

## Late Pleistocene Sedimentation History of the Shirshov Ridge, Bering Sea

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**Abstract**—The analysis of the lithology, grain-size distribution, clay minerals, and geochemistry of Upper Pleistocene sediments from the submarine Shirshov Ridge (Bering Sea) showed that the main source area was the Yukon–Tanana terrane of Central Alaska. The sedimentary materials were transported by the Yukon River through Beringia up to the shelf break, where they were entrained by a strong northwestward-flowing sea current. The lithological data revealed several pulses of ice-rafted debris deposition, roughly synchronous with Heinrich events, and periods of weaker bottom-current intensity. Based on the geochemical results, we distinguished intervals of an increase in paleoproductivity and extension of the oxygen minimum zone. The results suggest that there were three stages of deposition driven by glacioeustatic sea-level fluctuations and glacial cycles in Alaska.

**Keywords:** bottom sediments, Shirshov Ridge, Yukon River, Alaska, Bering Strait, grain-size analysis, clay minerals, geochemistry, sedimentation, sea-level, glaciation, paleoproductivity

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### INTRODUCTION

The Bering Sea consists of two nearly equal parts, a shallower northern shelf area with a shelf break at about 150 m water depth and a deeper southern part, which are separated by a narrow strip of the continental slope and continental rise (Fig. 1). The submarine Shirshov Ridge is located in the deeper western part of the Bering Sea. It extends almost north–south for about 700 km separating the Komandorsky (to the west) and Aleutian (to the east) basins and has a width of about 200 km in the north and 20 km in the south [1]. The ridge has two topographic highs, which rise above the basin bottom from 2000–2500 m in the north to 1000–2000 m in the south. The water depth ranges along the ridge crest from 233 m in the north to 2800 m in the south. The slope angle varies from 1.5°–4° in the north and up to 10° in the south. The ridge is separated from the continental slope (abreast the Cape Olyutorsky) by a transverse fault and cannot be considered as its geological extension [2].

Nowadays, the northern part of the ridge (to approximately 57°N) constitutes together with the continental shelf and slope an area where terrigenous sedimentation predominates, whereas its southern portion with the deep-sea basins and the Bowers Ridge is dominated by

diatom ooze sediments [3, 4]. It should be noted that two zones of low-silica diatom oozes were encountered in the center of Russia's part of the shelf and southward from St. Matthew Island [1].

Recent studies demonstrated that the river terrigenous flux to the existing sediment budgets of the Bering Sea is 157.1 million tons per year (9/10 of which is supplied by the Yukon River off Alaska), and the coastal abrasion supply is 200–400 million tons per year (dominated by Asian coastal material). It is known that material produced by coastal abrasion is deposited near the coastline. Existing surface water circulation patterns (Fig. 1) indicate that the annual sediment flux from the Bering Sea through the Bering Strait (maximum 58 m deep) to the Chukchi Sea ranges from 112 to 158 million tons [1]. The seasonal ice cover persists in the northern part of the sea for at least five months per year and for seven months in the extreme north [1]. The southern part of the sea is presently almost ice-free. The annual primary production of the basin is estimated as 150–500 g C/m<sup>2</sup> on the continental margin and 50–100 g C/m<sup>2</sup> in the deep sea [5]. In the northern part of the deep Bering Sea, the oxygen minimum zone pres-

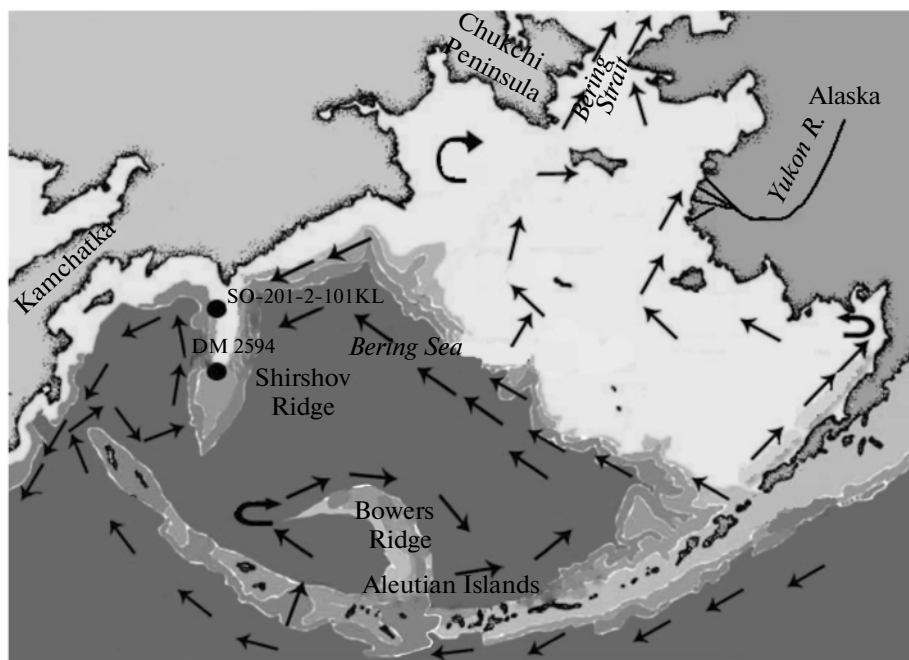


Fig. 1. Location of the cored section at the Shirshov Ridge and present-day surface circulation in the Bering Sea [46].

ently develops at water depths of 380–1200 m, being particularly distinct between 600 and 1100 m [6].

The history of Quaternary sedimentation at the Shirshov Ridge is still poorly understood. The only data available in the literature are those on sediment core DM 2594 [7] from the southern edge of the ridge (Fig. 1). The core records revealed that terrigenous sedimentation occurred at the end of Marine Isotopic Stage (MIS) 3, during MIS 2, and at the beginning (up to 9.07 cal. kyr BP) of MIS 1 and was followed by a period of diatom ooze deposition. The lithological and mineralogical boundaries coincide, indicating a change in provenance from the areas adjoining the coast of the Bering Sea (primarily, Alaska) to presumably the Aleutians [3].

A 1815 cm-long sediment core (S0201-2-101-KL) was recovered from the northern slope of the Shirshov Ridge at a water depth of 607 m during the R/V “Sonne” (Germany) cruise of 2009 carried out within the framework of the German–Russian KALMAR project (Fig. 1). We report the results of the study of the sediment core with emphasis on the history of terrigenous sedimentation. The investigation of other cores recovered during this project is under way, and only preliminary stratigraphic and paleoceanological data are now available [8–11].

#### FACTUAL MATERIAL AND METHODS

Shipboard studies included visual core description (R. Tiedemann and D. Nuernberg), physical property

logs, and color core photographs (R. Tiedemann, D. Nuernberg, L. Max, and J.-R. Riethdorf) [6]. The same authors were responsible for the stratigraphic subdivision of the sediment core. The upper part of the core was used for the accelerator mass spectrometric radiocarbon dating of the planktonic foraminifera. An age model for the cored section [11] was constructed based on AMS radiocarbon dates and correlation with color reflectance logs (color  $b^*$ ),  $\delta^{18}\text{O}$  records, and XRF Ca/Ti scanning data (Table 1).

The core was subsampled for analysis at 15–20-cm intervals in Moscow. Shore-based laboratory studies included grain-size analysis (L.A. Zadorina, Vernadsky Institute of Geochemistry and Analytical Chemistry, Russian Academy of Sciences), bulk sediment (M.A. Levitan and K.V. Syromyatnikov) and clay mineral composition of the  $<2\text{-}\mu\text{m}$  size fraction (V.L. Luksha), chemical analysis by X-ray fluorescence (I.A. Roshchina), and correlation and factor analysis of data on the grain size, mineralogy, and chemical composition of sediments (T.G. Kuzmina).

In wet-sieve analysis, the boundaries between gravel, sand, silt, and clay were placed at values of 2, 0.063, and 0.002 mm, respectively. The clay minerals were analyzed on a DRON-3M diffractometer with a copper anode, using the technique described by Biscaye [12]. Bottom sediments used in XRF analysis were not washed to remove sea salt, following the procedure described in detail by I.A. Roshchina [13]. A total of 90 sediment samples were analyzed.

## RESULTS AND DISCUSSION

**Lithology.** Based on the component analysis data, all the samples used in this study are terrigenous sediments. The grain-size results are presented in Table 2. Based on these data, three major types of lithologies were recognized: dominant, minor, and rare. Dominant lithologies are silts and clayey silts containing less than 10% of the >0.063-mm size fraction. Minor lithologies are silts with some amounts of sand (10–16% of the >0.063-mm size fraction). Rare lithologies are silty sands, mictites, and sandy–clayey silts, in which the >0.063-mm size fraction is higher than 16% reaching a maximum value of 60% (1675–1570 cm depth interval). Interestingly, almost all maxima in the distribution of this fraction are reflected in magnetic susceptibility values (Fig. 2), suggesting that at least part of this fraction is represented by ferrimagnetic particles. A few peaks in the distribution of the sand fraction (1674–1675, 1654–1655, 1494–1495 cm) show no correlation with the susceptibility maxima. This is indicative of a high proportion of coarser gravel material.

The component analysis of bulk sediments shows that light-colored and clay minerals dominate the sediment (27–30% each), with 11–18% each of colored minerals and rock fragments. Opaque minerals, volcanic glasses, remains of siliceous (diatoms and siliceous sponge spicules) and carbonate (bivalves, foraminifera, and calcareous nannofossils) organisms occur in trace amounts up to a few percent. Some grains contain fragments of radiolarian tests and authigenic pyrite occurring either as isolated grains or framboids within diatom frustules. The volcanic glasses in the sediments are dominated by green varieties above 920 cm and brown varieties below this level. Except for the above-mentioned interval of abundant sand fractions, all the maxima in the >0.063-mm size fraction can be accounted for by the contribution of ice-rafted material. The samples corresponding to these maxima contain well-rounded sand-sized quartz grains, which may indicate their possible transport by one-year ice.

Diagrams showing the relationship between the clay and sand versus silt fractions are of certain interest (Fig. 3). Some samples contain more than 70% silt, irrespective of the proportion of sand (Fig. 3a). Nevertheless, a negative correlation between the fractions is quite evident within the entire sample set ( $R^2 = 0.7263$ ). These relations may indicate different mechanisms of transportation and deposition of both fractions. For much of the section, the sand-sized terrigenous particles may have been rafted by sea ice (occasionally icebergs?), but the silts must have been transported by rivers. However, the silt and sand fractions of samples from the sandy interval also show a negative correlation.

**Table 1.** The age model of sediment core SO201-2-101KL, modified after [11]

Depth, cm	Calendar age, kyr BP	Analytical technique	Marine isotopic stage (MIS)
4	12.9	1	2
10.5	13.56	2	2
41	14.6	3	2
67	15.4	3	2
90.5	17.25	2	2
190.5	25.74	2	2
234	28.6	3	3
249	30.3	3	3
260.5	32.0	2	3
274	33.5	1	3
280	35.1	1	3
284	35.7	1	3
302	36.9	1	3
349	39.7	1	3
387	43.1	1	3
454	46.9	1	3
514	51.6	1	3
650	56.6	1	3
685	57.8	1	3
723	59.7	1	3
799	64.1	1	4
923	71.7	1	4
1004	76.4	3	5
1023	77.8	3	5
1092	84.4	1	5
1142	89.1	1	5
1236	94.5	3	5
1301	103.5	3	5
1526	116.0	4	5
1591	126.0	4	5
1611	129.0	4	5
1631	135.0	4	6
1646	140.0	4	6
1762	150.0	4	6
1816	156.0	4	6

Note: 1, age determined by correlation with reference sediment column SO201-2-85KL using XRF Ca/Ti core scanning data; 2, AMS  $^{14}\text{C}$ ; 3, correlation of the  $b^*$  values determined from color reflectance in reference column NGRIP; and 4, correlation of  $\delta^{18}\text{O}$  values with those in reference column LR04.

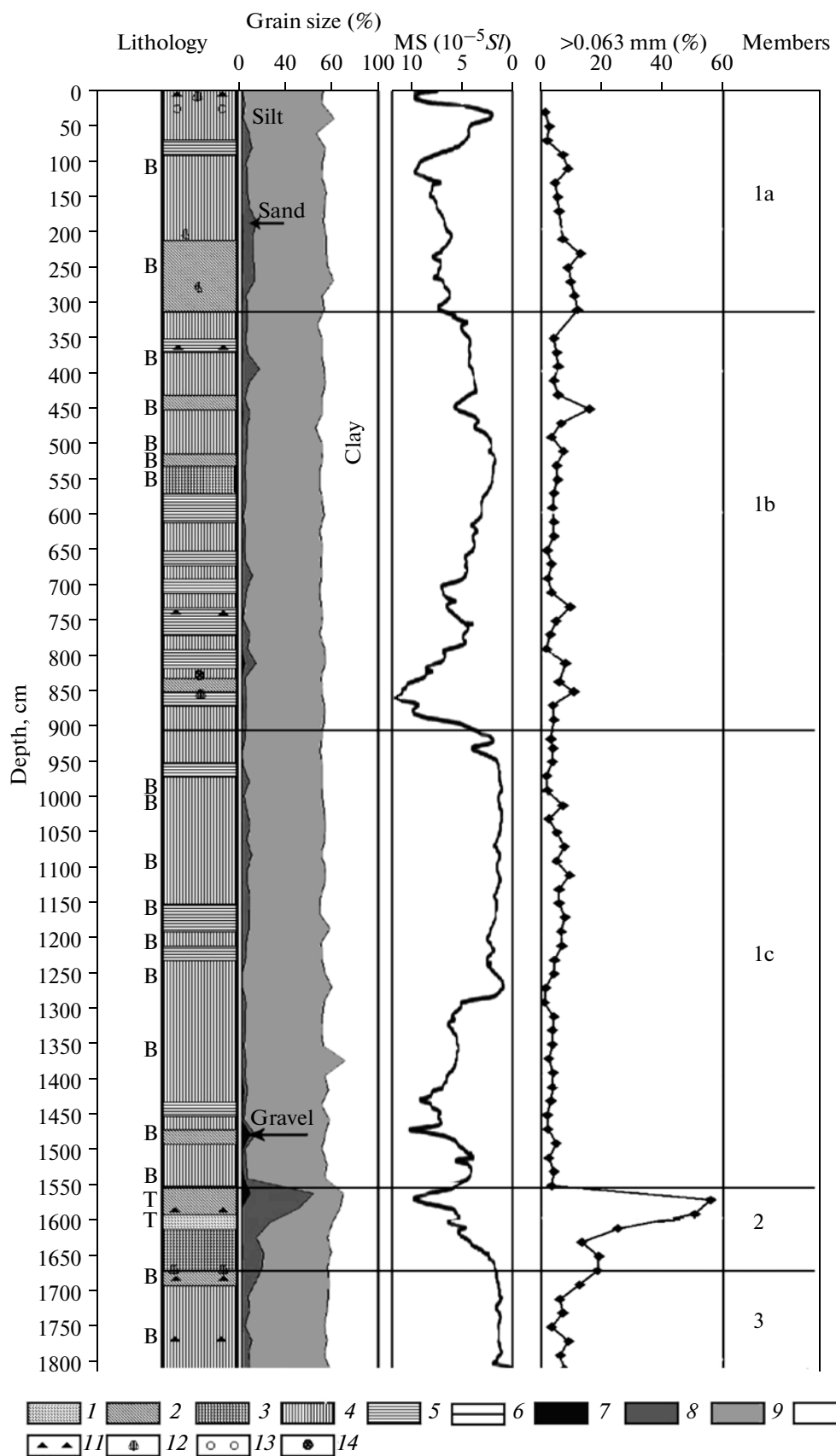
**Table 2.** Grain-size determinations in bottom sediments (fraction size in mm, fraction content in wt %)

