0.45	0.02	0.46	0.73	0.80	0.54	0.63	0.40	0.67
Tm	0.72	0.70	0.71	-0.14	-0.40	0.16	-0.44	0.61	0.66	0.47	0.04	0.50	0.77	0.79	0.48	0.68	0.48	0.69
Yb	0.78	0.76	0.78	-0.14	-0.42	0.15	-0.47	0.65	0.71	0.52	0.05	0.54	0.80	0.82	0.46	0.73	0.47	0.72
Lu	0.72	0.69	0.72	-0.12	-0.40	0.15	-0.44	0.63	0.68	0.48	0.02	0.52	0.78	0.81	0.43	0.70	0.50	0.70

Table 4	(continued)																	
	SiO_2	TiO_2	AI_2O_3	$\mathrm{Fe_2O_3}$	MnO	CaO	MgO	Na_2O	K_2O	P_2O_5	Ba	>	Sc	Co	Cu	Rb	Zn	Pb
Σ^{REE}	0.87	0.84	0.91	-0.14	-0.41	0.12	-0.50	0.76	0.83	0.57	0.05	0.70	0.84	0.81	0.29	0.86	0.42	0.69
	Th	U	Zr	Sr	La	Ce	Pr	S PN	im E	ı Gd	Tb	Dy	Но	Er	Tm	Yb	Lu	D REE
SiO_2																		
TiO_2																		
AI_2O_3																		
Fe ₂ O ₃																		
CaU M ₂ O																		
Na_O																		
K,0																		
۔ P,O,																		
ء م Ba																		
>																		
Sc																		
Co																		
Cu																		
Rb																		
Zn																		
Pb																		
Th	1.00																	
D	0.21	1.00																
\mathbf{Zr}	0.91	0.20	1.00															
Sr	-0.15	0.49	0.01	1.00														
La	0.91	0.11	0.88	-0.32	1.00													
Ce	0.91	0.10	0.85	-0.40	0.98	1.00												
Pr	0.86	0.11	0.81	-0.41	0.95	0.99	1.00											
PN	0.81	0.08	0.74	-0.44	0.90	0.95	0.99	1.00										
Sm	0.77	0.10	0.73	-0.44	0.87	0.93	0.97	0.99 1	.00									
Eu	0.56	0.09	0.54	-0.40	0.70	0.76	0.85	0.91 (.93 1.	00								
Gd	0.73	0.08	0.70	-0.44	0.84	0.90	0.95	0.98 (.0 0.	96 1.00								
Tb	0.68	0.09	0.66	-0.42	0.80	0.86	0.92	0.96 (.98 0.	6.0 76	1.00							
Dy	0.65	0.08	0.65	-0.43	0.77	0.83	0.90	0.94 (.97 0.	97 0.95	1.00	1.00						
Но	0.66	0.10	0.64	-0.43	0.77	0.83	0.90	0.94 (.96 0.	96 0.98	0.99	1.00	1.00					
Er	0.68	0.11	0.68	-0.39	0.79	0.84	0.91	0.94 (.97 0.	96 0.98	0.99	1.00	0.99	1.00				

2

REE

in solution as uranyl carbonate complexes (Langmuir 1978). The reducing conditions are favorable for its precipitation which tends to enrich the sediment in authigenic (nondetrital) uranium. The authigenic component of U can be calculated if detrital component is known. The U is assumed to be purely detrital, is 3 in the normal mudstone with minimum Th/U ratio. The concentration of U which is associated with the detrital fraction can thus be estimated by dividing the measured Th content by 3. Authigenic U in the sample is the difference between the actual U measured and the calculated detrital U. Reducing conditions and the presence of a sorbent, which is usually organic matter or phosphate, favoured the presence of U at the sediment-water interface (Kochenov et al. 1977; Holland 1984). Therefore, sediments enriched in authigenic U indicate anoxic depositional conditions that allow both large amounts of organic matter to accumulate and U to be fixed. Ratios of U/Th exceeding 1.25 are indicators of suboxic and anoxic conditions and have been utilized to infer such environments. Additionally, sediments from the Arabian Sea in the oxygen minimum zone exhibit a notably high U/Th ratio (>1.25) (Nath et al. 1997). The studied limestones exhibit a U/Th ratio (0.12–1.25, Table 3) which is lower than the samples deposited under anoxic and suboxic conditions. Nevertheless, the U/Th in these limestones is similar to shallow-marine carbonates found in southern India that were deposited under an oxic environment (Madhavaraju and Ramasamy 1999; Table 5). This strongly shows that the studied limestone was also deposited under oxic conditions. Moreover, the studied limestone samples show minimal variations in both U contents and U/Th ratios. This suggests that there were no significant changes in oxygen levels within the water column during deposition of these shallow-marine limestones.