Interval, cm	>2	2–1	1–0.5	0.5–0.25	0.25–0.125	0.125–0.063	0.063–0.01	0.01–0.002	<0.002	Lithotype
34–35	0.02	0.01	0.01	0.04	0.17	0.75	35.47	35.63	27.91	Silt
54–55	0.04	0.13	0.04	0.08	0.71	1.38	38.29	29.93	29.43	Silt
74–76	–	–	0.05	0.08	0.55	1.04	41.26	38.21	18.8	Silt
94–95	–	0.01	0.03	1.05	3.53	2.16	30.56	28.06	34.60	Clayey silt
114–115	0.13	0.03	0.13	0.19	3.83	4.46	32.24	32.46	26.53	Silt
134–135	0.06	0.01	0.09	0.14	1.55	2.55	33.94	33.25	28.41	Silt
154–155	–	0.03	0.07	0.14	2.36	2.60	33.04	32.69	29.06	Silt
174–175	–	0.07	0.03	0.14	2.50	2.92	36.77	32.01	25.56	Silt
214–215	–	0.51	0.11	0.16	2.50	3.62	34.08	31.76	27.26	Silt
234–235	0.39	0.10	0.10	0.26	6.18	5.99	26.24	32.62	28.13	Silt with sand
254–255	1.40	0.02	0.04	0.19	3.22	5.15	35.09	29.43	25.47	Silt with sand
274–275	0.65	0.11	0.12	0.34	4.87	4.20	33.13	30.91	25.67	Silt with sand
294–295	0.31	0.01	0.08	0.23	4.36	6.17	33.51	31.27	24.06	Silt with sand
314–315	0.12	0.15	0.15	0.27	5.08	6.16	35.89	33.08	19.10	Silt with sand
354–355	0.35	0.13	0.13	0.17	1.40	2.14	37.22	29.84	28.62	Silt
374–375	0.10	0.10	0.10	0.13	2.46	2.03	39.39	28.03	27.66	Silt
394–395	–	0.25	0.04	0.17	2.08	2.95	38.09	23.16	33.26	Clayey silt
414–415	–	0.04	0.04	0.09	0.94	2.69	32.99	33.89	29.33	Silt
434–435	0.41	0.22	0.10	0.16	1.05	3.84	32.55	32.77	28.91	Silt
454–455	–	0.34	0.17	0.29	7.33	7.46	29.96	26.80	27.64	Silt with sand
474–475	–	0.18	0.22	0.26	1.83	3.95	33.50	33.72	26.34	Silt
494–495	–	0.04	0.06	0.08	0.50	2.50	34.93	32.51	29.38	Silt
514–515	0.67	0.29	0.24	0.58	2.06	3.93	32.57	30.50	29.16	Silt
534–535	–	0.05	0.09	0.38	0.66	3.61	21.75	38.51	34.95	Clayey silt
554–555	–	0.02	0.05	0.40	0.94	3.67	32.97	32.37	29.39	Silt
574–575	–	0.11	0.08	0.27	0.65	2.89	37.23	30.21	28.57	Silt
594–595	–	0.08	0.04	0.08	0.65	2.59	33.90	31.55	31.11	Clayey silt
614–615	–	–	0.01	0.21	1.54	2.28	39.91	24.51	31.53	Clayey silt
634–635	–	–	0.06	0.08	1.31	2.39	35.92	31.87	28.37	Silt
654–655	–	0.01	0.02	0.05	0.33	1.35	36.23	34.51	27.51	Silt
674–675	–	–	0.05	0.20	0.66	2.16	28.97	36.51	31.46	Clayey silt
694–695	0.47	–	0.06	0.08	0.67	1.24	36.54	31.44	29.5	Silt
714–715	–	0.05	0.01	0.17	1.29	1.71	34.28	32.25	30.23	Clayey silt
734–735	–	–	0.09	0.32	4.22	4.59	34.59	26.31	29.89	Silt
754–755	–	0.19	0.13	0.15	2.00	2.3	33.60	30.39	31.25	Clayey silt
774–775	–	–	0.04	0.07	1.00	1.58	35.85	31.04	30.41	Clayey silt
794–795	0.37	0.09	0.08	0.06	0.31	0.97	36.28	33.27	28.57	Silt
814–815	–	0.40	0.24	0.24	4.05	2.94	23.65	37.59	30.88	Clayey silt
841–842	–	–	0.05	0.23	1.82	3.58	34.36	32.71	27.25	Silt
854–855	2.53	0.55	2.09	0.23	3.05	4.52	29.41	31.44	26.19	Silt with sand
874–875	–	0.16	0.10	0.14	1.89	1.27	34.73	31.41	30.24	Clayey silt
894–895	–	–	0.13	0.18	1.24	2.4	34.26	32.28	29.51	Silt
921–922	0.87	–	0.18	0.11	0.73	1.9	36.70	32.11	27.4	Silt
934–935	0.09	–	0.03	0.10	1.05	2.55	35.23	33.28	27.68	Silt
954–955	–	–	0.06	0.09	0.94	2.34	38.00	28.65	29.92	Silt
974–975	–	–	0.04	0.07	0.36	1.01	35.76	32.24	30.52	Clayey silt
994–995	–	–	0.04	0.05	0.49	1.17	36.16	33.29	28.81	Silt
1016–1017	–	0.12	0.67	0.36	2.57	3.14	33.31	31.05	28.79	Silt

Table 2. (Contd.)

Interval, cm	>2	2-1	1-0.5	0.5-0.25	0.25-0.125	0.125-0.063	0.063-0.01	0.01-0.002	<0.002	Lithotype
1034-1035	—	—	0.03	0.07	0.67	1.4	34.48	34.70	28.66	Silt
1054-1055	—	—	0.07	0.19	1.42	3.14	37.48	30.18	27.51	Silt
1074-1075	—	—	0.19	0.22	2.32	4.64	37.70	28.43	26.51	Silt
1094-1095	—	0.10	0.11	0.14	1.4	3.06	41.03	27.51	26.65	Silt
1114-1115	—	—	0.20	0.14	3.44	5.18	32.73	28.64	29.68	Silt
1134-1135	0.02	0.04	0.10	0.11	2.23	3.04	35.32	33.27	25.86	Silt
1154-1155	—	0.33	0.13	0.13	1.94	3.01	38.01	29.73	26.72	Silt
1174-1175	—	0.25	0.08	0.17	2.03	5.06	32.79	29.06	30.55	Clayey silt
1194-1195	—	—	0.12	0.12	1.94	4.25	35.50	27.00	31.07	Clayey silt
1214-1215	—	—	0.08	0.12	1.12	5.22	35.38	35.54	22.54	Silt
1234-1235	—	0.03	0.03	0.06	0.71	3.43	35.44	30.06	30.24	Clayey silt
1254-1255	—	0.12	0.03	0.06	0.90	2.7	35.96	31.46	28.77	Silt
1274-1275	—	—	0.06	0.07	0.22	0.87	37.67	34.50	26.62	Silt
1294-1295	0.12	—	0.08	0.08	0.15	0.46	42.62	35.53	20.96	Silt
1314-1315	—	0.02	0.06	0.09	0.90	2.81	36.89	31.44	27.79	Silt
1334-1335	—	0.01	0.02	0.08	1.48	1.94	36.88	30.97	28.63	Silt
1354-1355	0.08	0.08	0.12	0.15	0.71	2.3	33.86	33.47	29.22	Silt
1374-1375	—	0.03	0.06	0.10	0.58	1.55	35.26	34.46	27.95	Silt
1394-1395	0.05	0.09	0.19	0.42	1.35	1.68	51.63	35.62	8.96	Silt
1414-1415	0.04	0.31	0.09	0.22	0.84	1.95	37.14	33.14	26.26	Silt
1434-1435	1.96	0.33	0.24	0.29	1.0	1.2	35.53	35.72	23.72	Silt
1454-1455	1.28	0.04	0.24	0.20	0.52	0.72	33.38	33.63	30.0	Clayey silt
1474-1475	—	0.20	0.20	0.20	0.72	0.72	39.25	35.64	23.07	Silt
1494-1495	7.62	0.82	0.45	0.41	1.03	1.94	31.23	30.12	26.37	Silt with sand
1514-1515	0.78	0.16	0.20	0.33	0.90	0.61	34.29	33.88	28.84	Silt
1534-1535	0.39	0.24	0.58	0.34	0.83	1.85	39.18	31.7	24.89	Silt
1554-1555	2.28	0.11	0.11	0.42	0.69	1.91	36.06	32.08	26.34	Silt
1574-1575	7.75	1.09	0.76	1.15	32.26	20.52	14.68	11.3	10.48	Silt with sand
1594-1595	—	0.25	0.32	1.08	36.61	12.48	20.01	16.09	13.17	Silt with sand
1614-1615	—	0.05	0.14	0.34	18.34	6.03	32.77	22.25	20.08	Mictite
1634-1635	—	0.45	0.54	0.49	4.79	7.07	36.76	25.58	24.33	Silt and sand
1654-1655	0.29	0.06	0.23	0.69	11.0	6.77	31.09	28.55	21.31	Silt and sand
1674-1675	—	0.10	0.10	0.34	15.27	2.5	32.50	24.91	24.28	Silt and sand
1694-1695	0.16	0.16	0.32	0.16	2.26	9.69	38.29	25.69	23.26	Silt with sand
1714-1715	—	0.05	0.14	0.24	1.34	4.20	38.47	31.89	23.68	Silt
1734-1735	—	0.04	0.09	0.13	1.45	5.26	37.82	29.91	25.30	Silt
1754-1755	—	—	0.10	0.10	0.50	2.48	39.15	30.72	26.96	Silt
1774-1775	—	0.19	0.24	0.94	2.30	5.17	35.65	30.67	24.84	Silt
1794-1795	—	0.05	0.09	0.19	1.59	4.26	36.27	30.55	26.99	Silt
1814-1815	—	0.12	0.09	0.12	1.28	5.72	40.56	29.71	22.41	Silt

Note: Blank entry denotes the absence of the fraction.



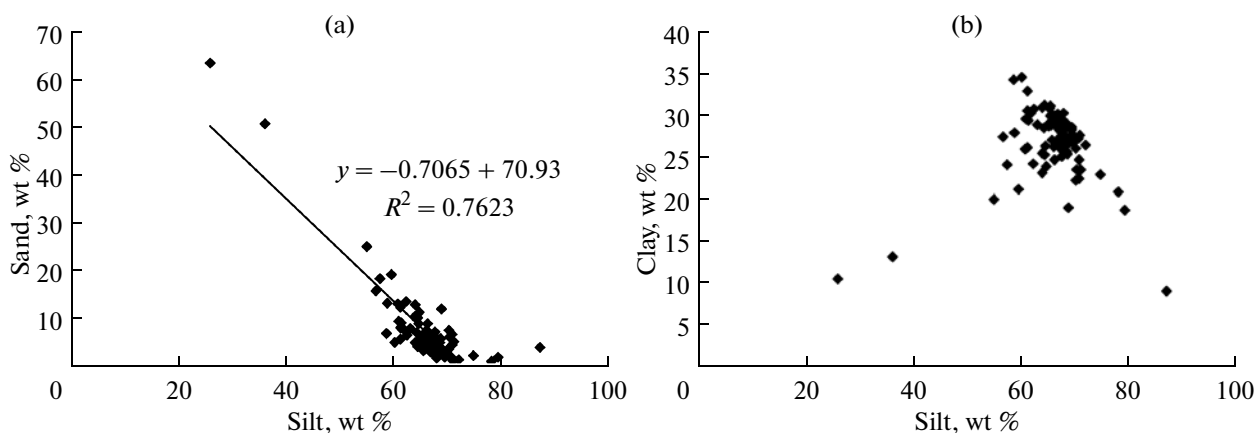


Fig. 3. Relationship between grain-size fractions in the sediments.

The relationship between silt and clay is of interest as well (Fig. 3b). It is evident that a few samples taken entirely from the sandy interval show a positive correlation between clay and silt, whereas the majority of other samples show a negative correlation. The negative correlation likely indicates washing out of the fine-grained clay fraction by bottom currents. Similar situations were also described on the slopes of the Lomonosov Ridge in the central Arctic Ocean [14] and the continental slope of the Kara Sea [15]. Variations in the silt/clay ratio can potentially be used to provide a semiquantitative estimate of fluctuations of the bottom currents. Most of the samples have a silt/clay ratio of 2.0–3.0; values higher than 3.0 suggest an increase in bottom current velocity, which is characteristic of layers (from bottom to top) at 74, 314, 1214, 1294, 1394, 1434, 1474, and 1814 cm. Silt/clay ratios of 1.0–2.0 correspond to a decrease in bottom current velocity at 394, 534, and 814 cm. Therefore, it can be concluded that moderate current velocities dominated during this time interval.

Based on the available data, it can be concluded that the entire section is a monotonous sequence of terrigenous sediments. The uppermost 3 m of the section is dominated by green, olive-grey, and greenish-grey sediments. From 3 to 9.1 m, the sediment is mostly dark olive-grey and grades to dark grey below this level to 15.7 m. The sandy interval contains black sandy turbidites (1594–1574 cm) with abundant mica and gravel clasts within the basal layers, as well as a minor debrite layer consisting of coarse grey detrital sand (1675–1674 cm) with abundant bivalve shell fragments. This sandy interval is underlain by finer grained dark grey and dark olive-grey sediments.

This monotonous sequence contains sporadic small (up to 1 cm) rock fragments and calcitic bivalve debris (Fig. 2). At certain levels (5 and 935 cm), elevated diatom contents were observed in the sediments. The layer at 829 cm contains a small lens of light grey volcanic ash, less than 1 cm thick. As was previously noted, this monotonous section is only disturbed by the sandy interval (1675–1570 cm) containing a single layer of sandy turbidites and laminae of sands, mictites, and sandy silts, up to 30 cm thick, probably representing grain-flow deposits or debrites. The same silts occur above and below this interval.

Minor bioturbation structures are common throughout the section; horizontal lamination and normal graded bedding were observed in the sandy interval.

Based on the above data, the section was subdivided into three lithostratigraphic subunits (from top to bottom): member 1 (0–1570 cm) comprising sub-members 1a (0–310 cm), 1b (310–910 cm), and 1c (910–1570 cm); member 2 (1570–1675 cm); and member 3 (1675–1815 cm, this was not penetrated completely) (Fig. 2).

**Clay minerals.** Since the cored section consists predominantly of silts, fine-grained non-clay detrital minerals become common in the clay fraction. These are mainly quartz, plagioclase, and hornblende grains.

Diocahedral finely dispersed micaceous clay minerals ( $d_{060} = 1.49\text{--}1.50 \text{ \AA}$ ) are the dominant mineral phases. The X-ray diffraction data proved most of these minerals to be illitic materials with Al and Fe cations in octahedral positions. More hydrated phases, such as hydromica with up to 10–15% expandable layers are less common. Moreover, the diffractograms of the gly-

Fig. 2. Lithology and grain size composition of core SO201-2-101-KL. MS is magnetic susceptibility. Symbols: (1) mictite, (2) silt with sand, (3) sandy-clayey silt, (4) sandy silt, (5) clayey silt, (6) lithologic boundary, (7) gravel, (8) sand, (9) silt, (10) clay, (11) coarse IRD, (12) bivalve shells and detritus, (13) diatoms, and (14) volcanic ash. B is bioturbation, and T is turbidites.

erolated samples often show a weak reflection typical of minerals containing up to 25–30% smectite layers. These seem to be mixed-layer illite–smectite formed by the intense supergene alteration of micaceous minerals. It should be noted that the number of samples containing such mixed-layer minerals is greater in the lower part of the column.

Chlorite is another important component of the clay material. The X-ray diffraction data show that the chlorite is trioctahedral ferromagnesian and well crystallized in most samples. The first-order chlorite reflection at 13.8–13.9 Å usually increases in intensity after calcining.