Source of REE and provenance characteristics

Total REE abundances (ΣREE) of the studied limestones are generally lower when compared to the PAAS. The ΣREE concentrations in studied limestone samples ranging from 13.7 to 68.3 are comparable to Upper Cretaceous shallowmarine carbonates (Madhavaraju and Ramasamy 1999) and modern Arabian Sea carbonates (Nath et al. 1997) (Table 5). The variation in the ΣREE content among different samples is primarily due to differences in the terrigenous sediment content within the limestone samples. This is likely attributed to lower levels of Si and Al and higher CaCO₃ content, suggesting that the ΣREE content is influenced by impurities other than carbonate.

The low Σ REE in certain samples of studied limestones is attributed to the presence of marine carbonate phases. These phases typically have lower rare earth elements (REEs) compared to detrital clays and heavy minerals (Piper 1974; Palmer 1985). In limestone samples, the REE contents are

Table 4 ((continued)	-																	
	Th	n	Zr	Sr	La	Ce	Pr	ΡN	Sm	Eu	Gd	Tb	Dy	Ho	Er	Tm	Yb	Lu	
Tm	0.72	0.10	0.71	-0.36	0.80	0.85	0.91	0.94	0.97	0.93	0.97	0.98	0.98	0.98	0.99	1.00			
Yb	0.79	0.15	0.78	-0.33	0.84	0.89	0.94	0.96	0.98	0.92	0.98	0.98	0.98	0.98	0.98	0.99	1.00		
Lu	0.72	0.08	0.71	-0.36	0.81	0.86	0.91	0.94	0.97	0.93	0.97	0.98	0.98	0.97	0.99	1.00	0.98	1.00	
\sum REE	0.88	0.10	0.83	-0.40	0.97	0.99	1.00	0.98	0.96	0.83	0.94	0.91	0.89	0.89	06.0	06.0	0.94	0.91	1.