The fine clay fraction incorporates so-called expandable clay minerals, including smectites and mixed-layer smectite–illite with the predominance of smectite interlayers. Their joint consideration is reasonable, because the glycerol-saturated samples of these minerals exhibit a similar first-order reflection at 17–18 Å, and the proportions of different components in the mixed-layer materials are estimated from the 9–10 Å peaks [16]. The interlayer cations are mostly divalent ( $Mg > Ca$ ), and only a few samples from the uppermost three meters of the section contain smectite layers with monovalent Na. The great majority of the expandable components is attributed to disordered mixed-layer minerals with 70–90% smectite interlayers.

Table 3 shows the results of the semiquantitative analysis of clay minerals in the  $<2 \mu m$  size fraction. Four groups of clay minerals were identified: illites and hydromicas (called for brevity illites), chlorites, smectites and mixed-layer minerals (called for brevity smectites), and kaolinites.

Qualitative characterization of clay minerals identified in the cored section is given below. The basal portion of the section (from the base to approximately 1250 cm) exhibits little variations in the clay minerals. The micaceous minerals are dominated by aluminous low-Fe illite (the intensity ratio of the second- and first-order reflections,  $I_{002}/I_{001}$ , is 0.30–0.34). Mixed-layer illite–smectite minerals with a ratio of 0.70–0.75 : 0.25–0.30 are also abundant. The expandable minerals (with dominant smectite interlayers) contain a series of labile structures with up to 70–100% smectite interlayers. The sand-rich interval 1550–1600 cm is characterized by a higher content of ferromagnesian chlorite and a lower content of mixed-layer minerals.

The overlying member (1114–1234 cm) of homogeneous grayish green clayey–silty sediments is characterized by the highest fraction of expandable minerals (up to 32–35%), which are primarily mixed-layers with 70–90% smectite interlayers.

Upsection (up to about 755 cm), the samples exhibit layer-to-layer variations in the composition and proportions of phyllosilicates. An interval of abundant smectite structures is distinguished between 934 and 974 cm, and the micaceous components contain a significant amount of hydromica.

The uppermost part of the section (above 755 cm) exhibits a sharp increase in the amount of illites; they contain more iron ( $I_{002}/I_{001} = 0.27–0.29$ ), and the percentage of hydromica remains about constant throughout the interval. As was previously noted, smectites from the uppermost 3-m interval are characterized by the presence of interlayer Na cations.

We distinguished two major assemblages of clay minerals (minerals are in order of increasing content): (1) kaolinite–chlorite–smectite–illite and (2) kaolinite–smectite–chlorite–illite. In the first assemblage, the illite/smectite and chlorite/smectite ratios are lower than those in the second assemblage. Eight intervals were identified on the basis of clay mineral assemblages in the cored section. The intervals 455–675 and 1095–1355 cm are dominated by assemblage I; 34–275, 675–755, and 1355–1655 cm are dominated by assemblage II; and 275–455, 755–1095, and 1655–1815 cm are characterized by fine interlayering of both assemblages.

The distribution of clay minerals and variations in proportions of minerals and mineral assemblages from base to top in the cored section are shown in Fig. 4. In the eighth interval (1815–1655 cm), illite content decreases cyclically upward, while the proportions of the remaining minerals fluctuate around constant values. The seventh interval (1655–1355 cm) exhibits strong variations in the proportions of all clay minerals, and chlorite is generally more abundant than in the previous interval. The sixth interval (1355–1095 cm) is characterized by the high average smectite content. Illite is present in small amounts and shows an upward decrease. Chlorite and kaolinite are present in small but fairly constant amounts. In the fifth interval (1095–755 cm), the contents of all clay minerals are almost constant, but smectite and chlorite appear in more variable proportions than in the overlying sediments. In the fourth interval (755–675 cm), illite content remains constant but slightly higher as compared to the fifth interval, whereas chlorite and kaolinite tend to decrease and smectite increases. In the third interval (675–455 cm), the contents of all clay minerals are constant, and smectite is always abundant. Illite content decreases slightly upward with a distinct decrease in the illite/chlorite ratio. In the second interval (455–275 cm), illite and chlorite contents are constant, with a decrease in smectite and increase in kaolinite contents. The first interval (275–34 cm) shows high illite (especially in the last three samples), chlorite, and kaolinite and low smectite contents.



Besides the boundaries between the assemblages and fine interlayering intervals, additional boundaries can be distinguished on the basis of variations in the contents and ratios of different clay minerals. For example, boundaries were placed at 75, 365, 890, 1175, 1315, 1425, 1540, and 1750 cm (Fig. 4).

As can be seen in Fig. 4, the distribution of the assemblages exhibits cyclic variations, making two complete cycles (34–670 cm and 670–1350 cm) and one incomplete cycle (from 1350 cm to the bottom of the hole). Each complete cycle shows the following succession (from bottom to top): kaolinite–chlorite–smectite–illite assemblage, fine interlayering of both assemblages, and kaolinite–smectite–chlorite–illite assemblage.

A comparison of the distribution of lithostratigraphic units (members and sub-members) and clay mineral assemblages revealed that sub-member 1a corresponds mainly to the kaolinite–smectite–chlorite–illite assemblage, in which the proportions of the major groups of clay minerals are fairly similar to their average values in the modern sediments of the Yukon Delta and Arctic continental margin of Alaska (which are also related to Yukon sediment supply [17]) (Table 4). The composition of clay minerals in the middle course of the Yukon River is different from that of the Yukon Delta in a sharp decrease in smectite and an increase in chlorite components [18]. Sub-member 1b is dominated by the kaolinite–chlorite–smectite–illite assemblage, whereas sub-member 1c exhibits both the dominance of the assemblages and their fine interlayering. Sub-member 2 is dominantly kaolinite–smectite–chlorite–illite, and sub-member 3 shows interlayering of both assemblages.

The observed ratios between the minerals may reflect climatic and hydrodynamic conditions, as well as their sources. It is well known in the Arctic Ocean, e.g., on the continental margin of the Laptev Sea, that sediments of cold stadial periods have higher illite content, and those of warm periods have higher smectite content [19]. The same results were reported from the southeastern part of the Bering Sea [20]. The data in Table 3 and Fig. 4 show that the influence of climate on clay minerals in the cored section (based on the high illite and low smectite contents and elevated illite/chlorite ratio) becomes relatively clear in the three uppermost samples, i.e., at 34–76 cm (deglacial sediments, Table 1). The second interval with the elevated illite/chlorite ratio is found between 460 and 650 cm. As previously noted, high illite content was observed in the upper 755 cm of the section.

There is abundant evidence on the effect of the settling velocity of sediment particle on the proportions of clay minerals [21]. Considering the above observations

on the relations between bottom current velocity and clay fraction content, a strong hydrodynamic effect might be expected to influence the ratios of clay minerals and the content of the <2- $\mu\text{m}$  size fraction; e.g., this fraction should correlate positively with smectite and negatively with illite. The correlation matrix (Table 5) indicates the presence or absence of correlations between the four groups of minerals and the content of the <2- $\mu\text{m}$  size fraction. As can be seen, the only significant correlation is that for chlorite ( $r = -0.318$ ). Therefore, the hydrodynamic factor is not dominant.

Table 5 also shows that the clay minerals can be categorized into three groups based on the correlation coefficients. The first group is illite, which exhibits a negative correlation with, primarily, smectite and chlorite and no correlation with kaolinite; the second group is chlorite and kaolinite; and the third group is smectite, which has a strong negative correlation with all other clay minerals. Generally, none of the clay minerals shows any positive correlation with other minerals. These results can be explained by the wide development of Early Paleozoic greenstone facies metamorphic rocks in Central Alaska; they are dominated by quartz–muscovite–chlorite schists locally intercalated with mafic metavolcanic rocks [22]. Chlorite and mica hydration products are obviously derived from this source. The Arctic climatic conditions of Alaska, which seem to have persisted since the Late Miocene, could also be important factor of illite enrichment. Smectite could also be supplied by soil erosion and weathering of Late Cenozoic (6–0 Ma) mafic volcanics, which are abundant along the western Alaskan margin and on the Bering Sea shelf [23]. An additional contribution of volcanic material from Kamchatka and even from the Aleutian Arc to the Shirshov Ridge area has to be taken into account. The reliable recognition of sources requires special methods and is beyond the scope of this study. Kaolinite is most likely derived from the Carboniferous sedimentary rocks hosting coal deposits in the Yukon–Tanana Upland [24].

Our results suggest that there are at least two groups of clay minerals, the old Paleozoic and young Pleistocene. The first group includes hydromica, partly illite, chlorite, kaolinite, and part of smectite. They represent the composition of clay minerals from the Lower Paleozoic shales of Central Alaska, which were metamorphosed to the greenstone facies, as well as from other Paleozoic rocks of this area. The average contents of these minerals in the <2- $\mu\text{m}$  size fraction from the Pleistocene rocks considered here is estimated as follows: 42–46% illite and hydromica, 20–25% chlorite, 7–10% kaolinite, and, probably, up to 5% smectite. The young clay minerals are part of illites (averaging 10%), mixed-layer minerals, and part of smectites (totaling about 13–15%). The young illites formed largely under

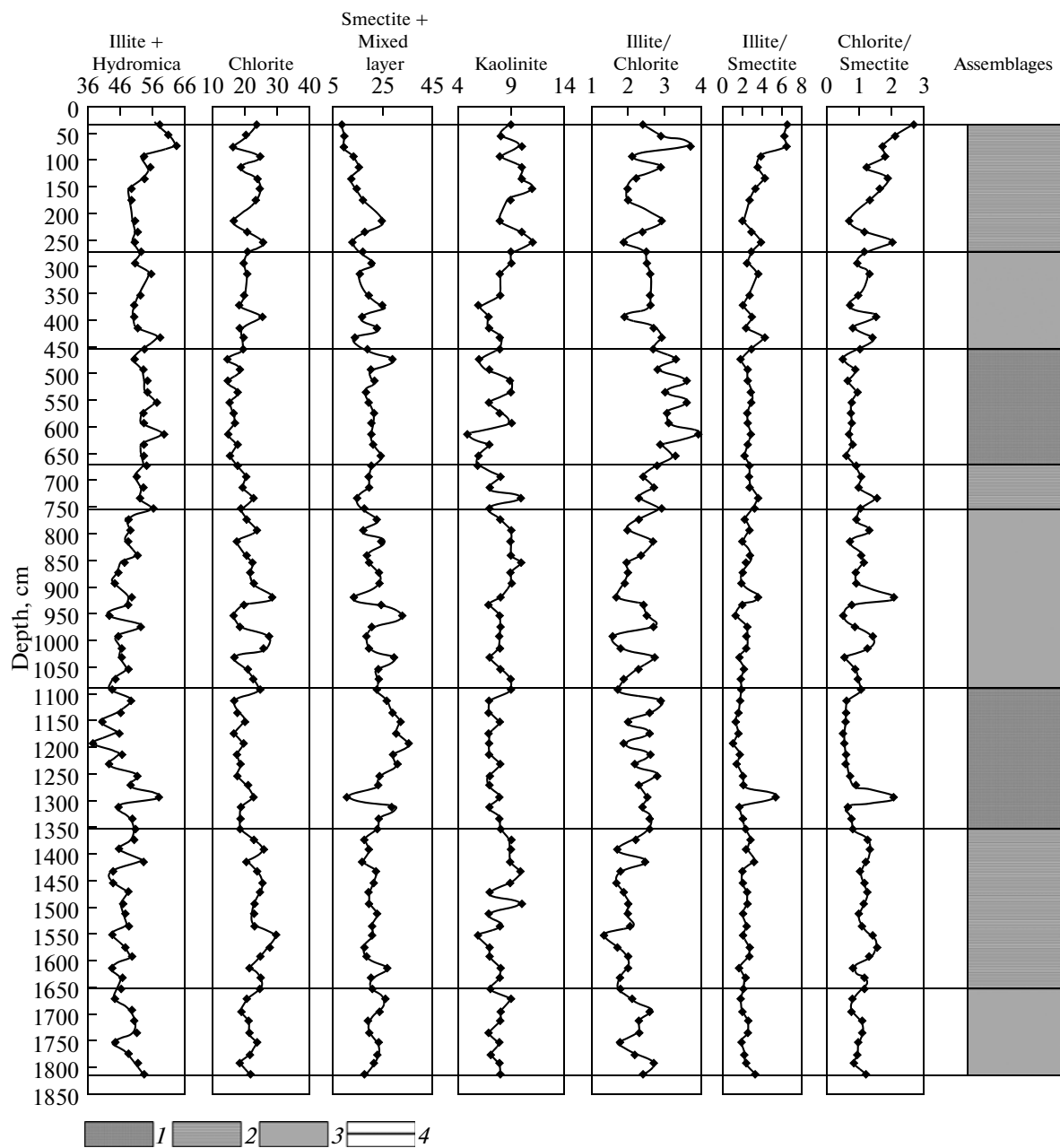
**Table 3.** The content of clay minerals in the <2- $\mu\text{m}$  size fraction (% rel.), mineral ratios, and mineral assemblages

Sample, cm	Illite	Chlorite	Smectite	Kaolinite	Illite/Chlorite	Illite/Smectite	Chlorite/Smectite	Assemblage
34–35	58	24	9	9	2.4	6.4	2.7	2
54–55	61	21	10	8	2.9	6.1	2.1	2
74–76	63	17	10	10	3.7	6.3	1.7	2
94–95	53	25	14	8	2.1	3.8	1.8	2
114–115	55	19	16	10	2.9	3.4	1.2	2
134–135	53	24	13	10	2.2	4.1	1.8	2
154–155	49	25	15	11	2	3.3	1.7	2
174–175	49	24	18	9	2	2.7	1.3	2
214–215	50	17	25	8	2.9	2	0.7	1
234–235	51	21	18	10	2.4	2.8	1.2	2
254–255	50	26	13	11	1.9	3.8	2	2
274–275	52	21	18	9	2.5	2.9	1.2	2
294–295	50	20	21	9	2.5	2.4	1	1
314–315	55	21	16	8	2.6	3.4	1.3	2
354–355	52	20	20	8	2.6	2.6	1	1
374–375	50	19	25	6	2.6	2	0.8	1
394–395	50	26	17	7	1.9	2.9	1.5	2
414–415	51	19	23	7	2.7	2.2	0.8	1
434–435	58	20	14	8	2.9	4.1	1.4	2
454–455	53	20	19	8	2.7	2.8	1.1	2
474–475	50	15	29	6	3.3	1.7	0.5	1
494–495	53	19	21	7	2.8	2.5	0.9	1
514–515	54	15	22	9	3.6	2.5	0.7	1
534–535	54	18	19	9	3	2.8	0.9	1
554–555	57	16	20	7	3.6	2.9	0.8	1
574–575	53	17	22	8	3.1	2.4	0.8	1
594–595	53	17	21	9	3.1	2.5	0.8	1
614–615	59	15	21	5	3.9	2.8	0.7	1
634–635	53	18	22	7	2.9	2.4	0.8	1
654–655	53	16	25	6	3.3	2.1	0.6	1
674–675	54	19	21	6	2.8	2.6	0.9	1
694–695	51	21	20	8	2.4	2.6	1.1	2
714–715	53	20	20	7	2.7	2.7	1	2
734–735	52	23	15	10	2.3	3.5	1.5	2
754–755	56	19	18	7	2.9	3.1	1.1	2
774–775	48	21	23	8	2.3	2.1	0.9	1
794–795	49	24	18	9	2	2.7	1.3	2
814–815	48	18	25	9	2.7	1.9	0.7	1
841–842	51	21	19	9	2.4	2.7	1.1	2
854–855	47	23	20	10	2	2.4	1.2	2
874–875	45	22	24	9	2	1.9	0.9	1
894–895	44	23	24	9	1.9	1.8	1	1
921–922	49	29	14	8	1.7	3.5	2.1	2
934–935	48	20	25	7	2.4	1.9	0.8	1
954–955	42	17	33	8	2.5	1.3	0.5	1
974–975	52	19	21	8	2.7	2.5	0.9	1
994–995	45	28	19	8	1.6	2.4	1.5	2

Table 3. (Contd.)

Sample, cm	Illite	Chlorite	Smectite	Kaolinite	Illite/Chlorite	Illite/Smectite	Chlorite/Smectite	Assemblage
974–975	52	19	21	8	2.7	2.5	0.9	1
994–995	45	28	19	8	1.6	2.4	1.5	2
1016–1017	46	26	20	8	1.8	2.3	1.3	2
1034–1035	46	17	30	7	2.7	1.5	0.6	1
1054–1055	48	21	23	8	2.3	2.1	0.9	1
1074–1075	44	23	24	9	1.9	1.8	1	1
1094–1095	43	25	23	9	1.7	1.9	1.1	2
1114–1115	49	17	27	7	2.9	1.8	0.6	1
1134–1135	46	18	29	7	2.6	1.6	0.6	1
1154–1155	40	20	32	8	2	1.3	0.6	1
1174–1175	45	17	31	7	2.6	1.5	0.5	1
1194–1195	37	20	36	7	1.9	1	0.6	1
1214–1215	46	18	29	7	2.6	1.6	0.6	1
1234–1235	42	19	31	8	2.2	1.4	0.6	1
1254–1255	51	18	24	7	2.8	2.1	0.8	1
1274–1275	49	21	23	7	2.3	2.1	0.9	1
1294–1295	58	23	11	8	2.5	5.3	2.1	2
1314–1315	45	19	29	7	2.4	1.6	0.7	1
1334–1335	49	19	24	8	2.6	2	0.8	1
1354–1355	50	19	23	8	2.6	2.2	0.8	1
1374–1375	50	23	18	9	2.2	2.8	1.3	2
1394–1395	45	26	20	9	1.7	2.3	1.3	2
1414–1415	53	21	17	9	2.5	3.1	1.2	2
1434–1435	43	24	23	10	1.8	1.9	1	2
1454–1455	43	26	22	9	1.7	2	1.2	2
1474–1475	48	25	20	7	1.9	2.4	1.3	2
1494–1495	47	23	20	10	2	2.4	1.2	2
1514–1515	47	23	23	7	2	2	1	2
1534–1535	48	23	21	8	2.1	2.3	1.1	2
1554–1555	43	30	21	6	1.4	2	1.4	2
1574–1575	47	28	18	7	1.7	2.6	1.6	2
1594–1595	49	25	19	7	2	2.6	1.3	2
1614–1615	43	22	27	8	2	1.6	0.8	1
1634–1635	46	25	21	8	1.8	2.2	1.2	2
1654–1655	46	25	22	7	1.8	2.1	1.1	2
1674–1675	44	21	26	9	2.1	1.7	0.8	1
1694–1695	49	19	24	8	2.6	2	0.8	1
1714–1715	50	22	20	8	2.3	2.5	1.1	2
1734–1735	51	22	20	7	2.3	2.5	1.1	2
1754–1755	44	24	24	8	1.8	1.8	1	1
1774–1775	48	22	23	7	2.2	2.1	1	1
1794–1795	51	19	22	8	2.7	2.3	0.9	1
1814–1815	53	22	17	8	2.4	3.1	1.3	2

Note: Assemblage 1 is kaolinite–chlorite–smectite–illite, and assemblage 2 is kaolinite–smectite–chlorite–illite.



**Fig. 4.** Content of clay minerals (% rel.) and the distribution of clay minerals ratios and assemblages in column SO201-2-101-KL. Symbols: (1) kaolinite–smectite–chlorite–illite assemblage, (2) kaolinite–chlorite–smectite–illite assemblage, (3) fine inter-layering of two assemblages, and (4) boundaries between clay mineral assemblages.

the climatic conditions of cold periods, whereas the mixed-layer and young smectites formed in soils in continental setting through the alteration of volcanic material.

We applied R-mode factor analysis with varimax rotation to the standardized data to explore the distribution of clay minerals in the cored section. There are three factors fully accounting for the variance of four variables (groups of clay minerals). Factor 1 accounts

for 46.924, factor 2 accounts for 35.346, and factor 3 explains 17.730 of the variance. Factor loadings are shown in Table 6. As can be seen, factor 1 reflects the mutual dilution of illite and smectite, i.e., the illite/smectite ratio. Factor 2 represents the mutual dilution of chlorite and smectite, with the dominance of chlorite, and factor 3 reflects the distribution of kaolinite. Based on factor 1 (Fig. 5), the cored section was divided into two nearly equal parts, the upper part (0–

**Table 4.** Composition of clay mineral assemblages (% rel.) in the <2- $\mu\text{m}$  size fraction from the Pleistocene sediments of the Shirshov Ridge (our data), Yukon Delta sediments [17], and surficial sediments from the northern continental margin of Alaska [17]

Location	Assemblage, number of samples	Illite (Ill)	Chlorite (Chl)	Smectite (Sm)	Kaolinite (K)
Shirshov Ridge	K-Chl-Sm-Ill ( $n = 44$ )	49	19	25	7
Shirshov Ridge	K-Sm-Chl-Ill ( $n = 43$ )	49	24	18	9
Shirshov Ridge	$n = 87$	49	21	21	8
Yukon Delta	$n = 6$	41	26	21	12
Northern Alaska continental margin	$n = 10$	55	22	14	10

**Table 5.** Correlation matrix for the clay mineral groups identified in the Shirshov Ridge sediments

Component	Illite + Hydromica	Chlorite	Smectite + Mixed-layer	Kaolinite	Illite/Chlorite	Illite/Smectite	(Illite + Chlorite)/Smectite	<0.002 mm
Illite + Hydromica	1.000	-0.324	-0.702	0.004	0.696	0.780	0.683	0.096
Chlorite		1.000	-0.414	0.271	-0.880	0.161	0.328	-0.318
Smectite + Mixed-layer			1.000	-0.408	-0.010	-0.888	-0.911	0.130
Kaolinite				1.000	-0.239	0.320	0.334	-0.052
Illite/Chlorite					1.000	0.226	0.058	0.263
Illite/Smectite						1.000	0.976	-0.073
(Illite + Chlorite)/Smectite							1.000	-0.125
<0.002 mm								1.000

850 cm) dominated by positive loadings and the lower part dominated by negative loadings. These results indicate the importance of illite for the upper part of the section and smectite for the lower part. The distribution of factor 2 revealed that, despite the overall dominance of chlorite (positive factor loadings), negative loadings clearly dominating within two intervals coincide with the occurrence of the kaolinite–chlorite–smectite–illite assemblage identified in this study. The distribution of factor 3 indicates an increasing supply of kaolinite at 0–285, 750–910, and 1390–1450 cm. The sum of factors 1 and 2 clearly indicates the dominance of the sum of illite and chlorite in smectite dilution in the upper part of the section (0–195 cm). Below this depth (at 195–450, 700–1050, 1290–1815 cm), the section shows alternation of positive and negative loadings, and the importance of smectite is recognized at 450–700 and 1050–1290 cm, especially in the lowermost interval.

An attempt was made to find out whether the distinction between these two mineral assemblages influences the bulk composition of samples. The semiquantitative approach of Biscaye applied in this study assumes that the sum of clay minerals in the quantified fraction is 100%. However, the fraction is known to contain X-ray amorphous and non-clay crystalline materials (quartz, feldspars, etc.). The difference in the contents of chlorite and smectite in the samples is about 6% rel. Since the <2- $\mu\text{m}$  size fraction accounts for 25–30 wt % of the samples (Table 2), it can be assumed that this difference in the bulk sample is about 1–2%, or even two times lower if the real fraction composition is taken into account. It should be noted that the two clay mineral assemblages are not very different in composition, especially compared with the differences between quartz and feldspars. This suggests that the above differences between the assemblages have no significant influence on the bulk sample composition.

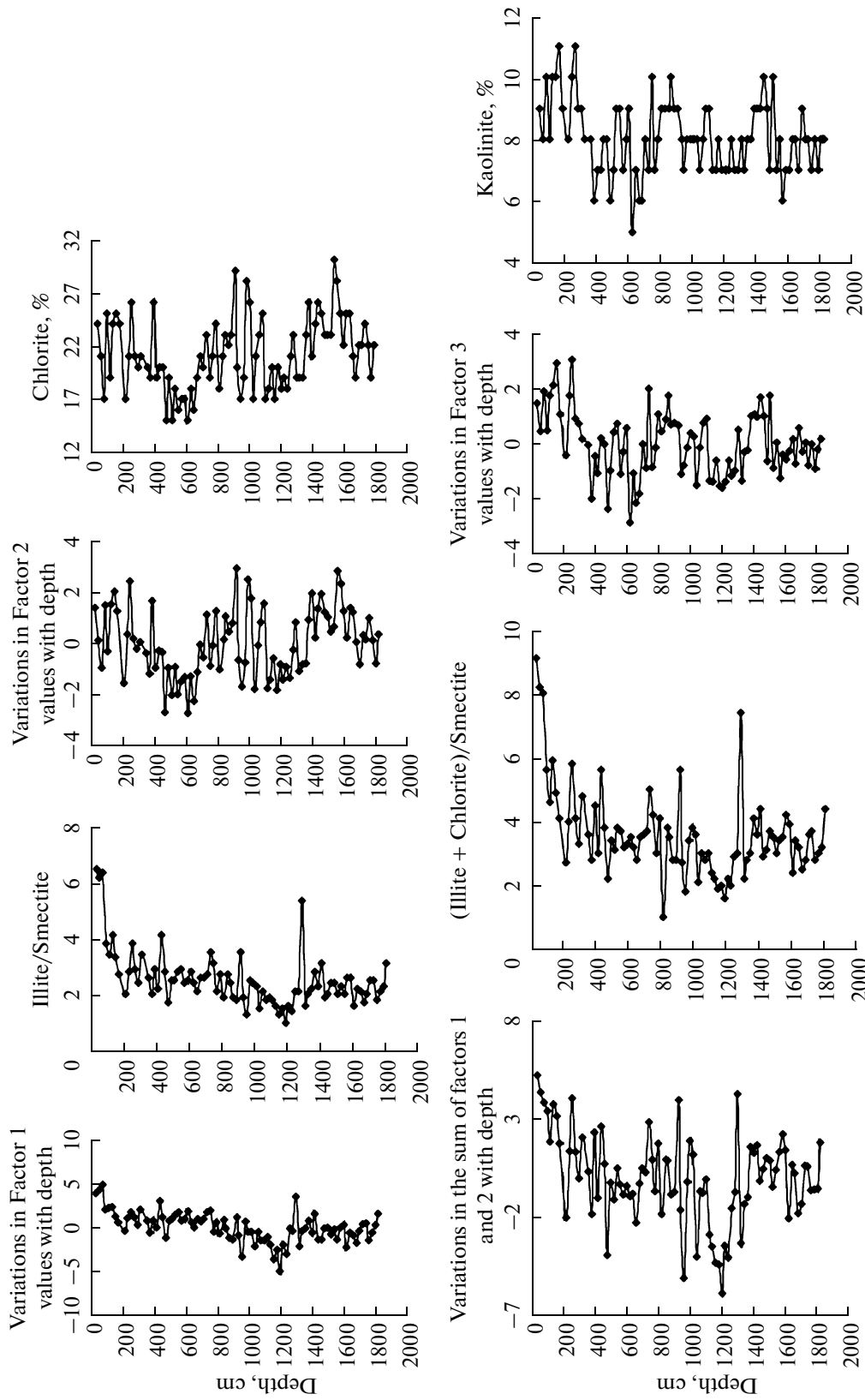


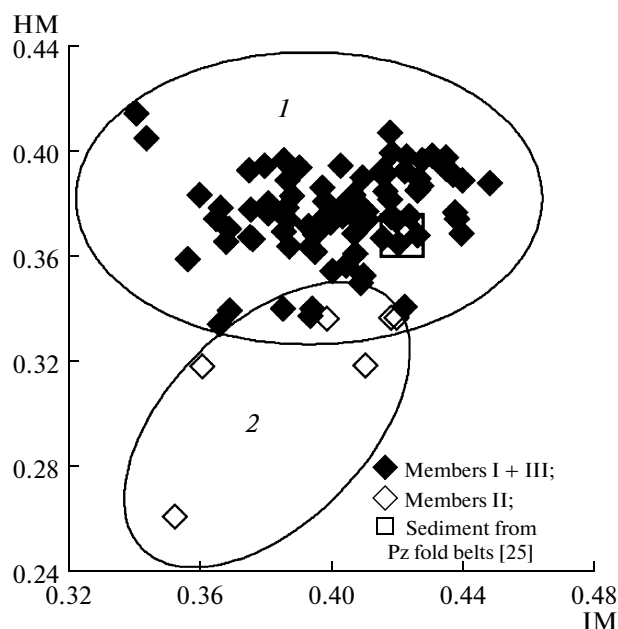
Fig. 5. Relationships between variations in the main factors controlling the distribution of clay minerals along the cored section and clay mineral contents and ratios.

**Sediment geochemistry.** Table 7 shows XRF data for the bottom sediments, which are consistent with the observed monotonous lithology of the sedimentary section (except for member 2). Minor variations in the content of the >0.063-mm fraction (up to 16%) do not influence the chemical composition of bulk sediments (Table 8), which is also supported by the average concentrations of the elements in sample groups with different sand content when calculated to a silica-free basis (Table 9).

The analytical results and data in Fig. 6 allow us to conclude that the sedimentary material in the samples (except for samples from the sandy interval) was most likely derived from the shales of Paleozoic fold belts [25]. The only appropriate source rocks in the islands and continental blocks around the Bering Sea are those of Alaska [26]. It is likely that the eroded material from Alaska was transported by the Yukon River draining the Brooks and Alaska ranges, as well as the Yukon–Tanana Upland. The latter seems to be the most likely provenance; its Paleozoic rocks are mostly carbonate-free phyllites, schists, and gneisses intruded by granites [24]. The upper reaches of several right tributaries of the Yukon River drain the Brooks Range, whose Paleozoic section (and that of the Alaska Range) is dominated by dolomites [27], which are found to be absent in our samples.

The slightly lower contents of  $\text{Al}_2\text{O}_3$ ,  $\text{TiO}_2$ ,  $\text{Fe}_2\text{O}_3^*$ ,  $\text{K}_2\text{O}$ , and  $\text{P}_2\text{O}_5$  in our samples relative to the Paleozoic shales (Table 8) can be explained by the aforementioned wash-out of the finest clay fraction from Shirshov Ridge sediments, and the lower MnO reflects anaerobic diagenetic conditions and periodic deposition in the oxygen minimum zone, which will be discussed below. The younger fine-grained marine sediments have higher loss on ignition values than the Paleozoic shales, which underwent diagenetic, catagenetic, and early metamorphic alteration and partial loss of volatile components. As was mentioned above, the high  $\text{Na}_2\text{O}$  values are due to sea salt contamination of the samples. It is also obvious that the composition of modern sediments of the Bering Sea shelf is distinctly different from that of our samples (Table 8), because the shelf area is dominated by marine blanket sands [1].

Figure 6 shows that the average composition of members 1 and 3 (cluster 1) is characterized by a higher value of hydrolysate module and a lower value of iron module relative to the average shale composition of Paleozoic fold belts (source rocks). Note that the hydrolysate module (HM) is  $(\text{Al}_2\text{O}_3 + \text{TiO}_2 + \text{Fe}_2\text{O}_3 + \text{FeO} + \text{MnO})/\text{SiO}_2$ , and the iron module (IM) is  $(\text{Fe}_2\text{O}_3 + \text{FeO} + \text{MnO})/(\text{TiO}_2 + \text{Al}_2\text{O}_3)$ . The first value reflects lower Al and K contents in the sediments, and the second value indicates the deposition of dissolved



**Fig. 6.** Relationship between the values of hydrolysate and iron modules [47] in the sediments. 1 and 2 are clusters. See text for explanation.