Elements ratios	Kajrahat	Kudankulam	Shallow marine	Shallow marine	Arabian Sea	Indian Ocean	Shahabad Formation	Limestone			Sea water
	TIMESTORE	LIIIestone	carbonate	piauoim carbonate ^c	caroonate sediments ^d	Carbonate Sediments ^e	Grey Limestone ^f (GL)	Yellowish Limestone ^f (YL)	Brown Limestone ^f (BL)	Bhima Limestone ^f	
Ce/Ce*	0.66-0.89	0.90 ± 0.06	0.76 ± 0.16	I	0.84 ± 0.06	0.30–0.84 Avg. 0.56	$\begin{array}{c} 0.35 - 0.63 \\ (0.54 \pm 0.1) \end{array}$	$\begin{array}{c} 0.50-0.64 \\ (0.57 \pm 0.05) \end{array}$	$\begin{array}{c} 0.58 - 0.66 \\ (0.62 \pm 0.12) \end{array}$	0.57 ± 0.08	$< 0.1 - 0.4^{g}$
(La/Yb) _N	0.98–2.13	2.69 ± 1.37	1.82 ± 0.46	0.68 ± 0.47	0.85 ± 0.2	0.166 ± 0.953	$\begin{array}{c} 0.52 - 0.84 \\ (0.73 \pm 0.12) \end{array}$	$\begin{array}{c} 0.48 - 0.89 \\ (0.71 \pm 0.15) \end{array}$	0.59-0.86 (0.77 ± 0.12)	0.732 ± 0.12	0.2–0.5 ^h
(La/Nd) _N	0.72-1.25	1.17 ± 0.09	1.22 ± 0.26	1.01 ± 0.20	0.98 ± 0.03	1.04 ± 0.19	$\begin{array}{c} 0.84-\\ 1.35(1.09\pm0.19) \end{array}$	$\begin{array}{c} 0.99 - 1.33 \\ (1.15 \pm 0.14) \end{array}$	$\begin{array}{c} 1.09 - 1.20 \\ (1.23 \pm 0.24) \end{array}$	1.15 ± 0.19	0.8–2.0 ⁱ
N(qA/pN)	1.24–2.13	2.29 ± 1.17	1.51 ± 0.30	0.65 ± 0.39	0.85 ± 0.17	0.83 ± 0.13	$\begin{array}{c} 0.60 - 0.74 \\ (0.67 \pm 0.05) \end{array}$	$\begin{array}{c} 0.48 - 0.72 \\ (0.42 \pm 0.09) \end{array}$	$\begin{array}{c} 0.53 - 0.76 \\ (0.61 \pm 0.11) \end{array}$	0.64 ± 0.08	0.205–0.497 ^k
(Dy/Yb) _N	1.20–1.59	1.59 ± 0.23	1.25 ± 0.17	1.10 ± 0.25	1.12 ± 0.11	1.37 ± 0.05	$\begin{array}{c} 1.15 - 1.51 \\ (1.27 \pm 0.11) \end{array}$	$\begin{array}{c} 1.08 - 1.34 \\ (1.19 \pm 0.11) \end{array}$	$\begin{array}{c} 0.96 - 1.20 \\ (1.07 \pm 0.12) \end{array}$	1.20 ± 0.13	0.8–1.1 ¹
Er/Nd	0.05-0.09	0.06 ± 0.02	0.07 ± 0.02	0.25 ± 0.17	0.11 ± 0.02	I	$\begin{array}{c} 0.12 - 0.19 \\ (0.2 \pm 0.03) \end{array}$	$\begin{array}{c} 0.13 - 0.20 \\ (0.2 \pm 0.02) \end{array}$	$\begin{array}{c} 0.12 - 0.19 \\ (0.1 \pm 0.02) \end{array}$	0.15 ± 0.03	0.27 ⁿ
CaCO ₃ %	42.27–66.53	88±5	75 ± 15	I	51±22	47–87 Avg. 65.3	$46.11-87.91$ (75.8 ± 13.31)	71.79-88.73 (81.9 \pm 6.82)	52.71-64.11 (61.5 ± 10.66)	74.65±12.89	I
Eu/Eu*	0.84–1.43	0.78 ± 0.31	0.58 ± 0.11	I	1.15 ± 0.08	>1	0.95-1.7 (1.28 ± 0.95)	$\begin{array}{c} 1.09 - 3.69 \\ (1.88 \pm 1.009) \end{array}$	$\begin{array}{c} 1.22 - 1.58 \\ (1.41 \pm 0.15) \end{array}$	1.51 ± 0.64	I
REE (ppm)	13.7-68.3	80 ± 40	73 ± 20	3.36 ± 2.55	78 ± 40	I	35.82 ± 16.2	25.55 ± 9.914	35.02 ± 9.79	32.2 ± 13.3	I
U	0.3 - 1.3	0.9 ± 0.5	0.7 ± 0.5	I	6 ± 2	I	1	I	I	1	I
U/Th	0.12-1.25	0.19 ± 0.10	0.22 ± 0.29	I	2.1 ± 0.5	I	I	I	I	I	I
^a Armstrong-A	ltrin et al. 2003	$(n=9); ^{b}Madh;$	avaraju and Ran	nasamy 1999; L ^ɛ	ate Cretaceous (n=8); ^c Mazumd	ar et al. 2003; Late	Neoproterozoic	$(n = 15); ^{d}$ Nath	et al. 1997 (n=	9); ^e Nath et al.

Table 5 Comparison of the average values of the studied Kajrahat Limestone to the shallow and deep water marine carbonate sediments

1992; ^fNagarajan et al. (2011) (n = 18); ^{g-i}Elderfield and Greaves 1982; Piepgras and Jacobsen 1992; ^jDe Baar et al. 1985; and Zhang and Nozaki 1996 respectively; ^k(50 m water depth samples); ^lWebb and Kamber 2000; ^mBau and Dulski 1996; ⁿDe Baar et al. 1988; ^oThis study (Kajrahat Limestone)

typically lower compared to clastic sediments. Seawater contributes lesser amount of REE to the chemical sediments but the samples with non-seawater like pattern exhibits higher REE concentration (Nothdurft et al. 2004) due to the contamination of non-carbonate materials such as silicates, Fe–Mn oxides, phosphates or sulfides during the chemical leaching (Zhao et al. 2009). The higher REE concentrations in clastic sediments primarily result from the presence of silt and clay fractions. This occurs because REEs are readily being incorporated into the clay structure (McLennan 1989). The variation in the REE contents of Kajrahat Limestone may be attributed to the varying amounts of terrigenous sediments present. The lower REE content in some samples may be attributed to REE dilution by carbonate minerals.

The average (La/Yb)_N ratio in the studied limestone samples (0.98-2.13) is similar to the shallow-marine carbonates of southern India, Kudankulam Limestone. However, (La/Yb)_N ratios are lower in the studied limestones than observed in Arabian Sea carbonate sediments, Indian Ocean carbonates and Shahabad Formation Limestone (Table 5). Additionally, the $(La/Yb)_N$ ratio is higher than the commonly assumed average value for terrigenous sediments $[(La/Yb)_N = 1.3;$ Sholkovitz 1990]. The $(La/Yb)_N$ ratios variation in different samples could be attributed to two factors: (1) variations in REEs input from the source area; and (2) potential diagenetic processes leading to remobilization and exchange with interstitial water (Murray et al. 1991), alongside a decreasing trend of the $(La/Yb)_N$ ratio with increasing depth (Worash and Valera 2002). Such diagenetic processes which is responsible for variation of (La/ Yb)_N ratio have been reported in recent, shallow, buried estuaries (Sholkovitz et al. 1989). The present study resembles shallow-marine conditions where REE fractionation should have been low.

REE patterns and (La/Yb)_N ratios can be used to identify the characteristics of source rocks. Additionally, specific trace-element ratios like Th/Sc, La/Sc, and La/Th can provide valuable information about the source rocks characteristics, as these ratios are influenced by the average compositions of the provenance. Th is considered a highly incompatible element, while Sc is somewhat more compatible. During the sedimentation processes, both of these elements are transferred relatively uniformly into terrigenous sediments as they are transported from their source areas and deposited (Taylor and McLennan 1985). The studied limestone samples show a slightly enriched light rare earth elements (LREEs) and flat heavy rare earth elements (HREEs) patterns (Fig. 5). Additionally, they have relatively high average ratios of (La/Yb)_N, Th/Sc, La/Sc and La/Th (1.48, 0.68, 2.45 and 3.78, respectively; Table 3) which are comparable to felsic rocks (Nagarajan et al. 2011). The felsic source rock for terrigenous sediments of the studied samples also confirmed by the positive correlation between Eu/Eu*

and K/Al, Na/Al. These findings suggest that the terrigenous sediments within the shallow-marine Kajrahat Limestone originate predominantly from felsic source rocks.

Europium anomaly

The Eu/Eu* ratio of the studied limestone samples ranges from 0.84 to 1.43, (Table 3). Typically, a dearth of negative Eu anomaly and the presence of a positive Eu anomaly in shale-normalized REE patterns can be attributed to the influence of eolian input (Elderfield 1988) or hydrothermal solutions (Michard et al. 1983; Worash and Valera 2002). Positive Eu anomalies in bulk sediments may arise from a small increase in the primary or detrital feldspar component (Murray et al. 1991). The presence of a positive Eu anomaly in some studied limestone samples (KJ5, KJ1, LB10) (Table 3) suggests that it could be attributed to local feldspar enrichment rather than a regional phenomenon such as eolian input or hydrothermal solutions. The correlation between Eu/Eu* and elemental ratios such as K/Al and Na/ Al provides support for this interpretation, which helps to determine the existence of detrital feldspar in the bulk sediments (Madhavaraju and Lee 2009). Based on the results of this study, a strong positive correlation exists between the K/Al and Na/Al ratios and the Eu/Eu* ($r^2 = 0.73$, $r^2 = 0.88$), respectively. This suggests that the presence of feldspars in these limestones could be the reason behind the observed positive Eu anomaly. This interpretation was strengthened by the high enrichment of Sr in studied samples (Table 2, Fig. 6). Eu is one of the REE shows a changing valency in the near surface environment (Brookins 1989), where Eu^{3+} reduced to Eu²⁺ under extremely reducing conditions. Redox potential of Eu/Eu* in aqueous solutions depends mainly



Fig. 6 Bivariate plot of Eu/Eu* verses Sr for the Kajrahat Limestone

on temperature and to a lesser extent on pressure, pH, and REE speciation (Bau 1991); which explains the positive Eu anomalies typically found in acidic, reducing hydrothermal fluids. Additionally, the presence of negative Eu anomaly suggests the oxic depositional environment.