Fe in the sediments in the river water–seawater mixing zone. This is a well-known phenomenon, which was described, for instance, in our previous study in the Yenisei Gulf of the Kara Sea [28].

The chemical composition of member 2 sediments differs markedly from that of the enclosing rocks, suggesting a proximal source area within the Shirshov Ridge on the basis on sedimentological considerations. The geology of the ridge is poorly understood, but available data [2] indicate that it consists of Triassic, Cretaceous, and Paleocene radiolarites, Lower–Upper Cretaceous amphibolitic gabbros, and Lower Oligocene–Pleistocene shallow-water terrigenous sediments.

For the better understanding of the geochemical features of the sediments, Table 10 presents a Pearson correlation matrix of geochemical and grain-size components. The analysis of these data revealed several major groups of element associations. The first association comprises  $\text{SiO}_2$ , Sr, Zr, Ni, and sand; i.e., it is controlled by the mineral components of the sand fraction. The second association is  $\text{Al}_2\text{O}_3$ , Ti, Fe, Mn, K, Mg, P, V, Co, Cu, Zn, Rb, Y, Nb, Ba, Pb, LOI, MnO, and silt. This is indicative of a mixture of the fine-grained clastic, silt and, in part, clay material with organic matter, and Fe and Mn oxyhydroxides (with the trace elements, both adsorbed onto the surface and incorporated into their mineral lattices). The third association comprises CaO, Sr, and sand and corresponds to the components of the sand fraction such as foraminiferal and mollusk shells, with isomorphic substitution of Sr for Ca. The

**Table 6.** The distribution of factor loadings for clay minerals

Sample no., cm	Factor 1	Factor 2	Factor 3	Factor 1 + Factor 2
34	3.83713	1.38016	1.50113	5.21729
54	4.22948	0.113844	0.461861	4.343324
74	4.80074	-0.95765	1.936	3.843086
94	1.9131	1.47626	0.486766	3.38936
114	2.16489	-0.34268	1.76771	1.822209
134	2.23685	1.4891	2.15979	3.72595
154	1.14046	1.99605	2.96135	3.13651
174	0.521494	1.22835	1.09406	1.749844
214	-0.48209	-1.5604	-0.42116	-2.04249
234	0.982453	0.34595	1.78124	1.328403
254	1.67062	2.38602	3.10309	4.05664
274	1.12326	0.165209	0.941035	1.288469
294	0.202471	-0.23483	0.755057	-0.03236
314	2.00492	0.016966	0.191108	2.021886
354	0.703599	-0.39623	-0.03956	0.307374
374	-0.65824	-1.19581	-1.99704	-1.85405
394	0.67536	1.62156	-0.43489	2.29692
414	-0.0784	-0.97638	-1.06566	-1.05479
434	2.89988	-0.29069	0.232451	2.609193
454	1.07402	-0.37998	0.005579	0.694044
474	-1.26079	-2.70183	-2.38505	-3.96262
494	0.696797	-0.9544	-0.97691	-0.2576
514	0.887528	-2.03905	0.452544	-1.15152
534	1.39704	-0.92717	0.740986	0.469871
554	1.67101	-2.00191	-1.08488	-0.3309
574	0.645345	-1.51661	-0.28646	-0.87127
594	0.897191	-1.32642	0.597899	-0.42923
614	1.78076	-2.7334	-2.86135	-0.95264
634	0.488568	-1.31328	-1.07134	-0.82471
654	-0.05243	-2.26018	-2.15025	-2.31261
674	0.845687	-1.13762	-1.81747	-0.29193
694	0.547466	-0.05546	0.009467	0.492011
714	0.85855	-0.5813	-0.8804	0.277253
734	1.70669	1.09913	2.01805	2.80582
754	1.81413	-0.90751	-0.84175	0.906622
774	-0.59816	-0.09369	-0.12443	-0.69185
794	0.497246	1.23577	1.09513	1.733016
814	-0.82955	-1.02343	0.470323	-1.85298
841	0.762939	0.145866	0.895445	0.908805
854	-0.15351	1.02035	1.79271	0.86684
874	-1.29583	0.420854	0.72102	-0.87498
894	-1.5045	0.777707	0.772386	-0.72679
921	1.06427	2.908	0.692861	3.97227
934	-0.98549	-0.66502	-1.10646	-1.65051
954	-3.44751	-1.68978	-0.7822	-5.13729



Table 6. (Contd.)

Sample no., cm	Factor 1	Factor 2	Factor 3	Factor 1 + Factor 2
974	0.58226	-0.7817	-0.13787	-0.19944
994	-0.58322	2.47114	0.415972	1.887916
1016	-0.55853	1.74799	0.269088	1.189457
1034	-2.23787	-1.80878	-1.48509	-4.04665
1054	-0.61028	-0.08998	-0.12389	-0.70026
1074	-1.55704	0.79379	0.774724	-0.76325
1094	-1.60598	1.52437	0.922688	-0.08161
1114	-1.15892	-1.75014	-1.34823	-2.90906
1134	-2.0842	-1.43321	-1.38823	-3.51741
1154	-3.73947	-0.59184	-0.58134	-4.33131
1174	-2.60425	-1.82627	-1.53041	-4.43052
1194	-5.07228	-0.84005	-1.60246	-5.91233
1214	-2.03772	-1.44744	-1.39029	-3.48516
1234	-3.15634	-0.93368	-0.58774	-4.09002
1254	-0.21187	-1.35815	-1.16342	-1.57002
1274	-0.46948	-0.27072	-0.96409	-0.7402
1294	3.44778	0.809446	0.519184	4.257226
1314	-2.25245	-1.08873	-1.33866	-3.34118
1334	-0.53305	-0.82921	-0.27311	-1.36227
1354	-0.20102	-0.80121	-0.22626	-1.00223
1374	0.68571	0.885101	1.04467	1.570811
1394	-0.67105	1.92007	1.10804	1.24902
1414	1.46944	0.188881	0.987255	1.658321
1434	-1.50781	1.33897	1.71018	-0.16884
1454	-1.44827	1.89871	1.01937	0.45044
1474	-0.19895	1.2073	-0.62294	1.008348
1494	-0.13734	1.01541	1.79199	0.878066
1514	-0.90905	0.449794	-0.86037	-0.45925
1534	-0.25849	0.64757	0.06787	0.389083
1554	-1.48607	2.79891	-1.25082	1.31284
1574	-0.09826	2.31469	-0.37793	2.216428
1594	0.157328	1.22788	-0.57717	1.385208
1614	-2.30469	0.203117	-0.2529	-2.10157
1634	-0.72433	1.37612	0.172761	0.651792
1654	-0.97617	1.18594	-0.7116	0.209766
1674	-1.88257	0.048203	0.5818	-1.83437
1694	-0.53912	-0.82736	-0.27284	-1.36647
1714	0.312525	0.30944	0.062003	0.621965
1734	0.421	0.138596	-0.77677	0.559596
1754	-1.60066	0.962479	-0.0152	-0.63818
1774	-0.72058	0.099126	-0.91084	-0.62146
1794	0.213857	-0.79857	-0.1831	-0.58471
1814	1.45209	0.349528	0.196165	1.801618

**Table 7.** Chemical composition of bottom sediments from the Shirshov Ridge, wt %

No., cm	LOI	SiO <sub>2</sub>	Al <sub>2</sub> O <sub>3</sub>	TiO <sub>2</sub>	Fe <sub>2</sub> O <sub>3</sub>	MnO	K <sub>2</sub> O	CaO	MgO
0	9.74	58.15	14.82	0.63	5.74	0.05	2.29	1.79	2.8
16	9.92	55.72	14.93	0.65	5.95	0.053	2.41	3.56	2.91
34	10.74	56.12	14.86	0.61	6.07	0.051	2.21	1.88	2.72
54	9.87	56.88	16.39	0.7	5.81	0.042	2.73	1.71	2.64
74	9.59	56.47	16.69	0.71	5.86	0.045	2.74	1.79	2.62
94	7.53	58.66	15.79	0.7	5.86	0.053	2.47	2.5	2.71
114	7.1	58.66	15.41	0.69	6.05	0.06	2.31	2.94	2.72
134	7.5	58.13	15.59	0.71	6.48	0.061	2.41	2.32	2.91
154	7.65	57.98	15.67	0.71	6.31	0.059	2.38	2.38	2.91
174	7.52	58.26	15.89	0.71	6.14	0.06	2.39	2.19	2.87
214	7.48	58.88	15.51	0.7	6.21	0.055	2.42	1.91	2.76
234	6.91	59.77	15.43	0.7	5.87	0.054	2.34	2.09	2.66
254	7.22	59.28	15.52	0.72	6.02	0.054	2.34	2.02	2.62
274	7.74	58.71	15.25	0.74	6.1	0.054	2.41	2.06	2.84
294	7.48	59.39	15.26	0.71	6.1	0.052	2.37	1.92	2.56
314	7.38	59.45	14.76	0.7	6.01	0.053	2.40	2.04	2.65
354	8.08	58.33	15.08	0.72	6.5	0.055	2.48	1.85	2.78
374	8.41	58.46	14.83	0.7	6.13	0.056	2.57	1.85	2.73
394	8.11	59.36	14.89	0.74	6.1	0.051	2.51	1.62	2.7
414	8.11	58.85	15.19	0.73	6.38	0.052	2.47	1.62	2.78
434	7.97	59.49	14.84	0.72	5.95	0.052	2.39	2.09	2.63
454	6.48	61.39	14.47	0.71	5.53	0.05	2.33	2.46	2.52
474	8.22	59.01	15.07	0.72	6.22	0.049	2.50	1.73	2.62
494	8.03	59.71	15.18	0.73	5.78	0.05	2.57	1.74	2.59
514	8.15	59.64	14.73	0.69	6.02	0.048	2.45	1.92	2.55
534	8.69	59.63	14.07	0.69	5.97	0.046	2.38	2.26	2.41
554	8.39	60.21	13.86	0.67	5.65	0.048	2.29	2.67	2.39
574	8.6	58.94	14.39	0.69	6.07	0.05	2.39	2.18	2.6
594	7.81	59.62	15.14	0.72	6.04	0.05	2.51	1.6	2.6
614	8.38	59.06	14.99	0.72	6.1	0.051	2.49	1.77	2.65
634	7.67	59.82	15.09	0.73	6.18	0.052	2.54	1.62	2.66
654	8.71	58.73	14.77	0.72	6.4	0.055	2.46	1.85	2.71
674	8.99	58.39	14.62	0.7	6.35	0.058	2.39	2.03	2.82
694	8.15	58.2	15.21	0.71	6.84	0.059	2.45	1.83	3.02
714	7.48	59.02	15.22	0.69	6.34	0.06	2.49	2.22	2.98
734	8.61	59.05	14.96	0.63	5.48	0.052	2.48	1.76	2.73
754	7.62	59.41	15.55	0.69	5.85	0.054	2.56	1.72	2.76
774	8.2	58.65	15.48	0.7	6.55	0.052	2.55	1.63	2.71
794	9.55	56.84	15.53	0.69	6.17	0.059	2.50	2.03	2.72
814	7.16	60.27	14.86	0.72	6.4	0.053	2.47	1.72	2.69
841	7.51	59.13	15.15	0.73	6.41	0.057	2.37	1.91	2.84
854	7.35	59.12	15.14	0.73	6.55	0.06	2.28	1.93	2.81
874	7.5	58.58	15.38	0.74	6.72	0.066	2.36	2.05	2.98
894	7.85	57.86	15.51	0.74	6.71	0.066	2.33	2.02	2.98
921	8.95	57.19	15.09	0.72	6.79	0.063	2.32	1.86	2.97
934	9.11	58.87	14.97	0.71	5.81	0.052	2.37	1.71	2.56
954	8.5	58.9	15.29	0.73	6.02	0.055	2.39	1.67	2.59
974	9.74	57.13	15.11	0.7	6.49	0.057	2.42	1.69	2.77

Table 7. (Contd.)

No., cm	Na <sub>2</sub> O	P <sub>2</sub> O <sub>5</sub>	Cr	S	V	Co	Ni	Cu
0	3.27	0.156	0.012	0.74	0.0195	0.0019	0.005	0.0031
16	2.87	0.161	0.009	0.41	0.0164	0.0022	0.0045	0.0029
34	3.32	0.145	0.010	0.86	0.0189	0.0013	0.0053	0.003
54	2.36	0.177	0.011	0.32	0.0214	0.0013	0.0046	0.0023
74	2.83	0.174	0.011	0.17	0.0266	0.0014	0.0043	0.0026
94	3.07	0.156	0.009	0.16	0.0172	0.0018	0.0038	0.0022
114	3.33	0.151	0.010	0.20	0.0172	0.0018	0.0041	0.0029
134	3.25	0.148	0.009	0.23	0.0178	0.0017	0.0046	0.0031
154	3.40	0.149	0.009	0.23	0.0172	0.0018	0.0042	0.0026
174	3.31	0.15	0.010	0.19	0.0172	0.0014	0.0044	0.0027
214	3.37	0.146	0.009	0.31	0.0186	0.0012	0.0041	0.0024
234	3.55	0.143	0.008	0.20	0.0186	0.0019	0.0035	0.0021
254	3.38	0.143	0.009	0.34	0.0172	0.0014	0.0038	0.0024
274	3.27	0.146	0.008	0.27	0.0171	0.0014	0.0035	0.0022
294	3.37	0.142	0.008	0.34	0.0176	0.0012	0.0037	0.0022
314	3.69	0.145	0.008	0.41	0.0155	0.001	0.0039	0.0023
354	3.47	0.143	0.010	0.28	0.018	0.001	0.0039	0.0029
374	3.41	0.141	0.008	0.34	0.0167	0.0017	0.0037	0.0027
394	3.05	0.151	0.009	0.34	0.0177	0.001	0.0038	0.0023
414	3.04	0.146	0.009	0.34	0.0165	0.0011	0.0039	0.0024
434	3.09	0.147	0.009	0.36	0.0178	0.0013	0.0038	0.0022
454	3.27	0.146	0.009	0.29	0.0151	0.0013	0.0032	0.0018
474	2.98	0.148	0.010	0.38	0.0184	0.0016	0.0045	0.0021
494	3.03	0.146	0.009	0.14	0.0168	0.0016	0.0034	0.0019
514	2.92	0.145	0.009	0.46	0.0137	0.0012	0.0039	0.002
534	2.93	0.137	0.009	0.49	0.0156	0.0015	0.0037	0.0022
554	2.95	0.141	0.009	0.41	0.0149	0.0009	0.0036	0.0018
574	3.06	0.146	0.009	0.49	0.0156	0.0013	0.0039	0.0019
594	3.10	0.145	0.010	0.27	0.0175	0.0014	0.0043	0.0024
614	2.94	0.144	0.009	0.28	0.0168	0.0011	0.0042	0.0028
634	2.89	0.144	0.010	0.21	0.0187	0.0016	0.0042	0.0022
654	2.80	0.144	0.009	0.28	0.0191	0.0017	0.0047	0.0025
674	2.92	0.147	0.009	0.24	0.0172	0.0013	0.0044	0.0028
694	2.84	0.151	0.010	0.26	0.0179	0.0018	0.0053	0.003
714	2.83	0.159	0.012	0.20	0.0186	0.0019	0.0052	0.003
734	3.49	0.152	0.009	0.25	0.0187	0.001	0.0037	0.0022
754	3.15	0.156	0.015	0.13	0.0189	0.0014	0.0048	0.0025
774	2.93	0.15	0.012	0.23	0.0197	0.001	0.0046	0.0033
794	2.92	0.153	0.010	0.30	0.0228	0.0014	0.0043	0.0034
814	2.91	0.155	0.009	0.23	0.0176	0.0016	0.004	0.0026
841	3.01	0.153	0.010	0.25	0.0216	0.0016	0.004	0.0029
854	3.10	0.153	0.011	0.35	0.0185	0.0017	0.0044	0.0028
874	2.97	0.158	0.011	0.15	0.0209	0.0012	0.0048	0.0035
894	3.13	0.155	0.009	0.21	0.0205	0.0028	0.0044	0.0034
921	3.02	0.152	0.010	0.51	0.019	0.0015	0.0042	0.0036
934	3.10	0.138	0.009	0.31	0.0171	0.0012	0.0037	0.0032
954	3.18	0.141	0.010	0.23	0.0181	0.0014	0.0037	0.0031
974	3.13	0.145	0.009	0.38	0.0194	0.001	0.0042	0.004

Table 7. (Contd.)