Cerium anomaly and paleo-redox conditions

Several studies have been conducted the use of the Cerium (Ce) within the marine environments to infer paleoceanographic conditions (Liu et al. 1988; Grandjean and Albarede 1989; German and Elderfield 1990; Nath et al. 1997). The depletion of Ce in oceanic water result in redox-related fluctuations in cerium concentration compared to other REEs (Elderfield 1988; Piepgras and Jacobsen 1992). In modern seawater, a distinctive feature is the lower concentration of Ce in comparison to its neighbouring rare earth elements (REEs). This anomaly is caused by the conversion of trivalent Ce into less soluble tetravalent Ce due to oxidation. Subsequently, these trivalent Ce ions are removed as they bind to suspended particles that sink through the water column, a process known as scavenging (Bolhar et al. 2004). The Ce/Ce* ratios within the examined limestones range from 0.66 to 0.89, with an average of 0.79. The consistent Ce anomalies in these limestones indicate a relatively stable level of oxygen in the bottom water during their deposition. The negative Ce anomalies observed in the studied limestones (Table 5) closely resemble those found in deep-sea carbonates from the Indian Ocean (Nath et al. 1992), Arabian Sea sediments (Nath et al. 1997), and shallow-marine Maastrichtian carbonates from the Cauvery Basin in southern India (Madhavaraju and Ramasamy 1999). The presence of both Ce concentrations and Ce anomalies is likely attributed to differences in terrigenous sediments within the studied limestones, as well as other factors like diagenesis.

Negative Ce anomalies are predominantly found in ocean basins due to absence of Ce in seawater, as Ce⁴⁺are selectively eliminated from the water column (Elderfield and Greaves 1982). The presence of a slight Ce anomaly allows for the measurement and representation of the extent of the La anomalies, which can be observed as a negative anomaly (Zhao et al. 2009). It is best to utilize a bivariate plot to properly analyze and discriminate between Ce and La anomalies (Bau and Dulski 1996; Zhao et al. 2009). The studied limestone samples represented in the discrimination diagram (Pr/ Pr* vs. Ce/Ce*, Fig. 7) indicates that these samples display original negative Ce anomalies. As there is no chemical explanation for the formation of Nd or Pr anomalies, the presence of a true Ce anomaly should result in $Pr/Pr^* > 1$. The range of Pr/Pr^* (1.06–1.24) in the studied samples indicates that the anomalous enrichment of La is likely the only factor contributing to any Ce anomaly. The degree of Ce



Fig. 7 Bivariate plot of Ce/Ce* verses Pr/Pr* for the Kajrahat Limestone

depletion in the samples is a reflection of oxygenation level of the water (Komiya et al. 2008).

The findings of the present study indicate that the Kajrahat Limestone exhibits negative Ce anomalies, indicating that it was formed in an oxygen-rich environment. Furthermore, minimal Ce anomalies variation within these limestones, suggests stable bottom water oxygen levels. The differences in Ce anomalies and Ce concentrations within the limestone samples are likely a result of variations in terrigenous sediments rather than diagenesis.

Conclusions

On the basis of detailed study of the major, trace and rare earth elements (REEs) geochemistry of carbonate rocks from the Palaeoproterozoic Kajrahat Limestone, Vindhyan Supergroup, Central India, the following conclusions were made:

- The studied limestones are calcitic in nature with dominance of CaO. SiO₂ present as second most abundant major oxide. The enrichment of certain trace elements such as Ba, V, Cu, Rb, Zn and Zr may be due to the siliciclastic input.
- 2. The PAAS-normalized REE pattern exhibits a relatively uniform pattern characterized by enriched LREE, a negative Ce anomaly, a negative Eu anomaly and a low U/ Th ratio. Variations in the abundance patterns of both LREE and HREE, as well as the variability observed in shale-normalized elemental ratios, are likely a result of varying contamination from continental material and

early diagenetic process that occurred in the carbonate sediments.

- The REE patterns and ratios, including La/Sc, La/Th, Th/Sc and (La/Yb)_N collectively indicate that the terrigenous sediments within the studied limestones predominantly originated from felsic source rocks.
- 4. The consistent presence of low U content and a low U/ Th ratio indicates that oxic conditions were predominant at the sediment/water interface during the deposition of the Kajrahat Limestone. The presence of a negative Ce anomaly suggests that REEs are directly incorporated from seawater or from pore water in oxygen-rich conditions. This suggests a mixing oftwo-component systems, including terrigenous clay (detrital), within the marine sediments.
- 5. Tectonic environment of the basin played major role for geochemical pattern of the present studied limestones.

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Author contributions AS carried out fieldwork, conception and design of the study, data analysis, and drafting of the original manuscript. BPS reviewed and edited the manuscript and provided overall supervision. SK carried out fieldwork, conception and design of the study, data analysis, and drafting of the original manuscript. MAQ conception and design of the study and drafting of the original manuscript. AP carried out fieldwork, conception and designed the work and contributed in the finalisation of manuscript. SS drafting of the original manuscript. VKS improved and revised the manuscript.

Data availability All the data generated or analyzed during the present study are included in this manuscript.

Declarations

Conflict of interest The authors declare that they have no competing interests.

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