No., cm	Zn	Rb	Sr	Y	Zr	Nb	Ba	As	Pb
0	0.0105	0.0086	0.0218	0.0024	0.0124	0.001	0.088	0.0022	0.0012
16	0.0105	0.0084	0.0277	0.0026	0.0125	0.0009	0.097	0.0015	0.0016
34	0.0106	0.0086	0.0196	0.0024	0.0121	0.0011	0.088	0.0023	0.0011
54	0.0106	0.0112	0.0197	0.0027	0.0144	0.0016	0.089	0.0011	0.0022
74	0.0108	0.011	0.0197	0.0025	0.0144	0.0018	0.093	0.0024	0.002
94	0.0106	0.0088	0.0229	0.0028	0.015	0.0008	0.079	0.0009	0.0013
114	0.0099	0.0075	0.0255	0.0025	0.0155	0.001	0.074	0.0016	0.0016
134	0.0102	0.0088	0.0232	0.0027	0.0149	0.0012	0.078	0.0025	0.0016
154	0.0105	0.0079	0.0229	0.0026	0.0142	0.0009	0.080	0.0022	0.0014
174	0.0108	0.0082	0.0223	0.0027	0.0153	0.0012	0.075	0.0018	0.0012
214	0.0102	0.0086	0.0228	0.0027	0.0152	0.0008	0.077	0.0018	0.0011
234	0.0095	0.0087	0.0241	0.0028	0.0161	0.0008	0.071	0.0013	0.0018
254	0.0099	0.0085	0.0232	0.0025	0.0153	0.001	0.073	0.0027	0.0018
274	0.012	0.0089	0.0228	0.0026	0.0162	0.0011	0.071	0.0029	0.0016
294	0.0098	0.0086	0.0224	0.0025	0.0159	0.0013	0.069	0.003	0.0012
314	0.0094	0.0084	0.0247	0.0026	0.0165	0.0014	0.076	0.0029	0.0021
354	0.0105	0.0092	0.0232	0.0027	0.0154	0.0014	0.070	0.0007	0.0012
374	0.0102	0.0102	0.0215	0.0027	0.0158	0.0011	0.078	0.0015	0.0017
394	0.0105	0.0104	0.021	0.0027	0.0167	0.0011	0.085	0.0024	0.0022
414	0.0108	0.0102	0.0222	0.0027	0.016	0.0015	0.071	0.0017	0.0024
434	0.0103	0.0098	0.0241	0.0027	0.0166	0.001	0.072	0.0027	0.0012
454	0.0095	0.0091	0.0268	0.0025	0.0173	0.001	0.073	0.0015	0.0017
474	0.0109	0.0099	0.0219	0.0028	0.0168	0.001	0.076	0.0026	0.0015
494	0.0102	0.0105	0.0229	0.0028	0.0163	0.0015	0.076	0.0023	0.0018
514	0.0104	0.01	0.022	0.0028	0.0156	0.0012	0.077	0.0023	0.002
534	0.0099	0.0094	0.0214	0.0025	0.0156	0.0014	0.073	0.0018	0.0015
554	0.0098	0.0087	0.0241	0.0027	0.0156	0.0008	0.073	0.0039	0.0016
574	0.0101	0.0094	0.0228	0.0027	0.0155	0.0012	0.076	0.0023	0.0013
594	0.0104	0.0098	0.0212	0.0027	0.0163	0.0015	0.077	0.002	0.0022
614	0.0109	0.0094	0.0223	0.0029	0.0163	0.0012	0.080	0.0028	0.0012
634	0.0103	0.0102	0.0213	0.0028	0.0158	0.0014	0.079	0.0026	0.0013
654	0.011	0.0102	0.0205	0.0028	0.0148	0.0013	0.083	0.0017	0.0016
674	0.0113	0.0091	0.0218	0.003	0.0138	0.0012	0.081	0.0018	0.0016
694	0.0115	0.0095	0.0205	0.0028	0.0135	0.0009	0.085	0.0019	0.0024
714	0.0113	0.0095	0.0242	0.0027	0.014	0.0014	0.095	0.0014	0.0015
734	0.0099	0.0092	0.0223	0.0025	0.015	0.0006	0.090	0.002	0.0015
754	0.0108	0.0096	0.022	0.0028	0.0163	0.0012	0.087	0.0019	0.0017
774	0.0105	0.0092	0.0204	0.0027	0.0149	0.0014	0.070	0.0013	0.0017
794	0.0112	0.0094	0.0209	0.0026	0.0138	0.001	0.074	0.0024	0.0018
814	0.01	0.009	0.0222	0.0028	0.0158	0.0013	0.073	0.0017	0.0012
841	0.0101	0.0087	0.0223	0.0029	0.0169	0.001	0.073	0.0008	0.0019
854	0.0099	0.0077	0.022	0.0027	0.0159	0.0014	0.067	0.0018	0.0015
874	0.0107	0.0088	0.0226	0.0028	0.0151	0.0014	0.072	0.0012	0.0014
894	0.0105	0.0083	0.0226	0.003	0.0151	0.0011	0.072	0.0005	0.0013
921	0.0109	0.0084	0.0205	0.0026	0.0141	0.0008	0.073	0.0025	0.0017
934	0.0107	0.0085	0.0221	0.0027	0.0153	0.0012	0.073	0.0019	0.002
954	0.0107	0.0087	0.0207	0.0028	0.0158	0.0013	0.072	0.0019	0.0018
974	0.0112	0.0088	0.0207	0.0027	0.014	0.0012	0.073	0.0019	0.0013

Table 7. (Contd.)

No., cm	LOI	SiO <sub>2</sub>	Al <sub>2</sub> O <sub>3</sub>	TiO <sub>2</sub>	Fe <sub>2</sub> O <sub>3</sub>	MnO	K <sub>2</sub> O	CaO	MgO
994	8.99	57.36	15.21	0.72	6.78	0.058	2.42	1.7	2.77
1016	9.17	56.79	15.04	0.69	6.64	0.062	2.20	1.82	3.15
1034	9.15	58.93	14.38	0.71	6.27	0.053	2.26	1.71	2.56
1054	9.36	59.81	13.96	0.69	5.57	0.052	2.16	1.9	2.4
1074	8.53	60.43	13.99	0.71	5.72	0.053	2.21	1.75	2.44
1094	8.7	60.38	14.05	0.68	5.32	0.052	2.17	2.08	2.34
1114	7.81	59.36	15.15	0.73	6.32	0.054	2.44	1.74	2.73
1134	8.2	58.97	14.78	0.72	6.49	0.053	2.42	1.76	2.64
1154	6.66	59.74	14.81	0.8	6.74	0.07	2.38	1.91	3.06
1174	6.47	60.15	14.59	0.77	6.66	0.069	2.34	1.96	2.95
1194	6.7	60.62	14.57	0.77	6.12	0.063	2.30	1.94	2.8
1214	6.5	61.39	13.95	0.72	6.11	0.065	2.31	1.99	2.34
1234	6.94	59.74	15.35	0.78	6.5	0.051	2.62	1.55	2.72
1254	7.15	59.18	15.32	0.77	6.47	0.052	2.51	1.67	2.74
1274	9.33	57.78	14.66	0.68	6.01	0.057	2.51	2.56	2.53
1294	8.01	59.16	15.62	0.72	5.91	0.054	2.63	1.47	2.63
1314	6.32	60.67	15.1	0.75	6.37	0.057	2.51	1.89	2.68
1334	6.58	60.25	15.31	0.76	6.4	0.059	2.54	1.77	2.78
1354	6.87	59.44	15.25	0.76	6.96	0.058	2.58	1.61	2.95
1374	6.43	59.82	15.44	0.79	7.01	0.061	2.61	1.7	2.87
1394	6.55	59.86	15.36	0.76	6.79	0.059	2.58	1.69	2.87
1414	8.36	57.56	15.34	0.72	6.71	0.059	2.52	1.66	3.15
1434	6.16	60.06	15.47	0.77	6.85	0.067	2.47	1.95	2.99
1454	9.2	55.72	15.27	0.69	6.58	0.064	2.31	1.86	3.2
1474	7.31	58.97	15.31	0.76	6.76	0.069	2.35	1.83	2.99
1494	6.93	59.26	15.07	0.76	7	0.07	2.33	1.88	2.99
1514	8.55	57.66	15.15	0.72	6.51	0.066	2.24	1.81	3.07
1534	9.46	57.01	14.81	0.68	6.08	0.062	2.17	1.91	3.09
1554	8.75	57.99	14.74	0.7	6.19	0.065	2.12	1.82	3.11
1574	3.92	68.23	12.43	0.56	4.51	0.05	2.02	2.22	1.84
1594	4.92	64.05	14.24	0.69	5.31	0.054	2.27	2.01	2.38
1614	5.27	63.33	14.41	0.77	5.97	0.059	2.36	1.93	2.64
1634	6.64	61.01	13.72	0.73	5.96	0.06	2.38	2.3	2.96
1654	5.31	63.04	13.47	0.72	5.74	0.063	2.31	2.7	2.76
1674	5.76	61.8	13.88	0.73	6.05	0.062	2.29	2.21	2.92
1694	7.53	59.94	14.43	0.7	5.98	0.055	2.56	1.86	2.86
1714	7.61	59.19	14.86	0.78	6.35	0.053	2.64	2.05	2.77
1734	6.78	60.61	14.8	0.79	6.57	0.052	2.68	1.59	2.75
1754	6.9	60.22	14.84	0.79	6.77	0.055	2.62	1.72	2.7
1774	7.6	59.21	15.05	0.74	6.41	0.051	2.65	1.65	2.83
1794	10.74	55.16	15.06	0.67	5.9	0.049	2.56	1.88	2.98
1814	6.94	60.94	14.49	0.71	6.15	0.057	2.41	1.66	2.58

**Table 7.** (Contd.)

No., cm	Na <sub>2</sub> O	P <sub>2</sub> O <sub>5</sub>	Cr	S	V	Co	Ni	Cu
994	3.08	0.145	0.010	0.49	0.0184	0.0018	0.0045	0.0037
1016	3.26	0.148	0.008	0.71	0.0181	0.0012	0.0038	0.0029
1034	3.10	0.135	0.010	0.43	0.0154	0.0011	0.0045	0.0031
1054	3.20	0.132	0.010	0.52	0.0165	0.0013	0.0037	0.0027
1074	3.12	0.144	0.022	0.51	0.0165	0.0011	0.0057	0.0026
1094	3.37	0.137	0.008	0.42	0.0141	0.0014	0.0035	0.0022
1114	2.93	0.152	0.011	0.33	0.0186	0.0016	0.0045	0.0027
1134	3.07	0.156	0.009	0.44	0.0179	0.0012	0.0045	0.0032
1154	3.14	0.156	0.010	0.26	0.0205	0.0018	0.0046	0.0033
1174	3.16	0.146	0.010	0.49	0.0208	0.0013	0.0049	0.0032
1194	3.41	0.149	0.010	0.28	0.0159	0.0011	0.0039	0.003
1214	3.59	0.143	0.006	0.50	0.0143	0.0011	0.0033	0.0033
1234	2.96	0.149	0.009	0.24	0.0207	0.0016	0.0046	0.0032
1254	3.29	0.152	0.010	0.27	0.0188	0.0014	0.0043	0.0026
1274	2.93	0.151	0.009	0.43	0.0211	0.0012	0.0046	0.0031
1294	2.88	0.158	0.009	0.42	0.0233	0.0015	0.0049	0.0032
1314	2.89	0.159	0.009	0.22	0.0212	0.0011	0.0045	0.0036
1334	2.81	0.161	0.009	0.20	0.0221	0.002	0.0042	0.0034
1354	2.79	0.16	0.009	0.21	0.0209	0.0018	0.0051	0.0032
1374	2.64	0.162	0.011	0.13	0.0228	0.0019	0.0049	0.0037
1394	2.83	0.164	0.009	0.17	0.0182	0.0023	0.005	0.0035
1414	3.10	0.164	0.013	0.38	0.0184	0.0019	0.0059	0.0035
1434	2.66	0.168	0.010	0.12	0.0216	0.0022	0.0051	0.0051
1454	4.08	0.144	0.009	0.64	0.0215	0.002	0.0049	0.0041
1474	2.98	0.147	0.009	0.22	0.0217	0.0026	0.0049	0.0056
1494	3.06	0.15	0.012	0.19	0.0202	0.0019	0.0054	0.0036
1514	3.45	0.145	0.009	0.44	0.0169	0.002	0.0042	0.0031
1534	3.75	0.137	0.009	0.57	0.0178	0.0014	0.0044	0.0032
1554	3.66	0.132	0.009	0.40	0.0176	0.002	0.0042	0.0036
1574	3.51	0.123	0.009	0.27	0.0115	0.0007	0.0055	0.0017
1594	3.32	0.141	0.010	0.25	0.015	0.0028	0.0193	0.0029
1614	2.68	0.144	0.010	0.14	0.0174	0.0019	0.0042	0.0021
1634	3.42	0.15	0.029	0.34	0.0168	0.0018	0.0064	0.0024
1654	3.14	0.15	0.010	0.33	0.0158	0.0017	0.0039	0.0023
1674	3.18	0.153	0.009	0.52	0.0164	0.0014	0.004	0.0023
1694	3.22	0.144	0.010	0.39	0.015	0.0018	0.0039	0.003
1714	2.86	0.153	0.010	0.22	0.0185	0.0013	0.0048	0.0032
1734	2.68	0.142	0.012	0.14	0.021	0.0011	0.0047	0.003
1754	2.53	0.146	0.011	0.21	0.0191	0.0012	0.0042	0.0035
1774	2.94	0.146	0.011	0.37	0.0186	0.0016	0.0043	0.0026
1794	3.74	0.138	0.009	0.84	0.0157	0.0017	0.0041	0.0026
1814	3.03	0.149	0.010	0.65	0.0155	0.0013	0.0042	0.0029

Table 7. (Contd.)

No., cm	Zn	Rb	Sr	Y	Zr	Nb	Ba	As	Pb
994	0.011	0.009	0.021	0.0027	0.0144	0.0011	0.079	0.0028	0.002
1016	0.0092	0.0077	0.0224	0.0025	0.0148	0.001	0.061	0.0021	0.0015
1034	0.0104	0.0086	0.0212	0.0025	0.0148	0.0013	0.074	0.0027	0.0016
1054	0.0102	0.0076	0.0235	0.0027	0.0152	0.001	0.069	0.0021	0.001
1074	0.01	0.0081	0.0235	0.0027	0.0167	0.0013	0.064	0.0018	0.0009
1094	0.0096	0.008	0.0252	0.0027	0.0167	0.0007	0.071	0.0009	0.0014
1114	0.0108	0.0092	0.0232	0.0028	0.0165	0.0012	0.077	0.0028	0.0015
1134	0.0114	0.0093	0.0233	0.0029	0.0163	0.0012	0.087	0.0021	0.0016
1154	0.011	0.0088	0.025	0.0026	0.0173	0.0013	0.073	0.0003	0.0011
1174	0.011	0.0081	0.0254	0.0031	0.0178	0.0012	0.076	0.0031	0.0013
1194	0.0105	0.0086	0.027	0.003	0.0178	0.0011	0.076	0.0046	0.0017
1214	0.0094	0.0078	0.0252	0.003	0.0187	0.0009	0.079	0.0029	0.0011
1234	0.0124	0.0104	0.0234	0.0031	0.0183	0.001	0.087	0.0034	0.0029
1254	0.0111	0.0101	0.0231	0.0029	0.018	0.0014	0.087	0.0028	0.0019
1274	0.012	0.0104	0.0218	0.003	0.0155	0.0012	0.095	0.0035	0.0012
1294	0.0121	0.0112	0.0192	0.0028	0.016	0.0013	0.098	0.0029	0.0018
1314	0.0114	0.0108	0.0249	0.003	0.0174	0.0016	0.096	0.0009	0.0018
1334	0.0116	0.0108	0.0231	0.003	0.0173	0.0013	0.097	0.0026	0.0022
1354	0.0118	0.0097	0.0209	0.0031	0.0163	0.001	0.094	0.0009	0.0015
1374	0.0128	0.0109	0.0219	0.0031	0.0166	0.0012	0.096	0.0011	0.0019
1394	0.0116	0.0105	0.0213	0.003	0.0176	0.0015	0.098	0.0027	0.0016
1414	0.0114	0.01	0.0206	0.0027	0.0153	0.0011	0.090	0.0024	0.0016
1434	0.0119	0.0089	0.0246	0.0031	0.0163	0.0013	0.089	0.0023	0.0017
1454	0.0111	0.0092	0.0255	0.0027	0.015	0.0011	0.079	0.0032	0.0016
1474	0.0116	0.0084	0.0213	0.0027	0.016	0.0011	0.081	0.0020	0.0019
1494	0.0111	0.0089	0.0236	0.0028	0.0163	0.0013	0.072	0.0012	0.0018
1514	0.0107	0.0082	0.0221	0.0027	0.0144	0.0011	0.069	0.0030	0.0010
1534	0.0099	0.0083	0.0246	0.0027	0.0145	0.0009	0.073	0.0011	0.0015
1554	0.01	0.0078	0.0224	0.0024	0.0145	0.0006	0.064	0.0030	0.0016
1574	0.0073	0.0074	0.0322	0.0024	0.0195	0.0011	0.081	0.0036	0.0014
1594	0.0087	0.0087	0.0291	0.0025	0.0179	0.0014	0.076	0.0021	0.0012
1614	0.0098	0.009	0.0263	0.0031	0.0185	0.0009	0.068	0.0011	0.0023
1634	0.0095	0.0081	0.0263	0.0028	0.0182	0.0011	0.072	0.0016	0.0022
1654	0.0091	0.0083	0.0289	0.0027	0.0184	0.0012	0.071	0.0025	0.0012
1674	0.0089	0.0084	0.0267	0.0029	0.0188	0.0011	0.067	0.0045	0.0013
1694	0.0102	0.0104	0.0265	0.0028	0.017	0.0012	0.067	0.0021	0.0020
1714	0.0117	0.0112	0.0228	0.003	0.0177	0.001	0.080	0.0023	0.0020
1734	0.0115	0.0108	0.0234	0.0029	0.0182	0.0014	0.074	0.0029	0.0013
1754	0.012	0.0104	0.023	0.003	0.0172	0.0009	0.074	0.0025	0.0018
1774	0.0114	0.011	0.0225	0.0031	0.018	0.0014	0.080	0.0052	0.0016
1794	0.0098	0.0101	0.0227	0.0028	0.0158	0.0013	0.073	0.0023	0.0023
1814	0.0097	0.0096	0.0243	0.0028	0.0169	0.0009	0.082	0.0027	0.0012

**Table 8.** Average chemical composition of bottom sediments, wt %

Sediment	>0.063-mm fraction	SiO <sub>2</sub>	Al <sub>2</sub> O <sub>3</sub>	TiO <sub>2</sub>	Fe <sub>2</sub> O <sub>3</sub>	MnO	K <sub>2</sub> O	CaO	MgO	Na <sub>2</sub> O	P <sub>2</sub> O <sub>5</sub>	LOI
Members 1 + 3	0–5 (n = 45)	58.61	15.13	0.72	6.32	0.06	2.45	1.84	2.76	3.05	0.15	8.27
	5–10 (n = 28)	59.26	14.93	0.72	6.22	0.06	2.41	1.96	2.76	3.15	0.15	7.72
	10–16 (n = 9)	59.55	14.95	0.72	6.18	0.06	2.38	2.01	2.77	3.34	0.15	7.24
Members 1 + 3, mean	(n = 82)	58.94	15.04	0.72	6.27	0.06	2.43	1.90	2.76	3.16	0.15	7.97
Shale from Pz fold belts [25]	—	59.46	17.53	0.74	6.56	0.14	3.04	1.82	2.38	1.52	0.13	6.04
Modern bottom sediments from Bering Sea shelf [44] (n = 123)	—	72.31	11.06	0.63	3.73	0.06	1.57	2.83	1.72	3.07	n.d.	n.d.
Member 2	>16 (n = 6)	63.64	13.82	0.70	5.52	0.06	2.26	2.26	2.51	3.18	0.14	5.28

Note: n.d. — not determined.

**Table 9.** Average concentrations of elements in the Shirshov Ridge sediments with variable sand content (wt %) recalculated to a silica-free basis

Sand content	Al <sub>2</sub> O <sub>3</sub>	TiO <sub>2</sub>	Fe <sub>2</sub> O <sub>3</sub>	MnO	K <sub>2</sub> O	CaO	MgO	Na <sub>2</sub> O	P <sub>2</sub> O <sub>5</sub>	LOI
0–5	36.55	1.74	15.27	0.14	5.92	4.45	6.67	7.37	0.36	19.98
5–10	36.64	1.77	15.27	0.15	5.92	4.81	6.77	7.73	0.368	18.95
10–16	36.96	1.78	15.28	0.15	5.88	4.97	6.85	8.26	0.371	17.90
>16	38.01	1.92	15.18	0.17	6.22	6.22	6.90	8.75	0.385	14.52

fourth association is Cr, Ni, S, As, LOI, and Al with clay and, in part, silt fractions. These occur in organic matter, clays, and pyrite.

R-mode factor analysis with varimax rotation was used for the quantification of the role of different factors in the development of the geochemical associations (Table 11). Factors 1 to 4 together explain more than 75% of the variance. The percentage of variance accounted for by each factor are 35.742% for factor 1 (significant positive loadings for Al, K, P, V, Zn, Rb, and Ba), 17.236% for factor 2 (Ti, Fe, Mn, Mg, and Cu), 14.725% for factor 3 (Si, Sr, and Zr), and 8.025% for factor 4 (Ca and Co). The analysis of variations in the factors with depth showed that factors 1, 2, 3 and 4 are controlled by the abundance of silt, clay, sand, and bio-

genic calcite, respectively (Fig. 7). Therefore, the compositional variations in the cored sediments are a function of grain-size fractionation.

In our opinion, the geochemical data in combination with micropaleontological and other indicators are particularly useful for the identification of sediments accumulated under conditions of high productivity and anoxia within the oxygen minimum zone. The Ba/Al [29] and Ca/Al ratios were used as productivity indicators. It should be noted that each of the methods used has its own shortcomings. For example, part of Ba in sand-rich sediments may be related to detrital feldspars. Sand-size bivalve fragments could be derived by slope processes and lateral transport. Therefore, data on the samples with low sand contents and showing compara-



**Table 10.** Correlation matrix for geochemical and grain-size components from the Shirshov Ridge sediments using Pearson correlation coefficients

Component	SiO <sub>2</sub>	Al <sub>2</sub> O <sub>3</sub>	TiO <sub>2</sub>	Fe <sub>2</sub> O <sub>3</sub>	MnO	K <sub>2</sub> O	CaO	MgO
SiO <sub>2</sub>	1.000	-0.666	0.031	-0.444	-0.037	-0.232	0.183	-0.563
Al <sub>2</sub> O <sub>3</sub>		1.000	0.279	0.462	0.005	0.551	-0.294	0.453
TiO <sub>2</sub>			1.000	0.651	0.356	0.416	-0.302	0.387
Fe <sub>2</sub> O <sub>3</sub>				1.000	0.543	0.295	-0.340	0.723
MnO					1.000	-0.345	0.134	0.633
K <sub>2</sub> O						1.000	-0.384	0.120
CaO							1.000	-0.132
MgO								1.000

**Table 10.** (Contd.)

Component	Na <sub>2</sub> O	P <sub>2</sub> O <sub>5</sub>	Cr	S	V	Co	Ni	Cu	Zn	Rb
Na <sub>2</sub> O	1.000	-0.530	-0.064	0.431	-0.411	-0.038	-0.005	-0.122	-0.518	-0.549
P <sub>2</sub> O <sub>5</sub>		1.000	0.129	-0.362	0.648	0.244	0.018	0.266	0.487	0.438
Cr			1.000	-0.053	0.049	0.055	0.233	-0.008	-0.014	-0.050
S				1.000	-0.298	-0.223	-0.079	-0.062	-0.282	-0.216
V					1.000	0.238	-0.017	0.526	0.661	0.411
Co						1.000	0.412	0.400	0.166	-0.015
Ni							1.000	0.145	-0.103	-0.023
Cu								1.000	0.543	-0.003
Zn									1.000	0.564
Rb										1.000

**Table 10.** (Contd.)

Component	Sr	Y	Zr	Nb	Ba	As	Pb	LOI	Sand	Silt	Clay
Sr	1.000	0.001	0.638	-0.160	-0.262	0.147	-0.160	-0.679	0.740	-0.664	-0.414
Y		1.000	0.431	0.089	0.329	0.060	0.200	-0.294	-0.226	0.234	0.075
Zr			1.000	0.077	-0.018	0.261	0.092	-0.763	0.501	-0.360	-0.427
Nb				1.000	0.234	-0.032	0.115	-0.012	-0.052	0.135	-0.116
Ba					1.000	0.043	0.216	-0.050	-0.193	0.303	-0.106
As						1.000	-0.025	-0.062	0.136	-0.050	-0.196
Pb							1.000	0.018	-0.104	0.101	0.045
LOI								1.000	-0.624	0.498	0.449
Sand									1.000	-0.873	-0.597
Silt										1.000	0.130
Clay											1.000

ble results from both methods were found to be most reliable. The calculation revealed correspondence between high Ba/Al and Ca/Al ratios only at 16, 1274, and 1574 cm. The most reliable data on anoxic conditions (V/Cr higher than 2.5) [29] were found at 74, 1294, 1334, and 1474 cm.

**Age of sediments and sedimentation rates.** The analysis of data in Table 1 showed that Holocene sediments are missing in the cored section. The identified litho-

stratigraphic units are attributed to the following marine isotopic stages (from top to bottom): sub-member 1a to MIS 2–MIS 3a, sub-member 1b to MIS 3–MIS 4, sub-member 1c to MIS 5a–d, member 2 to MIS 5e, and member 3 to MIS 6. It was found that sedimentation rates in sub-member 1a were highest during deglaciation and lowest during the Last Glacial Maximum and MIS 3a. In sub-member 1b, the interval 5–7 m with the highest sedimentation rates generally coincides with the

**Table 11.** Factor loadings for major oxides and elements

Component	Factor 1	Factor 2	Factor 3	Factor 4
SiO <sub>2</sub>	-0.06819	-0.31711	0.842454	0.275716
Al <sub>2</sub> O <sub>3</sub>	0.513087	0.345267	-0.57856	-0.01034
TiO <sub>2</sub>	0.455857	0.639282	0.438591	-0.09948
Fe <sub>2</sub> O <sub>3</sub>	0.271723	0.846487	-0.10571	-0.14548
MnO	-0.34492	0.816329	0.136247	0.257109
K <sub>2</sub> O	0.906295	0.045018	-0.06975	-0.10311
CaO	-0.36449	-0.21372	-0.08326	0.549555
MgO	-0.02457	0.785638	-0.31982	0.040134
Na <sub>2</sub> O	-0.72364	-0.06326	-0.06227	-0.13195
P <sub>2</sub> O <sub>5</sub>	0.620359	0.281102	-0.28415	0.402451
Cr	0.042226	0.038163	0.102307	0.16357
S	-0.47869	-0.13822	-0.26743	-0.36383
V	0.54354	0.485106	-0.30441	0.079586
Co	0.008865	0.432956	-0.099	0.482589
Ni	-0.01535	-0.01358	0.123096	0.426854
Cu	0.07149	0.742896	-0.13648	0.113389
Zn	0.643363	0.527713	-0.19109	-0.11951
Rb	0.866747	-0.05949	0.013845	-0.19397
Sr	-0.43373	-0.18552	0.578093	0.465109
Y	0.478559	0.404892	0.442376	-0.04819
Zr	0.133207	-0.03418	0.925301	0.013724
Nb	0.46065	-0.02785	0.046697	-0.00727
Ba	0.602194	-0.01684	-0.22391	0.310377
As	-0.05183	-0.11861	0.224516	-0.259
Pb	0.406653	0.050037	0.029925	-0.09681

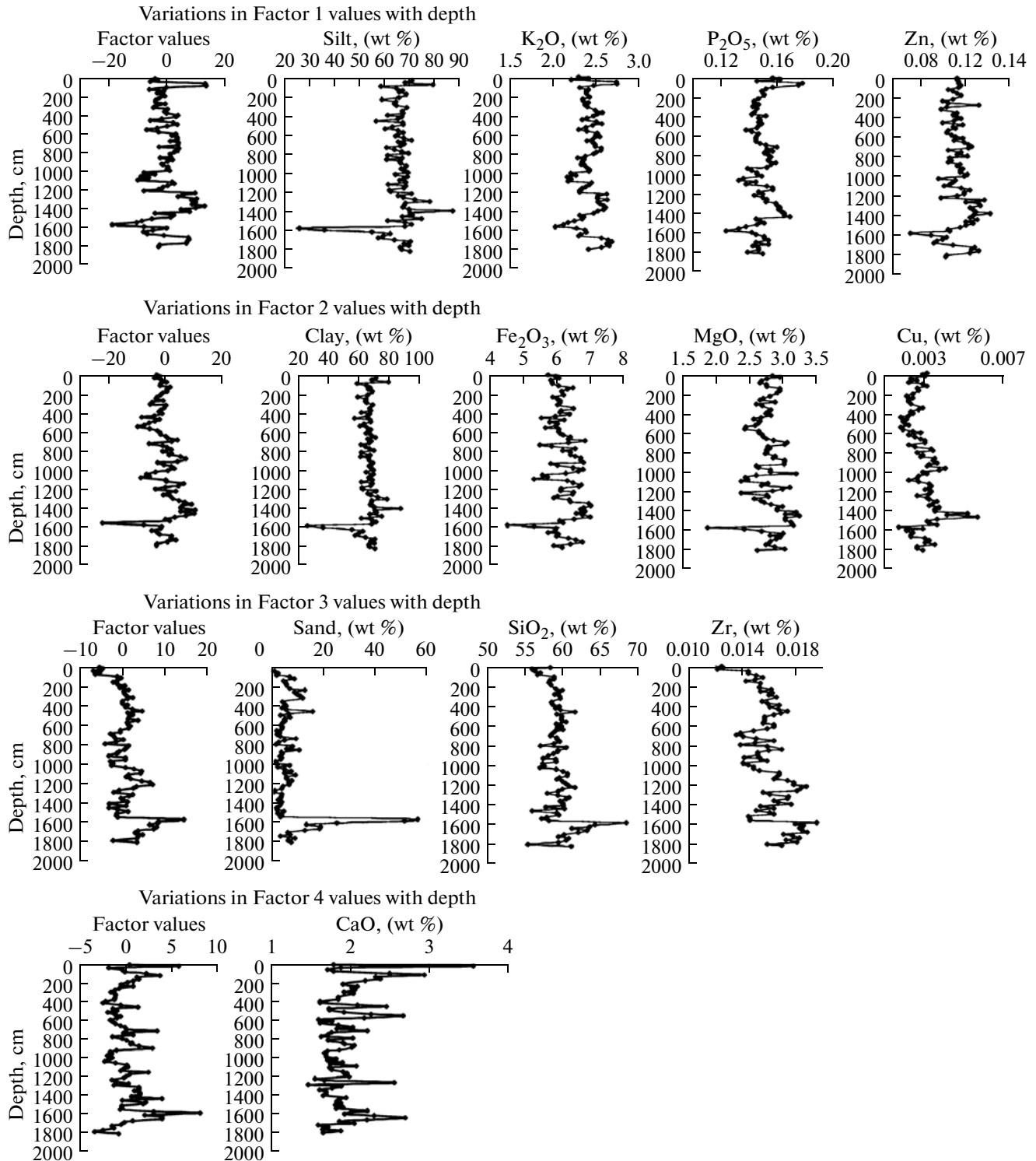
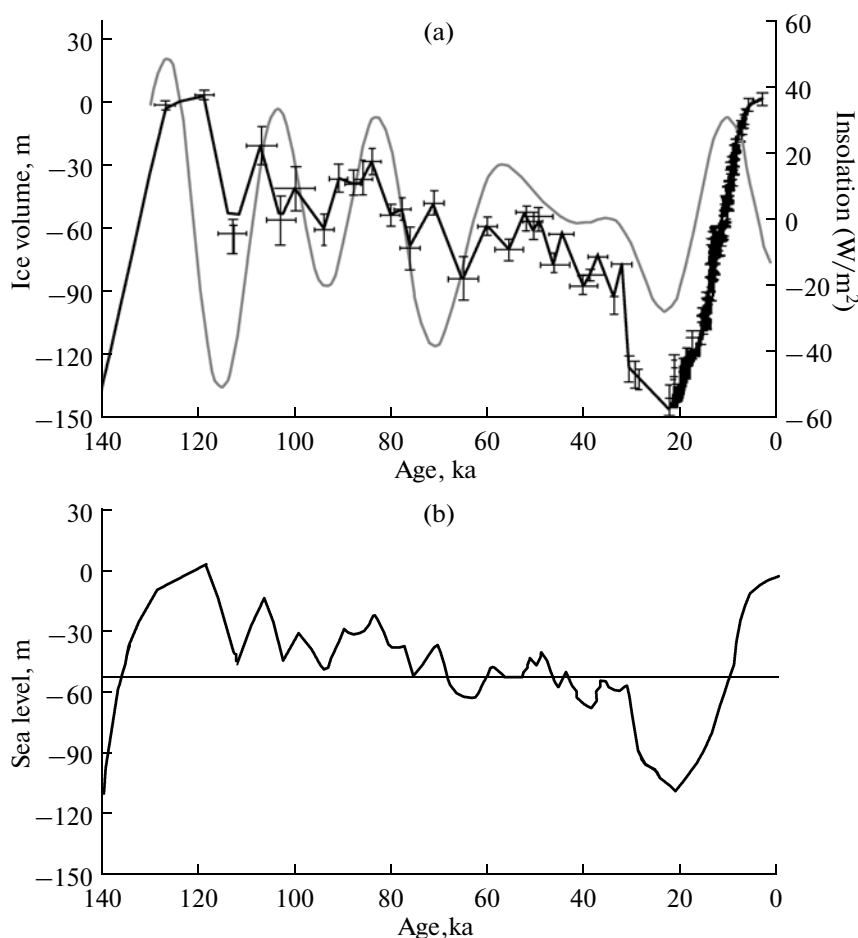


Fig. 7. Relationship between variations in the main factors controlling the distribution of major elements and components with depth.

interstadial of MIS 3. The average sedimentation rate for members 1 and 3 is about 15 cm/kyr. It is noteworthy that sedimentation rates were lowest (about 3 cm/kyr) during the last interglacial (MIS 5e), which was caused

by Yukon sediment transport to the Arctic Ocean as well as a sharp base level rise.

The above results lead to the sound conclusion that all of these samples (except for those from the sandy



**Fig. 8.** Changes in ice volume in the Northern Hemisphere proxied by variations in sea-level in meters (grey line), July insolation at 65°N (black line), and sea level [45].

interval) were derived from the Yukon–Tanana terrane and transported by the Yukon River across Beringia. The available literature data [30] suggest that the major sediment supply to areas near the Aleutians in the Pacific was from Alaska during Late Pleistocene cold periods and from Asia and Aleutians during interglacials.

**Geochronological constraints on sedimentological events.** Since our grain-size record begins (from top to bottom) from the sample at 34–35 cm, it does not reflect the first peak of magnetic susceptibility at 0–14 cm. Based on the data from Table 1, its age may roughly correspond to Heinrich event H0 (ages of these events were taken from [4, 31]). Note that Heinrich events are periods of extensive iceberg melting in the North Atlantic and deposition of iceberg-rafted debris [31]. The subsequent peaks of the >0.063-mm size fraction at 124 and 233 cm correspond to H1 and H2. Maxima at 320, 454, 754, 854, and 1500 cm cannot be correlated with the available geochronological data because of their poor accuracy (Table 1). Therefore, some of the sand maxima in

member 1 correlate with the deposition of iceberg-rafted debris in the North Atlantic (Heinrich events), whereas other maxima may reflect either the local deposition of ice-rafted debris in bottom sediments or the lack of precise geochronological control. The sand maxima of member 2 have no correlation with sea-ice and iceberg activity, and such maxima are absent in member 3.

A sharp increase in bottom current intensity is reliably identified only at 74 cm during the deglaciation of MIS 2. The second reliable increase can be related to the interstadial of MIS 3a, and the strong increase of current influence coincides with MIS 5a–d and MIS 6.

Relatively high paleoproductivity near the Shirshov Ridge was recorded during the last deglacial period of MIS 2 (especially for the Bölling-Alleröd warm event) and two episodes during the Last Interglacial, i.e. MIS 5e. Even higher primary production is inferred for the Holocene [8, 10]. In general, higher productivity during interglacials is characteristic of the Arctic and Subarctic regions [32]. Deposition in the oxygen mini-

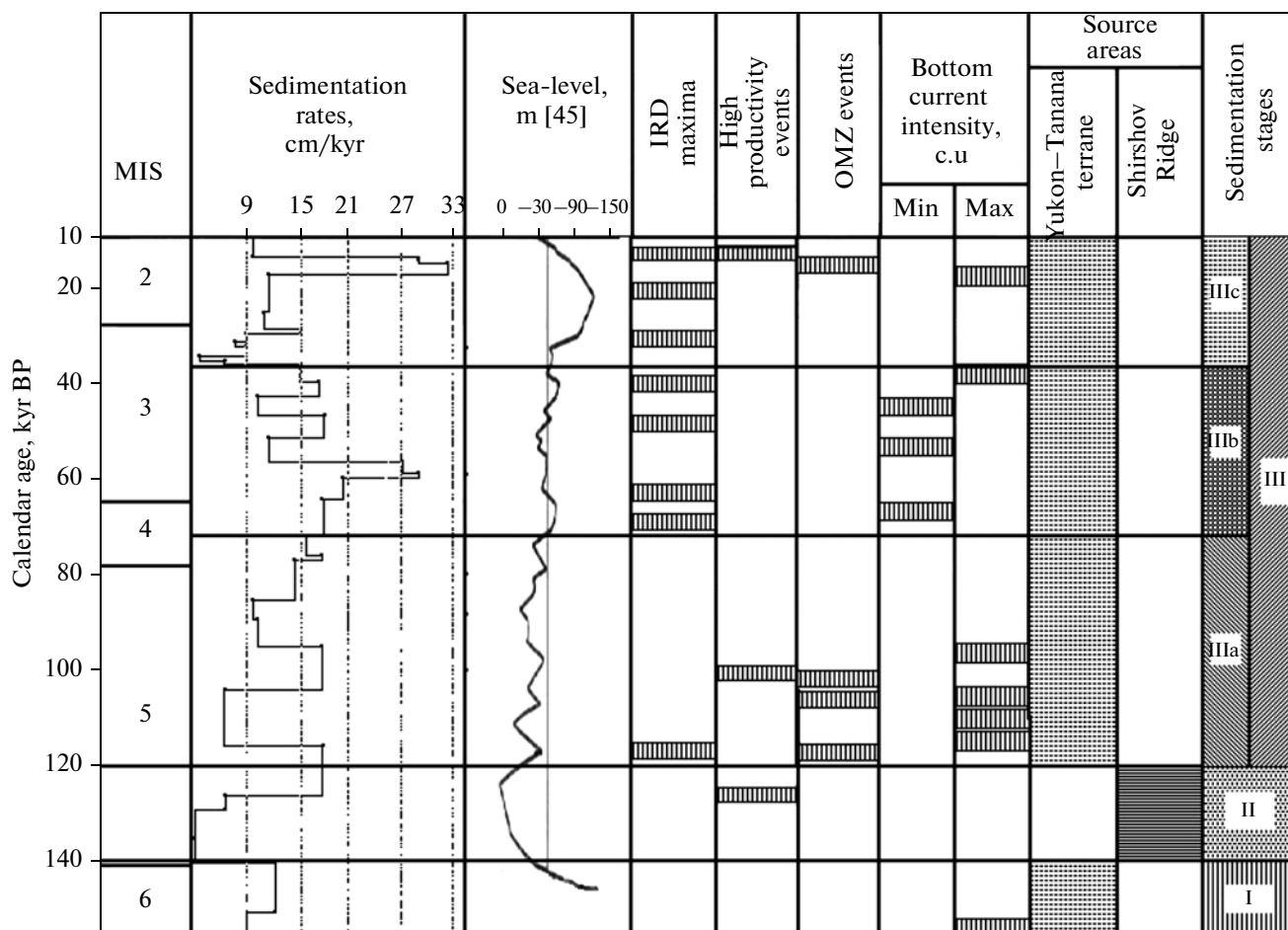


Fig. 9. Reconstruction of the Pleistocene sedimentary history of the Shirshov Ridge.

imum zone took place synchronously with the increase in paleoproductivity.

Figure 8 shows relationships between the June insolation at 65°N, Northern Hemisphere ice volume (Fig. 8a), and inferred sea-level variations in the Bering Sea (Fig. 8b) for the past 140 kyr. The analysis of these data suggests that the major climate changes were driven by insolation changes, and the history of Northern Hemisphere glaciation was a major control on glacioeustatic sea-level fluctuations.

Intervals with the highest smectite abundance (with chlorite/smectite ratio less than 1.0) may reflect the maximum supply of altered volcanic products, which roughly corresponds to MIS 5d–MIS 3. In this case, the oldest interval corresponds to an episode of MIS 5d, and the youngest interval corresponds to an episode of MIS 3 [33]. The highest illite abundance and illite/chlorite ratio in the three upper samples (Fig. 4) correspond to Termination 1a (deglacial of MIS 2) [11].

## CONCLUSIONS

Our results suggest that the sedimentation history of the Shirshov Ridge (Fig. 9) was dominated by low sea levels over the last 156 kyr. During this time, the lowest sea level, roughly corresponding to Substage IIIc, exposed much of the continental shelf, and the Bering Strait did not exist. This dry land is called Beringia [34]. During MIS 2 and 4, in some places of Kamchatka, Koryakia [35], Chukchi Peninsula, and Alaska [36], glaciers may have advanced to the present coastline of the Bering Sea. This has been a much disputed issue, and most authors suggest that there is no evidence that glaciers have ever extended far across Kamchatka, Koryakia, and Chukchi Peninsula and that a Beringia ice sheet has ever existed during glaciations [37–40]. Rivers currently draining into the Bering Sea, first of all Yukon, spread out over the exposed shelf area far beyond their modern deltas and estuaries. For example, 16–11 ka paleo-Yukon sediments were described on a shoal located southward from St. Lawrence Island on the Bering Sea shelf [41]. The Yukon River, which carried the suspended sediment load formed mainly by the

erosion of Paleozoic rocks within the Yukon–Tanana terrane, previously entered the Bering Sea near the Shirshov Ridge. The terrigenous material was supplied by the Yukon River to its mouth and further transported by a strong surface current (which also exists at present) over the continental slope and further in the WNW direction. This material was transported to the northern part of the Shirshov Ridge, where it partly accumulated in sediments.

During much of that period, the sea level was well below the present level but slightly above the bottom of the Bering Strait (Fig. 9); therefore, large shelf areas were under the sea. This corresponds to substages IIIb (mean sea-level) and IIIa and Stage I (sea-level highstand). As was previously shown, that time was dominated by the deposition of Yukon terrigenous materials on the Shirshov Ridge. Therefore, this material was transported along the shelf under water by the paleo-Yukon River. A similar situation exists at present, for instance, in the Ob–Yenisey shoal of the Kara Sea within the Ob and Yenisey paleo-valleys [3], on the Laptev Sea shelf within the paleo-valleys of the Khantanga, Lena, and Yana rivers [42], as well as in many other shelf areas worldwide. Moreover, the proposed scenario clearly suggests that no active connection existed between the Bering Sea and the Arctic Ocean. Otherwise, much of the Yukon sedimentary load, like at present (Fig. 1), would have been transported through the Bering Strait by northward- and northwestward-flowing longshore currents. This scenario was recorded in the sediments of Stage II, when much of the Yukon sediments were transported into the Arctic Ocean during the Last Interglacial, which was reflected in much lower sedimentation rates and changes in the chemical and mineralogical compositions of sediments.

On the ridge slope, the clay fraction was washed out from the sediments by permanent bottom currents. The major portion of the material was supplied to the bottom from catchment areas, which was primarily controlled by glacial history and sea-level fluctuations. In both cases, cyclic climate variations were the main reason.

During the maximum lowstand, the land area was a tundra-covered alluvial plain [34], and drifting ice, probably of the pack type, existed in the northern part of the sea, which is indicated by the data of [20] for MIS 4 and MIS 2. Icebergs were most likely absent here, although for Koryakia located close to the Shirshov Ridge, although the possibility of iceberg drift to the Bering Sea during the MIS 4–3 last glacial and MIS 2 was advocated by some authors [35]. The input of ice-rafted debris to the sediment was probably related to sea ice melting. Similar observations were previously reported in the Sea of Okhotsk [3, 43]. The coarse size (sand and gravel) indicates that the ice-rafted debris was

mostly transported by one-year ice, in particular fast ice. The synchrony of North Atlantic Heinrich events and ice-rafted debris maxima was observed in the Sea of Okhotsk [4], but they are only partly synchronous in the cored section from the Shirshov Ridge. In addition to the atmospheric teleconnections transmitting climate signals over very long distances in the Northern Hemisphere, of particular importance are specific regional features of ice regime, which are still poorly known.

Increases in primary productivity typical of interglacials and interstadials in the Arctic and Subarctic [3, 20] led to an extension of the oxygen minimum zone, which was reflected in the geochemistry of sediments. It is likely that an offshore seasonal sea ice either existed in the northern deep water part of the sea (as in Recent times) or did not exist at all. This issue requires further examination.

Sediments deposited during Substage IIIa are typically of dark grey color, clearly seen in the Arctic and Subarctic marine sediments, and have proximal sources located in areas with extensive continental ice sheets [3, 14, 15]. This observation is in agreement with our suggestion about the Yukon–Tanana terrane as the major source area, where the extensive MIS4–3 glacial in the Central Alaska took place [38]. Variations of the terrigenous fluxes were partially driven by bottom current intensity (Fig. 9). Episodes of slower currents typical of glacial periods are recorded during Substage IIIb. During Substage IIIc, fluctuations in the terrigenous fluxes appear to be largely controlled by mountain glaciation. The extent of glaciation in Alaska at the Last Glacial Maximum was 727 800 km<sup>2</sup> compared with the 74 700 km<sup>2</sup> of modern mountain glaciers [38].

Our results suggest that the Late Pleistocene sedimentary history of the Shirshov Ridge was essentially controlled by climate oscillations, which were crucial for the evolution of the Alaskan glaciation and glacioeustatic sea level change. The latter, in turn, affected relations of sea and land over the vast area of the northern Bering Sea shelf.

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